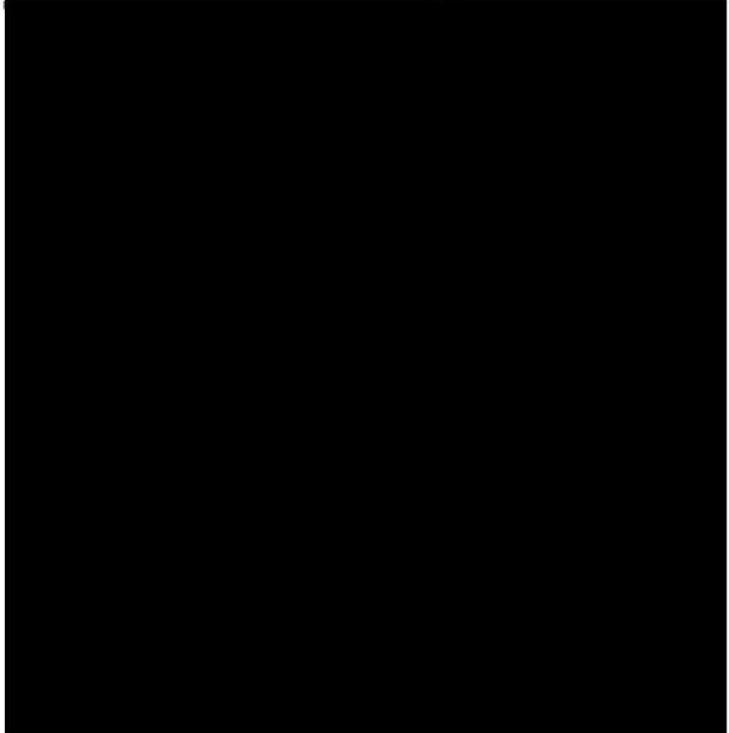


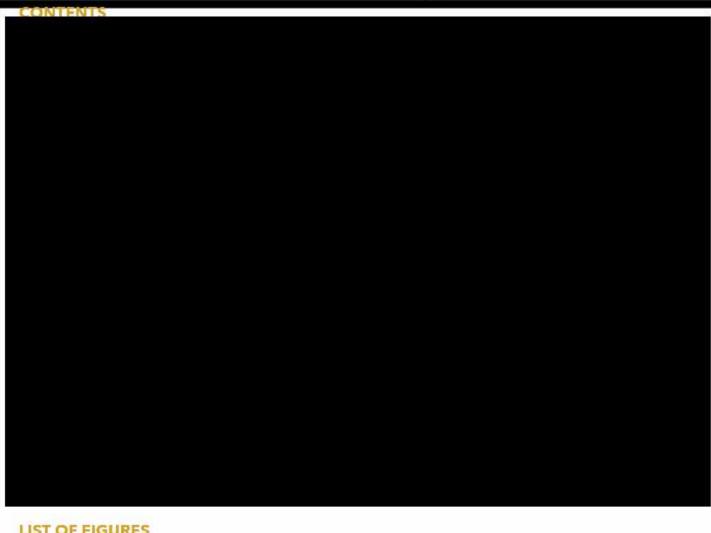
L7815/2001/12 LICENCE AMENDMENT SUPPORTING ATTACHMENTS

Version	1.0
Date	15 August 2025
Tenements	L 36/155, L 36/157, L 36/158, L 37/61, L 37/73, L 37/142, L 37/166, L 37/199, L 37/215, L 37/216, L 37/222, M 36/35, M 36/177, M 36/421, M 36/462, M 36/473, M 36/494, M 36/503, M 36/504, M 36/512, M 36/516, M 36/525, M 36/527, M 36/541, M 36/542, M 36/582, M 36/584, M 36/585, M 36/586, M 36/587, M 36/589, M 36/599, M 36/600, M 37/339, M 37/340, M 37/356, M 37/357, M 37/358, M 37/359, M 37/360, M 37/361, M 37/367, M 37/368, M 37/437, M 37/465, M 37/1148
Licence Holder	Northern Star (Thunderbox) Pty Ltd
Mining Centre	Thunderbox Operations









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TERMS USED

Acronym / Abbreviation	Definition	
CCIR	Critical containment infrastructure report.	
DMPE	Department of Mines, Petroleum and Exploration	
DWER	Department of Water and Environmental Regulation	
ESA	Environmentally sensitive area	
EP Act	Environmental Protection Act 1986	
IFC	Issued for construction	
L7815	Operating licence L7815/2001/12	
Northern Star	Northern Star (Thunderbox) Pty Ltd / Northern Star Resources Ltd	
TSF	Tailings Storage Facility	
W6601	Works approval W6601/2021/1	



1 Company Authorisation (Attachment 1C)





LETTER OF AUTHORITY IN REGARD TO APPROVAL APPLICATIONS

Please be advised that individuals holding the following positions (including their alternate or acting counterparts), are authorised to sign each of the following approval applications on behalf of Northern Star Resources Ltd and its subsidiaries:

Position:

- Site Senior Executive/General Manager Carosue Dam Operations
- Site Senior Executive/General Manager KCGM
- Site Senior Executive/General Manager Kalgoorlie Operations
- Site Senior Executive/General Manager Jundee Operations
- Site Senior Executive/General Manager Thunderbox Operations
- Site Senior Executive/General Manager Bronzewing Operations
- Site Senior Executive/General Manager Hemi / Pilbara Operations

Approval Applications:

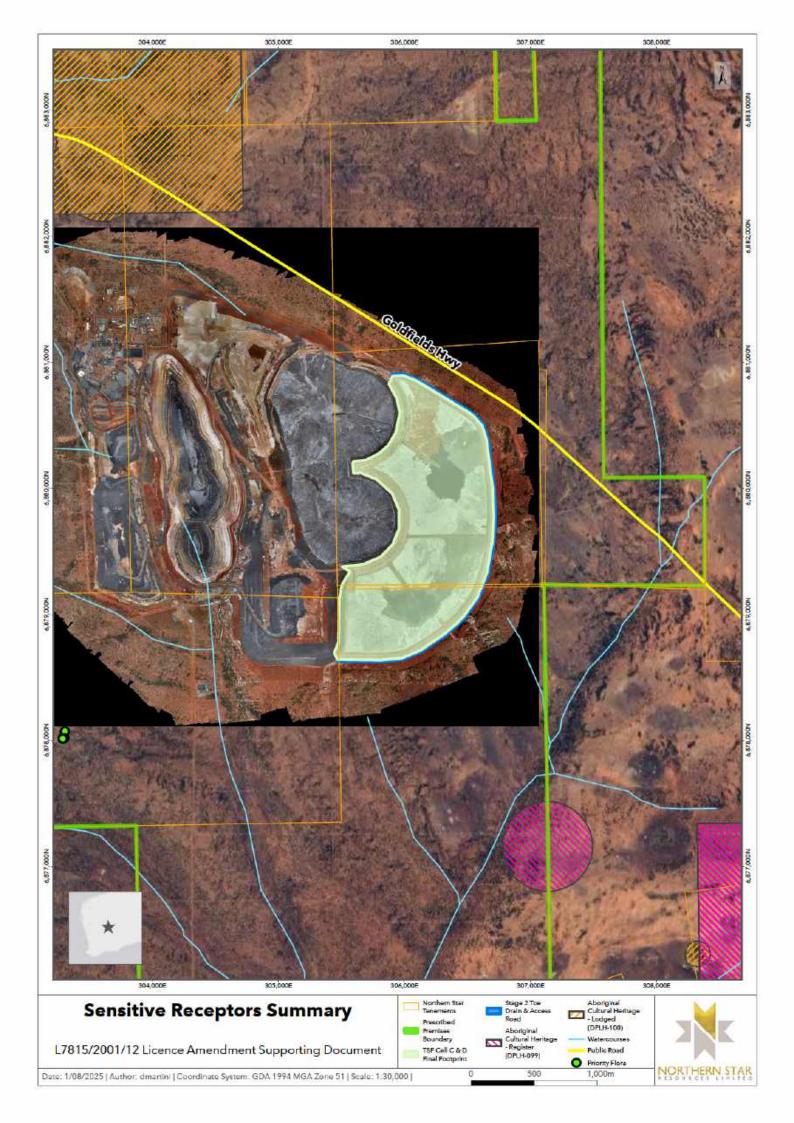
- Application for: Clearing Permit (Area Permit or Purpose Permit) / Amendment to a clearing permit / Surrender a clearing permit / Notification of change of land ownership
- Application form: Works Approval / licence / Renewal / Amendment / Registration / Surrender works approval or licence / Transfer works approval or licence, or Notify new occupier of registered premises
- Application for a water licence under section 26D of the Rights in Water and Irrigation Act 1914
- Application for a 5C licence to take groundwater
- Applications for a section 11/17/21A permit to interfere with bed and banks
- Annual Audit Compliance Report
- **Annual Environmental Report**
- Annual and Triennial Groundwater Reviews





2 Premises Maps (Attachment 2)







3 Proposed Activities (Attachment 3B)

3.1 Proposal Overview

Northern Star (Thunderbox) Pty Ltd (hereinafter referred to as Northern Star) is seeking an amendment to operating licence L7815/2001/12 (L7815) pursuant to Section 59B of the *Environmental Protection Act* 1986 (EP Act). The purpose of this amendment is to request for future lift stages of Thunderbox Tailings Storage Facility (TSF) Cells C & D to be authorised for construction and operation under L7815. A summary of the requested amendment and justification are outlined in Table 3-1 below.

Table 3-1: Proposed Licence Amendments Summary

Schedule 1 Category	Requested Amendment	Justification
Activity Amendments	e de la companya de l	
Category 5: Processing or beneficiation of metallic or non-metallic ore	Permit construction and operation of future lift stages (Stage 2 - 5) of Thunderbox TSF Cell C & D on Licence L7815 by including conditions pertaining to the construction of each lift stage.	In accordance with written advice from DWER "the best way forward for the Thunderbox TSF Cell C and D is to amend the licence to put in the individua lift stages. This will allow for future lifts to be captured in the licence framework."

3.2 Category 5: Processing or beneficiation of metallic or non-metallic ore

3.2.1 TSF Cell C & D

Thunderbox TSF is a paddock style TSF which manages disposal of all tailings generated from mining and processing across Thunderbox Operations (TBO). Thunderbox TSF is split into Cells A & B (original cells constructed in October 2002) and Cells C & D (new cells constructed in November 2023). As a combined TSF, the cells will facilitate tailings deposition at a rate of up to a permitted 7 mega tonnes per annum (Mtpa) throughput at TBO (estimated to 2034). It is noted that Cells A & B are managed independently to C & D to allow staggered raises whilst maintaining continuous operations.

Construction of Thunderbox TSF Cell C & D was authorised under Works Approval W6601/2021/1 (W6601) in May 2022. Construction of the first stage (starter embankment) was completed to 498.2 m RL in accordance with conditions specified in W6601. Following construction completion a critical containment infrastructure report (CCIR) was submitted to the Department of Water and Environmental Regulation (DWER).

Following acceptance of the CCIR by DWER, L7815 was amended on 29 January 2025, to permit ongoing deposition of tailings into Thunderbox TSF Cell C & D and use of associated infrastructure and monitoring equipment. Northern Star plan to construct the next stage (Stage 2) of TSF Cell C & D which comprises a downstream embankment raise in January 2026 (tentative).

Future lift stages (Stages 2-6) were detailed in the original design specification prepared by engineer of record Knight Piésold and were provided to support the W6601 application; however, the Works Approval does not include conditions pertaining to these stages. Whilst L7815does not have a limit on the embankment height, there are no conditions in the licence which would permit construction of a TSF raise. Written correspondence received from DWER on 11 June 2025 confirmed that Northern Star should seek a licence amendment to L7815 facilitating the construction of future stages.

3.2.1.1 TSF Life-of-Mine Assessment

Engineer of record Knight Piésold was recently engaged by Northern Star to prepare a Life-of-Mine (LOM) assessment for Thunderbox TSF which is provided in **Attachment 8B**. The LOM assessment reviewed the disposal requirements for management of tailings at TBO until 2033 based upon the following parameters:

amendments to tailings throughput;



- estimated achieved in-situ dry tailings densities;
- achieved tailings beach slope; and
- as-built records of existing cells.

In summary, in comparison to the original permitting design the LOM assessment has determined that insitu dry tailings densities are higher than assumed, tailings beach slope is flatter than assumed, and throughputs between 2021 to 2024 were lower than assumed (Knight Piésold, 2025). Consequently, the LOM assessment has revised the lift heights for future stages as outlined in Table 3-2 below. No changes have been made to Stage 2 (downstream raise) parameters and consequently the overall footprint of the TSF will not change from the original permitting design.

Table 3-2: Thunderbox TSF Cell C & D Stages

Stage	Lift Type	Original Permitting Design		Revised LOM Assessment Design		
		Lift Height (m)	Operating Height (m RL)	Lift Height (m)	Operating Height (m RL)	Change (m)
Stage 1 (constructed)	Starter embankment	10.0	494.1			
Stage 2	Downstream raise	4.1	498.2	4.1	498.2	0
Stage 3	Upstream raise*	2.0	500.2	2.4	500.6	+0.4
Stage 4	Upstream raise*	2.4	502.6	2.4	503.0	+0.4
Stage 5	Upstream raise*	2.4	505.0	1.0	504.0	-1.0
Stage 6	Upstream raise*	0.8	505.8 Not required based upon LOM assessment			

^{*} Upstream raised against external perimeters. Embankments abutting the Eastern Waste Dump and TSF Cell A & B are downstream raised.

It is noted that these levels are best estimates based upon current information available, and that changes to tailings throughput, in-situ dry tailings densities and beach slope will impact future levels (Knight Piésold, 2025). Detailed issued for construction (IFC) design reports will be prepared by Knight Piésold prior to construction of each stage based upon updated parameters. Should this result in any material changes to lift heights, a subsequent licence amendment application will be sought by Northern Star.

3.2.1.2 Raise Construction

Construction of Stages 2 - 5 of Thunderbox TSF Cell C & D will be undertaken generally in accordance with the original permitting design drawings included in **Attachment 8C** and detailed below:

- General Arrangement Final (801-296-D3000-115)
- Divider embankment (801-296-D3000-301)
- Embankment raises (801-296-D3000-302)
- Toe drain (801-296-D3000-302)
- Buttress (01-296-D3000-303)
- Underdrainage tower (801-296-D3000-452)
- Decant tower (801-296-D3000-551)
- Piezometer (801-296-D3000-905)

It is noted that the raise heights will vary from the original design as outlined previously in this application, and that construction will be conducted in accordance with IFC designs issued prior to each stage. Where



minor deviations occur that do not present material defects from the approved design (i.e. minor variations to RLs), these will be reflected in submitted audit reports.

3.2.1.3 Suggested Licence Text

Suggested licence amendment text has been presented in Table 3-3 below.

Table 3-3: Suggested Licence Text

Infrastructure	Design and construction requirements	Infrastructure location
Thunderbox TSF Cell C & D Stage 2	 (a) Embankment lift to 498.2 m RL (b) Divider embankment lift to 498.2 m RL (c) Decant tower raise to 498.2 m RL (d) Underdrainage tower raise to 498.2 m RL (e) Installation of ten new piezometers to replace decommissioned piezometers from Stage 1. (f) Installation of new perimeter toe drain and berm to replace decommissioned toe drain and berm from Stage 1. 	TSF Cell C & D as shown in Figure 4 of Schedule 1.
Thunderbox TSF Cell C & D Stage 3	 (a) Embankment lift to 500.6 m RL (b) Divider embankment lift to 500.6 m RL (c) Decant tower raise to 500.6 m RL (d) Underdrainage tower raise to 500.6 m RL (e) Installation of ten new piezometers to replace decommissioned piezometers from Stage 2. 	TSF Cell C & D as shown in Figure 4 of Schedule 1.
Thunderbox TSF Cell C & D Stage 4	 (a) Embankment lift to 503.0 m RL (b) Divider embankment lift to 503.0 m RL (c) Decant tower raise to 503.0 m RL (d) Underdrainage tower raise to 503.0 m RL (e) Installation of ten new piezometers at Stage 4 embankment crest to replace decommissioned piezometers from Stage 3. 	TSF Cell C & D as shown in Figure 4 of Schedule 1.
Thunderbox TSF Cell C & D Stage 5	 (a) Embankment lift to 504.0 m RL (b) Divider embankment lift to 504.0 m RL (c) Decant tower raise 504.0 m RL (d) Underdrainage tower raise to 504.0 m RL (e) Installation of ten new piezometers at Stage 5 embankment crest to replace decommissioned piezometers from Stage 4. 	TSF Cell C & D as shown in Figure 4 of Schedule 1.



4 Emissions & Discharges (Attachment 6A)

Potential emissions / discharges, associated risk pathways and proposed operational controls are summarised in Table 4-1 below, noting many of these controls are existing controls adopted from W6601 and L7815 with no changes.

Table 4-1: Emissions & Discharges

Activity	Potential Emissions / Discharge	Potential Risk Pathways	Controls
Construction	Burgardovidovido and		
Clearing, earthworks, mobile plant movements	Fugitive dust	Airborne dispersion.	Proposed Controls Provision of watercarts during construction activities with regular spraying as required to prevent dust liftoff. Monitoring of wind conditions and ceasing activities during excessive visual dusting towards Goldfields Highway.
	Noise	Airborne dispersion.	No nearby noise sensitive receptors - no specific controls required.
Operation			A
Discharge of Decant tailings into TSF water and or tailings		Decant pipeline / tailings delivery pipeline rupture or leaks.	Pipelines constructed from HDPE PE100 and installed in accordance with AS4130 and AS413. Stored in V-drains sufficient to contain spillages between routine inspections. Fitted with telemetry.
		Overtopping of TSF into surrounding environment.	Existing Controls / Proposed Controls A minimum top of embankment freeboard of 500 mm maintained at all times.*
	Seepage	Seepage through base or walls of TSF into surrounding natural soils / groundwater table.	 Proposed Controls Seepage collection and recovery system implemented (decant and underdrainage tower raises). Seepage is returned to the TSF, or pumped back into the mine water circuit. Perimeter toe drain to be installed adjacent to downstream embankment (Stage 2) raise to replace existing toe drain. Seepage recovery bores installed when trigger levels reached (standing water levels rise higher than 6 mbgl).
			Low permeability compacted soil liner and underdrainage collection system installed. Diversion drain in place to ensure natural surface water flows are diverted around TSF maintain downstream flows. Berm positioned between toe drain and diversion drain to prevent mixing of clean and potentially contaminated water. Cut-off trench installed. Groundwater monitoring in accordance with licence.
	Dry tailing dusting	Airborne dispersion.	Proposed Controls Capping of tailings and rehabilitation in accordance with approved Mine Closure Plan.

^{*}existing licence condition states or containment of a 1-in-100 year 72-hour rainfall event (whichever is greater) must be maintained. This rainfall event is approximately 200 mm which is significantly exceeded by 500 mm and is therefore redundant.



5 Other Approvals and Consultation (Attachment 5)

Northern Star is aware of its obligation to seek approvals and comply with requirements under various Federal and State Legislation in addition to securing this Works Approval. At the time of preparing this document environmental legislation applicable to this application has been summarised in Table 5-1 below.

Table 5-1: Environmental Approvals Summary

Legislation / Regulation	Environmental Aspect(s)	Approval Status
Aboriginal Heritage Act 1972	Aboriginal heritage sites	A heritage survey was completed with representatives of Darlot Group over TSF Cell C & D in 2021. No disturbance to Aboriginal Sites will occur as part of the proposed activities.
Biodiversity Conservation Act 2016 (BC Act)	Threatened flora, fauna and ecological communities	No Threatened flora, fauna or ecological communities within the prescribed premises boundary during surveys – no Section 40 approval is required under BC Act.
Environment Protection and Biodiversity Conservation Act 1999 (EPBC Act)	Matters of national environmental significance	No matters of national environmental significance applicable - no referral required under EPBC Act.
Environmental Protection Act 1986 (Part IV)	Matters of state environmental significance	Not a 'significant proposal' - no triggers for referral under Section 38 of the EP Act.
Environmental Protection (Clearing of Native Vegetation) Regulations 2004	Native vegetation clearing	Clearing permit approved over Thunderbox 10369/1, clearing for downstream embankment raise will be conducted under permit conditions.
Environmental Protection (Noise) Regulations 1997 (Noise Regulations)	Social surroundings	Construction noise management plan not required - no nearby sensitive receptors nearby.
Mining Act 1978	Biodiversity Water resources Land and soils Closure and rehabilitation	Thunderbox TSF Cell C&D is approved for a maximum height of 505.8 m RL in Mining Proposal Reg ID 99935, which is greater than the proposed height of Stage 5 (504.0 m RL)
Native Title Act 1993	Land access	All tenements within the proposed prescribed premises are granted. Northern Star are in progress of negotiating an agreement with Darlot Group Native Title holders across Thunderbox.
Rights in Water and Irrigation Act 1914 (RIWI Act)	Groundwater resources	Thunderbox is covered by several groundwater licences (GWL) with a total abstraction allocation of 9,050,000 kL annually as per below: GWL 635550 - 6,000,000 kL GWL 154472 - 1,050,000 kL GWL 1586766 - 2,000,000 kL



5.1 Stakeholder Engagement

Stakeholder engagement has been ongoing throughout the life of TBO commencing under previous ownership. Northern Star has a dedicated Community & Heritage team to enhance stakeholder outcomes across the business. Northen Star's stakeholder engagement strategy is informed by the principles outlined below.

- Communication Communication must be open, accessible, clearly defined, two-way and appropriate.
- Transparency The process and outcomes of community and stakeholder engagement should, wherever possible, be made open and transparent, agreed upon and documented.
- Collaboration A co-operative and collaborative approach to seek mutually beneficial outcomes is considered key to effective engagement.
- Inclusiveness Inclusiveness involves identifying and involving communities and stakeholders early and throughout the process, in an appropriate manner.
- Integrity Community and stakeholder engagement should establish and foster mutual trust and respect

Key stakeholders to TBO and adopted engagement strategies for each party are summarised in Table 5-2 below, noting that this is subject to change as the project develops over time.

Group	Key Stakeholder	Engagement	Timing
Native Title (Darlot)	Watarra Aboriginal Corporation RNTBC	Organising heritage surveys and clearance prior to any new land disturbance.	Heritage survey of TSF Cell C & D (including Stage 2 clearing footprint) completed in May 2021.
Pastoralist	Weebo Station	Overview of Thunderbox including TSF Cell C & D project.	Completed June 2021.
State Government	DMPE (formerly DEMIRS)	Overview of TBO TSF Cell C & D project.	Completed August 2021.
		Provision of updated Mining Proposal for any change to activities.	Not required - proposed TSF raise height lower than currently approved and no changes to footprint.

Table 5-2: Stakeholder Engagement Summary

5.2 Stakeholder Engagement Register

Northern Star utilise INX InForm as a Stakeholder Engagement Register (SER). The purpose of this SER is to document consultation activities undertaken with all stakeholder groups and individuals, including any comments or concerns raised and follow up actions. The SER is a live database and will continue to document all consultation through the LOM.



6 Siting and Location (Attachment 7)

6.1 TSF Cell C & D

Detailed siting and location information was provided in support of TSF Cell C & D as referenced in the W6601 decision report (DWER 2022). There are no proposed changes to the design of Stage 2 (downstream raise) which has already been risk assessed and determined to have appropriate controls commensurate to the risk. Minor changes to Stages 3 – 5 will not result in any materials changes to the risk profile, as the overall footprint is unchanged and the maximum height has decreased from the original permitting design.

6.2 Environmental Sensitive Receptors Summary

Distances and directions to nearest environmentally sensitive receptors from the prescribed premises boundary have been reviewed as of July 2025 and summarised in Table 6-1 below. Attachment 2B – Sensitive Receptors Summary shows the location of these receptors in relation to TSF Cell C & D.

Table 6-1: Environmentally Sensitive Receptors

Туре	Description	Distance and Direction	Proposed Controls	Data Source(s)
Environmentally Sensitive Areas (ESA)	No ESAs are located within TBO. The nearest ESA is associated with Wanjarri Nature Reserve.	70 km north of TBO.	N/A - significant separation.	DWER-046
Threatened Ecological Communities	No known TECs are located within TBO.	75 km west of TBO.	N/A - significant separation.	DWER-038
Threatened and / or Priority fauna	No known Threatened or Priority fauna are located within TBO.	12 km north (database record 91,829)	N/A - no Threatened or Priority fauna likely.	DBCA-037 Botanica (2025)
Threatened and / or Priority flora	No known Threatened flora are located within TBO. Lysiandra baeckeoides (P3) are located within the prescribed premises boundary but not within TSF footprint.	2,200 m southwest of TSF (within prescribed premises boundary).	N/A - significant separation.	DBCA-036 Botanica (2025)
Aboriginal and other heritage sites	Registered site 2552 (Leonora - Leinster 23).	1,500 m southeast from TSF (partially within prescribed premises boundary).	N/A - significant separation.	ACHIS 2025
Public drinking water source areas	Leonora Water Reserve is the nearest proclaimed public drinking water source area.	60 km southeast of TBO.	N/A - significant separation.	DWER-033
Rivers, lakes, oceans, and other bodies of surface water, etc.	Ephemeral watercourses traverse the proposed prescribed premises boundary.	300 m east of TSF (within prescribed premises boundary).	As detailed in Table 4-1.	GEODATA TOPO 250K (Geoscience Australia)
Acid sulphate soils	TBO is mapped as having "extremely low probability of occurrence" in the Atlas of Australian Acid Sulphate Soils*	14 km north to nearest natural high probability area.	N/A - significant separation.	CSIRO 2011
Other	Goldfields Highway	200 m from TSF (L7815 premises intersects Goldfields Highway).	As detailed in Table 4-1.	LGATE-195

^{*} Atlas of Australian Acid Sulphate Soils mapping has incorrectly identified TSF Cell A & B as a lake.



7 Works Approval W6601/2021/11 (Attachment 8A)

Works Approval

Works approval number W6601/2021/1

Works approval holder Northern Star (Thunderbox) Pty Ltd

ACN 107 154 727

Registered business address Level 1, 388 Hay Street, PERTH, 6008

DWER file number DER2021/000505

Duration 3 May 2022 to 3 May 2027

Date of amendment 13/11/2024

Premises details North Eastern Goldfields Operations

Mining tenements: M36/512, M36/582, M36/585

Prescribed premises category description (Schedule 1, Environmental Protection Regulations 1987)	Assessed production / design capacity
Category 5: Processing or beneficiation of metallic or non-metallic ore	7,000,000 tonnes per annum

This works approval is granted to the works approval holder, subject to the attached conditions, on 13 November 2024, by:



Works approval history

Date	Reference number	Summary of changes
3/05/2022	W6601/1/2021	Construction of Cells C and D of Thunderbox TSF
13/11/2024	W6601/1/2021	Extension of time limited operations

Interpretation

In this works approval:

- the words 'including', 'includes' and 'include' in conditions mean "including but not limited to", and similar, as appropriate;
- (b) where any word or phrase is given a defined meaning, any other part of speech or other grammatical form of that word or phrase has a corresponding meaning;
- (c) where tables are used in a condition, each row in a table constitutes a separate condition:
- (d) any reference to an Australian or other standard, guideline, or code of practice in this works approval:
 - (i) if dated, refers to that particular version; and
 - if not dated, refers to the latest version and therefore may be subject to change over time;
- unless specified otherwise, any reference to a section of an Act refers to that section of the EP Act; and
- (f) unless specified otherwise, all definitions are in accordance with the EP Act.

NOTE: This works approval requires specific conditions to be met but does not provide any implied authorisation for other emissions, discharges, or activities not specified in this works approval.

Works approval conditions

The works approval holder must ensure that the following conditions are complied with:

Construction phase

Infrastructure and equipment

- The works approval holder must:
 - (a) construct the critical containment infrastructure:
 - in accordance with the corresponding design and construction requirements;
 - (c) at the corresponding infrastructure location;
 as set out in Table 1.

Table 1: Critical containment infrastructure design and construction requirements

	Infrastructure	Design and construction requirements	Infrastructure location
1.	Tailing Storage Facilities - Cells C and D	Seepage control and underdrainage system consisting of: A low permeability clay/soil liner of at least 6 x 10-7 m/s permeability. Cut-off trench; Embankment upstream toe drain; Basin underdrainage collection system; Underdrainage collection tower; Low permeability basin liner; and External toe drain A diversion drain sized 6m wide by 1 m deep, with a side slope of 2H:1V A berm positioned between the external toe drain and the diversion drain 1m high by 6m wide.	Underdrainage layout – as depicted in Schedule 1, Figure 2 Cut off trench and upstream toe drain details - as depicted in Schedule 1, Figure 3 and Figure 4. External toe drain, berm and diversion drain – as depicted in Schedule 1, Figure 4 and Figure 5
2.	Pipelines carrying tailings and decant return water	 Constructed from HDPE PE100 and installed in accordance with AS4130 and AS413, and the Plastics Industry Pipe Association of Australia Limited (PIPA) Guideline POP003. Stored in V-drains sufficient to contain spillages between routine inspections. Fitted with telemetry (the Citect processing plant control system) which monitors pressure. 	

The works approval holder must design, construct, and install groundwater monitoring wells in accordance with the requirements specified in Table 2.

Table 2: Infrastructure requirements - groundwater monitoring wells

Infrastructure	Design, construction, and installation requirements	Monitoring well location(s)	Timeframe	
Groundwater monitoring wells TSF-CD-01, TSF-CD-02 TSF-CD-03 TSF-CD-04	Well design and construction: Designed and constructed in accordance with ASTM D5092/D5092M-16: Standard practice for design and installation of groundwater monitoring bores. Well screens must target the part, or parts, of the aquifer most likely to be affected by contamination¹. Where temporary/seasonal perched features are present, wells must be nested, and the perched features individually screened.	As depicted in Schedule 1, Figure 1.	Must be constructed, developed (purged), and determined to be operational by no later than 14 calendar days prior	
	Logging of borehole: Soil samples must be collected and logged during the installation of the monitoring wells. A record of the geology encountered during drilling must be described and classified in accordance with the Australian Standard Geotechnical Site Investigations AS1726. Any observations of staining / odours or other indications of contamination must be included in the bore log.		to discharge of tailings to TSF Cell C and Cell D.	
	Well construction log: Well construction details must be documented within a well construction log to demonstrate compliance with ASTM D5092/D5092M-16. The construction logs shall include elevations of the top of casing position to be used as the reference point for water-level measurements, and the elevations of the ground surface protective installations.			
	Well development: All installed monitoring wells must be developed after drilling to remove fine sand, silt, clay and any drilling mud residues from around the well screen to ensure the hydraulic functioning of the well. A detailed record should be kept of well development activities and included in the well construction log.			
	Installation survey: the vertical (top of casing) and horizontal position of each monitoring well must be surveyed and subsequently mapped by a suitably qualified surveyor.			
	Well network map: a well location map (using aerial image overlay) must be prepared and include the location of all monitoring wells in the monitoring network and their respective identification numbers.			

Note 1: refer to Section 8 of Schedule B2 of the Assessment of Site Contamination NEPM for guidance on well screen depth and length.

- The works approval holder must, within 60 calendar days of the monitoring wells being constructed, submit to the CEO a well construction report evidencing compliance with the requirements of condition 2.
- 4. The Licence Holder shall ensure that prior to, and during any disturbance to the 'affected area', as denoted by Figure 6 in Schedule 1, the area is continually wetted using water sprays, dribble bars or other suitable methods to ensure there is no visible windblown dust.
- The licence holder must ensure that no visible dust generated from the construction activities crosses the boundary of the premises.

Compliance reporting

- 6. The works approval holder must within 30 calendar days of the Critical Containment Infrastructure identified by condition 1 being constructed:
 - (a) undertake an audit of their compliance with the requirements of condition 1;
 and
 - (b) prepare and submit to the CEO a Critical Containment Infrastructure Report on that compliance.
- 7. The Critical Containment Infrastructure Report required by condition 6 must include as a minimum the following:
 - (a) certification by a suitably qualified geotechnical engineer that each item of critical containment infrastructure or component thereof, as specified in condition 1, has been built and installed in accordance with the requirements specified in condition 1;
 - (b) as constructed plans and a detailed site plan showing the location and dimensions for each item of critical containment infrastructure or component thereof, as specified in condition 1;
 - (c) photographic evidence of the installation of the infrastructure;
 - (d) be signed by a person authorised to represent the works approval holder and contains the printed name and position of that person;
 - (e) monitoring data indicating the baseline ambient environmental conditions at the premises prior to and immediately following construction of the items of infrastructure.
- The monitoring of the baseline ambient environmental conditions required under condition 7(e) must
 - (a) be undertaken in accordance with Table 3;
 - (b) all sample analysis must be undertaken by laboratories with current accreditation from the National Association of Testing Authorities (NATA) for the relevant parameters.

Table 3: Determination of baseline ambient environmental conditions

Parameter	Monitoring		-	Averaging	Method	
	location	Unit Freque	Frequency	period	Sampling	
pH ¹	Groundwater monitoring	9	No later	Spot	In accordance with AS/NZS	
SWL	wells		sample	5667.11		

120 07	Monitoring			Averaging	Method
Parameter	location	Unit	Frequency	period	Sampling
TDS	TSF-CD-01,	mg/L	days prior		
Weak acid dissociable cyanide (WAD CN)	TSF-CD-02 TSF-CD-03 TSF-CD-04	경화되었다. 경기자인 기업병이	discharge of tailings to TSF Cell C and Cell D.		
Total cyanide (CN)					
Arsenic (As)					
Antimony (Sb)					
Bicarbonate (HCO ₃)					
Calcium (Ca)					
Carbonate (CO ₃)					
Cadmium (Cd)					
Chloride (CI)					
Chromium (Cr)					
Cobalt (Co)					
Copper (Cu)					
Iron (Fe)					
Lead (Pb)					
Magnesium (Mg)					
Manganese (Mn)					
Mercury (Hg)					
Molybdenum (Mo)					
Nickel (Ni)					
Nitrate (NO ₃)					
Potassium (K)					
Selenium (Se)					
Sodium (Na)					
Sulphate (SO ₄)					
Thallium (TI)					
Zinc					

Note 1: In-field non-NATA accredited analysis permitted.

Time limited operations phase

Commencement and duration

- 9. The works approval holder may only commence time limited operations for an item of critical containment infrastructure identified in condition 1:
 - (a) where the CEO has notified the works approval holder that the Critical Containment Infrastructure Report for that item of infrastructure as required by condition 6 meets the requirements of that condition; or
 - (b) where at least 10 business days have passed after the Critical Containment Infrastructure Report for that item of infrastructure as required by condition 6 has been submitted to the CEO.

Time limited operations requirements and emission limits

- 10. The works approval holder may conduct time limited operations for an item of infrastructure specified in condition 11 (as applicable):
 - (a) for a period not exceeding 270 calendar days from the day the works approval holder meets the requirements of condition 9 for that item of infrastructure; or
 - (b) until such time as a licence for that item of infrastructure is granted in accordance with Part V of the Environmental Protection Act 1986, if one is granted before the end of the period specified in condition 10(a).
- During time limited operations, the works approval holder must ensure that the premises infrastructure and equipment listed in Table 4 and located at the corresponding infrastructure location is maintained and operated in accordance with the corresponding operational requirement set out in Table 4.

Table 4: Infrastructure and equipment requirements during time limited operations

	Site infrastructure and equipment	Operational requirement	Infrastructure location
1.	TSF Cell C and Cell D	Low points are filled, and a beach is established in each cell to allow for efficient recovery of decant water within 16 weeks of commencing discharge to the TSF.	As depicted in Figure 1, Schedule 1
		 A minimum freeboard of 500mm from top of embankment. 	
		 Inspection of TSF at least twice per 12-hour shift. 	
2.	Pipelines carrying tailings and decant return water	Inspection of pipelines at least twice per 12-hour shift.	

- 12. The Licence Holder shall ensure that prior to, and during any disturbance to the following TSF components, these areas are continually wetted using water sprays, dribble bars or other suitable methods to ensure there is no visible windblown dust:
 - (a) The surface of the TSF
 - (b) The onsite roadways in the immediate vicinity of the TSF
 - (c) TSF embankments
 - (d) The 'TSF affected area', as denoted by Figure 6 in Schedule 1.

Monitoring during time limited operations

- 13. The works approval holder must:
 - (a) monitor the groundwater during time limited operations for concentrations of the identified parameters in accordance with Table 5.
 - (b) all sample analysis must be undertaken by laboratories with current accreditation from the National Association of Testing Authorities (NATA) for the relevant parameters.

Table 5: Monitoring of ambient concentrations during time limited operations

	Monitoring			Averaging	Method
Parameter	location	Unit Frequency		period	Sampling
pH ¹		,5			
SWL		mbgl			
TDS				in	In accordance
Weak acid dissociable cyanide (WAD CN)		mg/L	Quarterly	Spot sample	with AS/NZS 5667.11
Total cyanide (CN)		1 5			
Arsenic (As)		10			
Antimony (Sb)					
Bicarbonate (HCO ₃)					In accordance with AS/NZS 5667.11
Calcium (Ca)			Six monthly or at least once prior to the end of the limited time operations period.	Spot sample	
Carbonate (CO ₃)					
Cadmium (Cd)					
Chloride (CI)					
Chromium (Cr)					
Cobalt (Co)					
Copper (Cu)	Groundwat er				
Iron (Fe)	monitoring				
Lead (Pb)	wells TSF-CD-	touranara			
Magnesium (Mg)	01,	mg/L			
Manganese (Mn)	TSF-CD-02				
Mercury (Hg)	TSF-CD-03				
Molybdenum (Mo)	137-00-04				
Nickel (Ni)					
Nitrate (NO ₃)					
Potassium (K)					
Selenium (Se)					
Sodium (Na)					
Sulphate (SO ₄)					
Thallium (TI)					

Parameter	Monitoring location	Unit	Frequency	Averaging period	Method Sampling
Zinc					
Weak acid dissociable cyanide (WAD CN)	Decant	mg/L	Quarterly	Spot sample	In accordance
Arsenic (As)					with AS/NZS 5667.11

Note 1: In-field non-NATA accredited analysis permitted.

14. The works approval holder must record the results of all monitoring activity required by condition 13.

Compliance reporting

- 15. The works approval holder must submit to the CEO a report on the time limited operations within 60 calendar days of the completion date of time limited operations or 60 calendar days before the expiration date of the works approval, whichever is the sooner.
- 16. The works approval holder must ensure the report required by condition 15 includes the following:
 - (a) a summary of the time limited operations, including timeframes and amount of tailings discharged;
 - (b) a summary of groundwater monitoring results obtained during time limited operations under condition 13.
 - (c) a review of performance and compliance against the conditions of the works approval; and
 - (d) where the manufacturer's design specifications and the conditions of this works approval have not been met, what measures will the works approval holder take to meet them, and what timeframes will be required to implement those measures.

Records and reporting (general)

- 17. The works approval holder must record the following information in relation to complaints received by the works approval holder (whether received directly from a complainant or forwarded to them by the Department or another party) about any alleged emissions from the premises:
 - (a) the name and contact details of the complainant, (if provided);
 - (b) the time and date of the complaint;
 - the complete details of the complaint and any other concerns or other issues raised; and
 - (d) the complete details and dates of any action taken by the works approval holder to investigate or respond to any complaint.
- 18. The works approval holder must maintain accurate and auditable books including the following records, information, reports, and data required by this works approval:
 - (a) the works conducted in accordance with condition 1;

- (b) any maintenance of infrastructure that is performed in the course of complying with condition 11;
- (c) monitoring programmes undertaken in accordance with conditions 8 and 13; and
- (d) complaints received under condition 17.
- **19.** The books specified under condition 18 must:
 - (a) be legible;
 - (b) if amended, be amended in such a way that the original version(s) and any subsequent amendments remain legible and are capable of retrieval;
 - (c) be retained by the works approval holder for the duration of the works approval; and
 - (d) be available to be produced to an inspector or the CEO as required.

Definitions

In this works approval, the terms in Table 1 have the meanings defined.

Table 1: Definitions

Term	Definition
AS/NZS 5667.11	means the Australian Standard AS/NZS 5667.11 Water quality - sampling - guidance on sampling groundwater.
books	has the same meaning given to that term under the EP Act.
CEO	means Chief Executive Officer.
	CEO for the purposes of notification means:
	Director General Department administering the Environmental Protection Act 1986 Locked Bag 10 Joondalup DC WA 6919
	info@dwer.wa.gov.au
critical containment infrastructure	means the items of infrastructure listed in condition 1.
Critical Containment Infrastructure Report	means a report to satisfy the CEO that works of critical containment infrastructure have been constructed in accordance with the works approval.
Department	means the department established under section 35 of the <i>Public Sector Management Act 1994</i> and designated as responsible for the administration of Part V Division 3 of the EP Act.
discharge	has the same meaning given to that term under the EP Act.
emission	has the same meaning given to that term under the EP Act.
Environmental Compliance Report	means a report to satisfy the CEO that the conditioned infrastructure and/or equipment has been constructed and/or installed in accordance with the works approval.
EP Act	Environmental Protection Act 1986 (WA).
mg/l	Milligrams per litre
mbgl	Metres below ground level
EP Regulations	Environmental Protection Regulations 1987 (WA).
Quarterly	means the 4 inclusive periods from 1 April to 30 June, 1 July to 30 September, 1 October to 31 December and in the following year, 1 January to 31 March."

Term	Definition					
premises	the premises to which this licence applies, as specified at the front of this licence and as shown on the premises map in Schedule 1 to this works approval.					
prescribed premises	has the same meaning given to that term under the EP Act.					
Six monthly	means the two inclusive periods 1 October to 31 March and 1 April to 30					
	September					
suitably qualified	means a person who:					
geotechnical engineer	a) holds a Bachelor of Engineering recognised by the Australian Institute of Engineers; and					
	 b) has a minimum of five years of experience working in geotechnical engineering including experience in the design of tailings storage facilities. 					
SWL	Standing water level					
time limited operations	refers to the operation of the infrastructure and equipment identified under this works approval that is authorised for that purpose, subject to the relevant conditions.					
TSF	Tailings storage facility					
works approval	refers to this document, which evidences the grant of the works approval by the CEO under section 54 of the EP Act, subject to the conditions.					
works approval holder	refers to the occupier of the premises being the person to whom this works approval has been granted, as specified at the front of this works approval.					

END OF CONDITIONS

Schedule 1: Maps

Premises map

The boundary of the prescribed premises is shown in the map below (Figure 1).

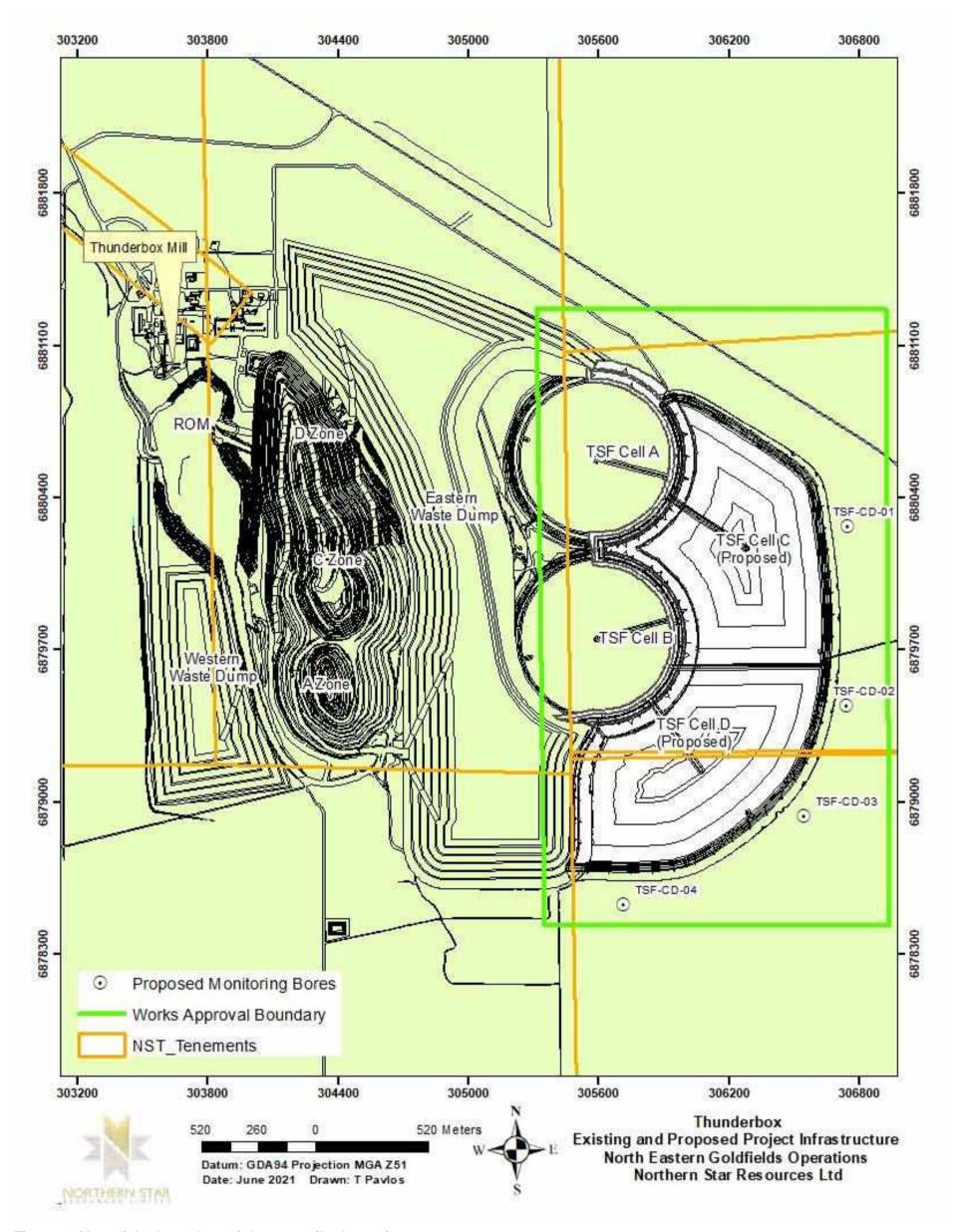


Figure 1: Map of the boundary of the prescribed premises

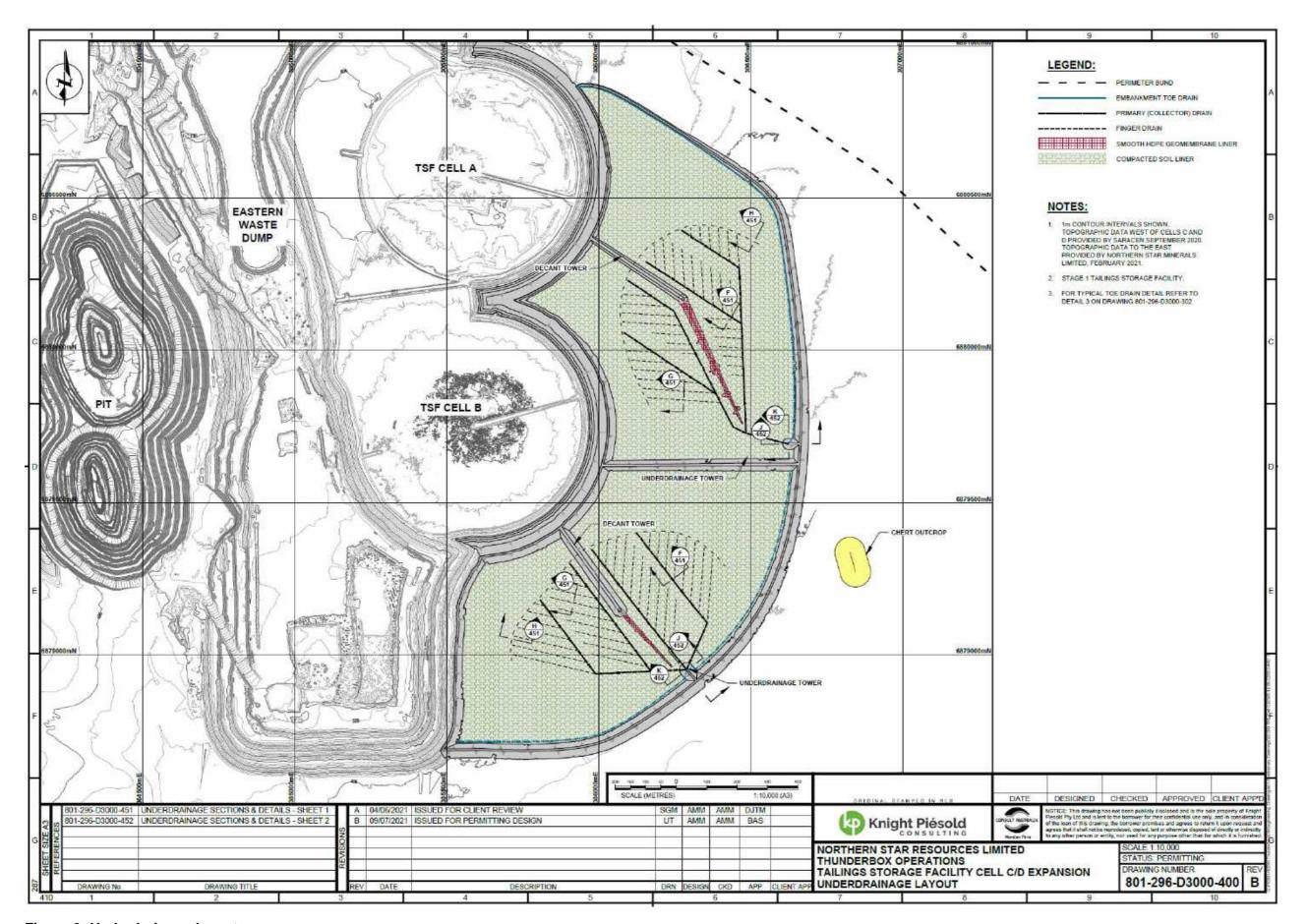


Figure 2: Underdrainage layout

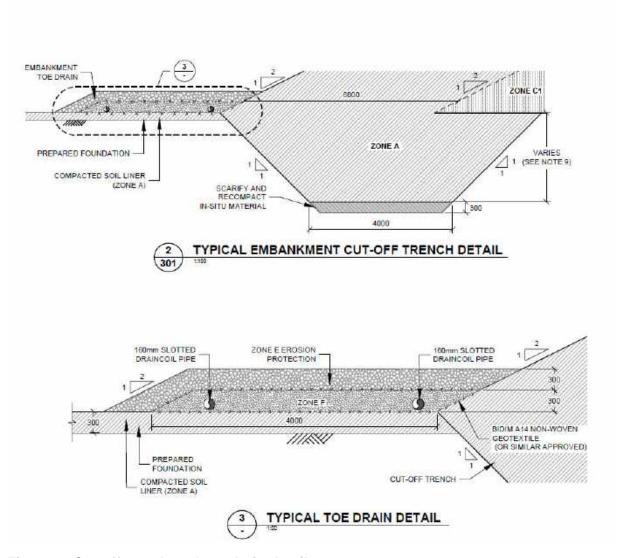


Figure 3: Cut off trench and toe drain details

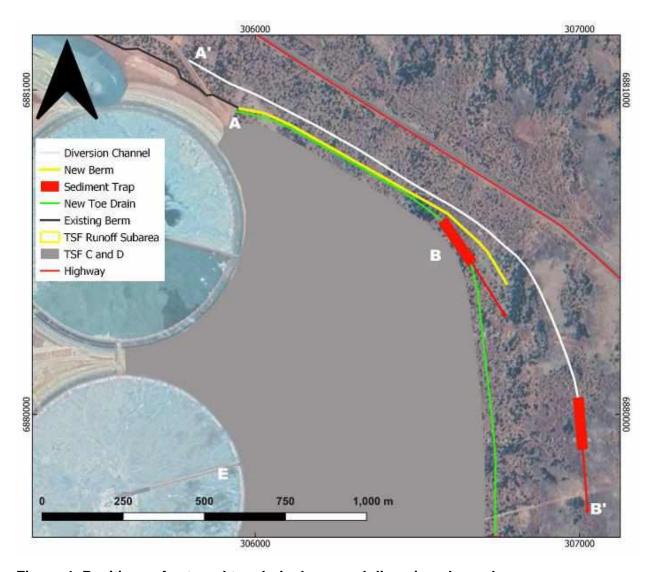


Figure 4: Positions of external toe drain, berm and diversion channel

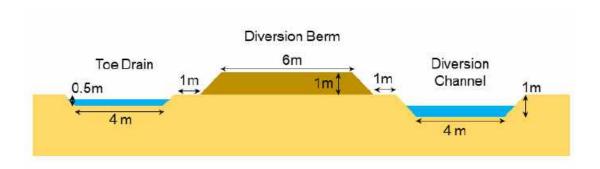


Figure 5: Schematic of toe drain, berm and diversion channel

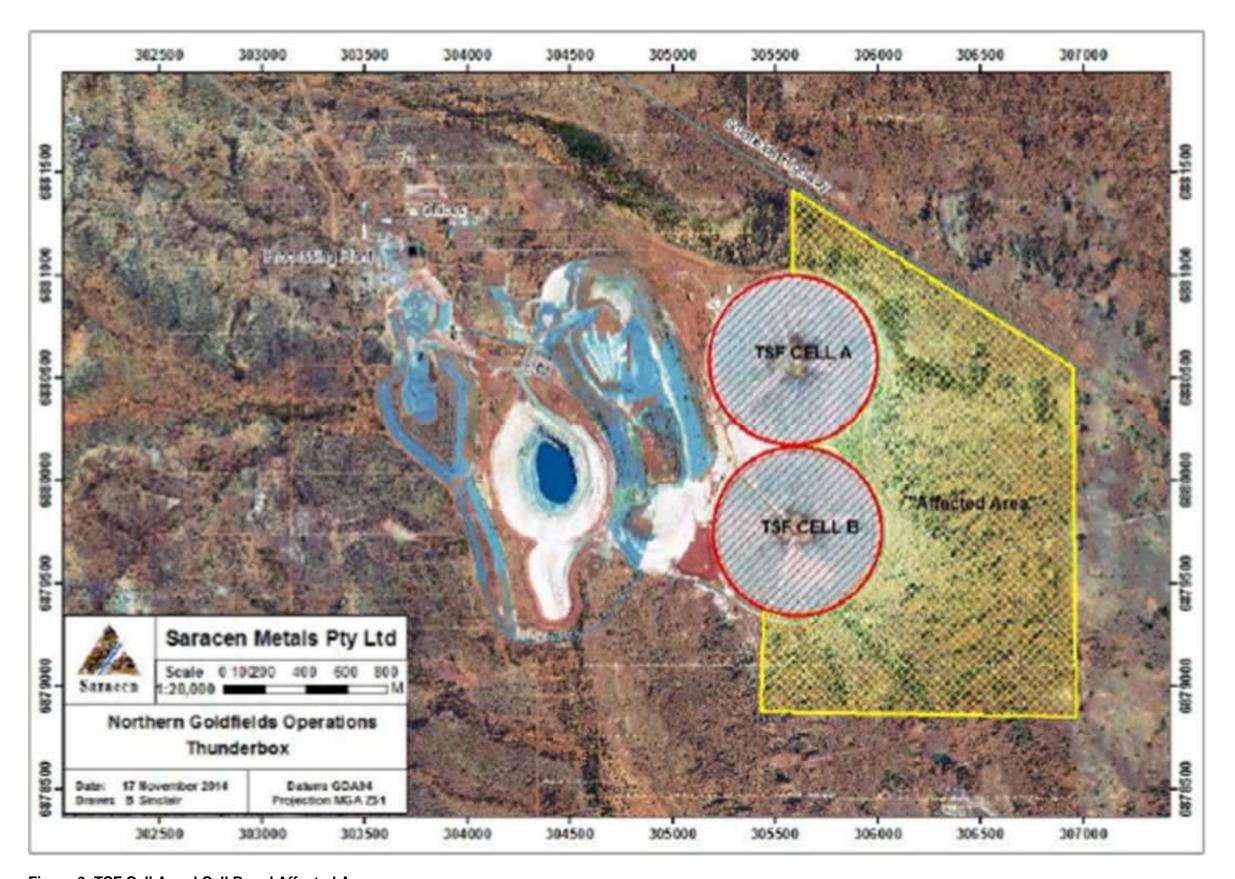


Figure 6: TSF Cell A and Cell B and Affected Area



8 Thunderbox TSF Life of Mine Assessment (Attachment 8B)



MEMORANDUM

To: Northern Star Resources Limited

Date: 28 July 2025

Our Ref: PE25-00885

KP File Ref.: PE801-00296/52-A djtm M25005

From:

RE: THUNDERBOX GOLD PROJECT – TAILINGS STORAGE FACILITY LIFE OF MINE ASSESSMENT

1. INTRODUCTION

As requested by Northern Star Resources, please find herein a Life of Mine (LOM) assessment of the Thunderbox Gold Project (Thunderbox) tailings storage facility (TSF).

The Thunderbox TSF comprises four cells, Cell A, Cell B, Cell C and Cell D, and occupies an area of approximately 301 ha. The historical development of the TSF is summarised below in Table 1.1.

Table 1.1: TSF historical development sequence

Stage	Ce	ell A	С	ell B	Cell C Cell D		Cell D	
	Crest mRL	Completed	Crest mRL	Completed	Crest mRL	Completed	Crest mRL	Completed
1	497.0	Oct 2002	493.5	Oct 2002	494.1	Nov 2023	494.1	Nov 2023
2	499.5	Jul 2003	496.0	Sept 2003				
3	502.0	Oct 2004	498.5	Oct 2005				
4	504.5	Nov 2006	501.0	Mar 2007				
5	507.0	Mar 2016	503.5	Apr 2017				
6	509.7	Apr 2018						
6a	503.5	Dec 2018	503.5	Jan 2019				
7	506.0	May 2019	506.0	May 2019				
8	508.8	Mar 2020	508.8	Aug 2020				
9	511.4	Jan 2021	511.4	Dec 2021				
10	514.6	Nov 2022	514.6	Jun 2023				
11	516.7	Ongoing	516.7	Ongoing				







Cell C and Cell D Stage 1 are currently in operation and are forecast to reach capacity in Q2/Q3 2026, after which deposition will be relocated to Cell A and Cell B Stage 11.

2. PERMITTING DESIGN

A permitting design was completed for Cell A and Cell B in October 2021 (Ref. 1), and Cell C and Cell D in July 2021 (Ref. 2). The levels determined as part of these designs are summarised below in tables 2.1 and 2.2.

Table 2.1: Permitted design levels (Cell A and Cell B)

Stage	Storage	Storage	Storage	Permitted
	Capacity	Capacity	Capacity	Level
	(Months)	(Mt)	(Mm ³)	(mRL)
10	12	4.76	3.53	514.6
11	8	3.21	2.38	516.7
Total	20	7.97	5.91	516.7

Note: Storage capacity based on throughputs and other information available at the time of design.

Table 2.2: Permitted design levels (Cell C and Cell D)

Stage	Stage Storage Capacity (Months)	Total Storage Capacity (Mt)	Storage Capacity (Mm³)	Permitted Level (mRL)
1	24	9.5	6.15	494.1
2	24	9.5	6.08	498.2
3	12	4.9	3.18	500.2
4	12	5.8	3.75	502.6
5	12	6.0	3.87	505.0
6	4	2.0	1.29	505.8
Total	88	37.70	24.32	505.8

Note: Storage capacity based on throughputs and other information available at the time of design.

3. LIFE OF MINE

3.1 GENERAL

As requested by NSR an update to the forecast levels was completed to reflect the following:

- Amendments to the tailing's throughput.
- Estimated achieved in-situ dry tailings densities.
- · Achieved beach slope estimated from capacity reviews.
- As Built records for the existing cells.

The revised stage storages for each of the cells are provided in figures 3.1 and 3.2. The differences between the designs are summarised in the sections below.

3.2 THROUGHPUT

A comparison between the density used in the permitting design and the revised basis is summarised below in Table 3.1. The throughputs are shown in figures 3.3 to 3.5.



Table 3.1: Throughput comparison

	Revised Basis			Pe	ermitting Bas	sis
Financial	To TSF	Paste	Total	To TSF	Paste	Total
Year		Backfill			Backfill	
	(Mt)	(Mt)	(Mt)	(Mt)	(Mt)	(Mt)
2021	2.88	0.04	2.92	2.95	0.05	3.00
2022	2.52	0.53	3.05	2.15	0.85	3.00
2023	3.28	0.56	3.84	4.75	1.04	5.79
2024	4.82 ³	0.89^{3}	5.71 ³	4.85	1.15	6.00
2025	6.06	0.83	6.89	4.86	1.14	6.00
2026	4.97	1.33	6.30	4.97	1.03	6.00
2027	5.29	1.04	6.34	5.35	0.65	6.00
2028	5.56	0.97	6.53	5.75	0.25	6.00
2029	6.00	0.55	6.55	6.00	-	6.00
2030	5.74	0.79	6.53	6.00	-	6.00
2031	6.11	0.42	6.53	6.00	-	6.00
2032	5.72	0.27	5.99	-	-	-
2033	3.23	-	3.23	-	-	-
Total	62.18	8.22	70.41	53.63	6.16	59.79

Note:

- 1 Throughputs provided by NSR during the design.
- 2 Red values represent actual throughput recorded to the TSF as provided by NSR.
- 3 No accurate record provided by NSR.

Total Tonnage (Paste and Tailings to TSF):

From FY2025 onwards the total throughput (paste and tailings reporting to the TSF) for the revised schedule is higher than the permitting design. The total tonnage (paste and tailings reporting to the TSF) for the revised schedule is also higher than the permitting design.

Tailings to the TSF:

The throughput reporting to the TSF in the revised schedule is similar to the permitting design, however the total tonnage is higher for the revised schedule. The duration of deposition has also increased to allow for the increased total tonnage. Assuming that the beach slope and in-situ tailings density are unchanged then the embankment levels will increase, due to the higher total tonnage.

Paste tailings:

The throughput of paste tailings is similar between the revised schedule and the permitting design; however, the paste life has extended resulting in a higher overall paste tailings tonnage.



3.3 DENSITY

A comparison between the density used in the permitting design and the revised basis is summarised below in Table 3.2.

Table 3.2: Density comparison

Cell	Measured Value (t/m³)	Permitting Design (t/m³)
Α	1.37	1.35
В	1.37	1.35
С	1.61	1.55
D	1.61	1.55

Note: Density based on capacity reviews completed for the existing facility to reflect actual conditions.

The revised in-situ tailings dry density has improved from the permitting design basis; therefore, the existing facilities will have more storage capacity which will extend the estimated filling date for the facility. The increased densities will assist in reducing the final embankment height required for the Thunderbox TSF.

3.4 BEACH SLOPE

A comparison between the density used in the permitting design and the revised basis is summarised below in Table 3.3.

Table 3.3: Beach slope comparison

Cell	Measured Value (_H:1V)	Permitting Design (_H:1V)
A	150	125
В	150	125
С	150	100
D	150	100

Note: Beach slopes based on capacity reviews completed for the existing facility to reflect actual conditions.

The tailings beach slope is flatter in the revised design compared with the permitting design. The tailings beach slope assumptions have been based on the available beach surveys at the time of design to inform the beach slope used in design. The tailings beach slope is impacted by many factors such as throughput, grind size, percent solids, ore type and ore weathering therefore changes to the achieved beach slope are possible. Regular capacity reviews are recommended to ensure that any changes to beach slope are captured in design and raise planning.

The flatter beach slope has increased the tailings storage capacity which has extended the filling date. It is noted that the stormwater capacity has reduced due to the flatter beach slope, which has been considered in current raise designs. The flatter beach slope will assist in reducing the final embankment height required for the Thunderbox TSF.

3.5 AS BUILT CAPACITY

Following the construction of Cell C and Cell D Stage 1 a capacity assessment was completed using the as-built survey which indicated a 0.7 Mm³ (+11%) increase to storage capacity from design (Ref. 3).



The improvement in storage capacity was due to clearing, grubbing, topsoil removal and borrowing from within the TSF basin, which has increased the storage capacity. The increase in the Stage 1 storage capacity will help lower the overall embankment levels.

3.6 REVISED LEVELS

A summary of the permitting design levels presented in the reports (Ref. 1 and 2) compared with the current estimated levels are provided below in tables 3.4 and 3.5.

Table 3.4: Permitted design levels (Cell A and Cell B)

Stage	Current Design Estimate	Permitted Level	Change
	(mRL)	(mRL)	(m)
10	514.6	514.6	-
11	516.7	516.7	•
12	518.7*		-
13	520.7*	*	-
14	522.8*		-

Note: *No permitting design completed for stages 12+.

Table 3.5: Permitted design levels (Cell C and Cell D)

Stage	Current Design Estimate	Permitted Level	Change
	(mRL)	(mRL)	(m)
1	494.1	494.1	-
2	498.2	498.2	-
3	500.6	500.2	+0.4
4	503.0	502.6	+0.4
5	504.0	505.0	-1.0
6	-	505.8	-1.8

It is noted that these levels should be considered indicative, as changes to throughput, tailings densities and beach slope will impact the future levels. A detailed design report will be prepared for all future levels based on the best available information at the time, which will estimate the required level for each raise.

Cell A and Cell B:

The Cell A and Cell B embankment levels are the same as the permitting design levels. A permitting design should be completed for Cell A and Cell B, and submitted to the regulator for approval to allow the construction of stages 12, 13 and 14.

Cell C and Cell D:

The revised estimate for the Stage 1 and Stage 2 Cell C and Cell D embankment levels are the same as the permitting design. Stage 3 and Stage 4 are slightly higher (+0.4 m). Stage 5 is lower (-1.0 m) than the permitting design as the overall density achieved in the facility is improved and a flatter beach slope has increased the tailings storage capacity. Stage 6 is no longer required based on the revised estimates, and the final level is 1.8 m lower than the permitting design basis, due to higher in situ dry density, flatter beach slope and increased storage capacity.



CONCLUSIONS 4.

The following conclusions are drawn from the above assessment:

- i. The throughput reporting to the TSF in the revised schedule is similar to the permitting design, however the total tonnage is higher for the revised schedule. The duration of the processing schedule has also extended compared with the permitting design.
- The measured in-situ tailings dry density has improved which has assisted in ii. lowering the final TSF embankment level.
- The measured beach slope has flattened which has assisted in lowering the iii. final TSF embankment level.
- iv. The Cell C and Cell D Stage 1 storage capacity is higher than the permitting design which has assisted in lowering the ultimate embankment level.
- ٧. The forecast Thunderbox TSF levels are similar to the permitting designs submitted previously. The main differences are associated with updates to account for changes in the observed in-situ tailings density, beach slope. revised throughput and total tonnage.
- The Cell A and Cell B embankment levels are the same as the permitting Vi. design levels. A permitting design should be completed for Cell A and Cell B, and submitted to the regulator for approval to allow the construction of stages 12, 13 and 14.
- vii. The revised estimate for the Stage 1 and Stage 2 Cell C and Cell D embankment levels are the same as the permitting design. Stage 3 and Stage 4 are slightly higher (+0.4 m) and Stage 5 is lower (-1.0 m) than the permitting design. Stage 6 is no longer required based on the revised estimates and the ultimate level is 1.8 m lower.

We trust that this is sufficient for your current requirements, should you require additional information or have questions please contact us.

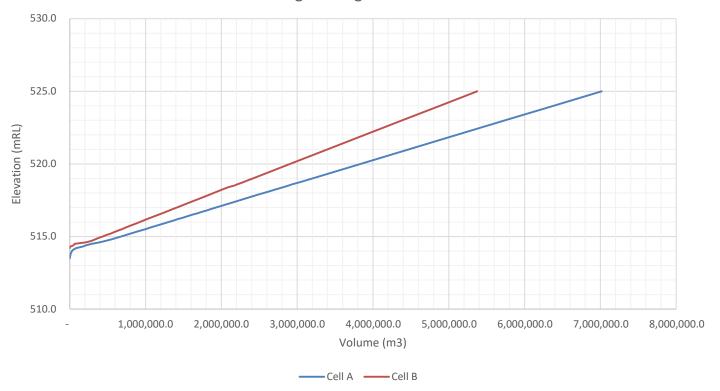


REFERENCES

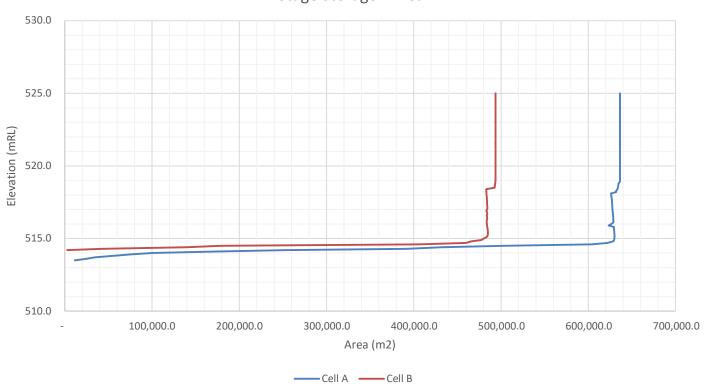
- 1. Knight Piésold Report. Ref PE801-00296/22, "Thunderbox Gold Mine Tailings Storage Facility Expansion Cell A and Cell B Stages 10 and 11 Final Design", Rev 0, October 2021.
- 2. Knight Piésold Report. Ref PE801-00296/17, "Thunderbox Gold Mine Tailings Storage Facility Cell C and Cell D Permitting Design Report", Rev 3, July 2021.
- Knight Piésold Report. Ref PE801-00296/36, "Thunderbox Gold Mine Tailings Storage Facility Cell C and Cell D Stage 1 Construction Report", Rev 1, April 2024.



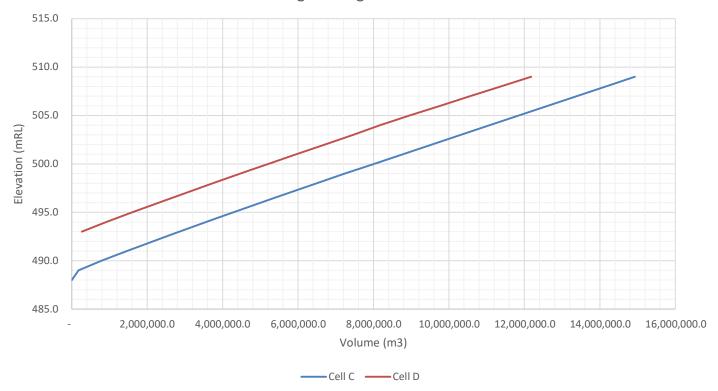
Stage Storage - Volume



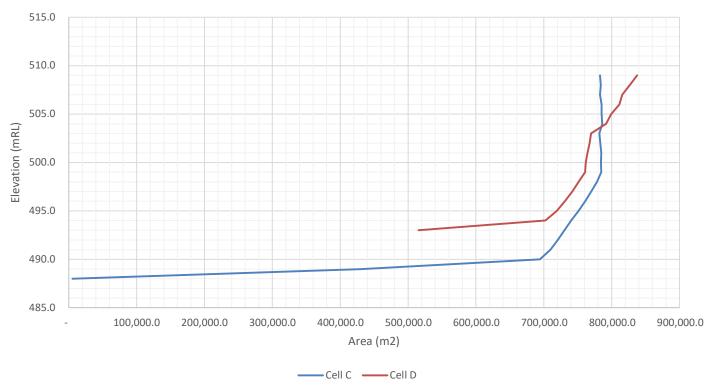
Stage Storage - Area



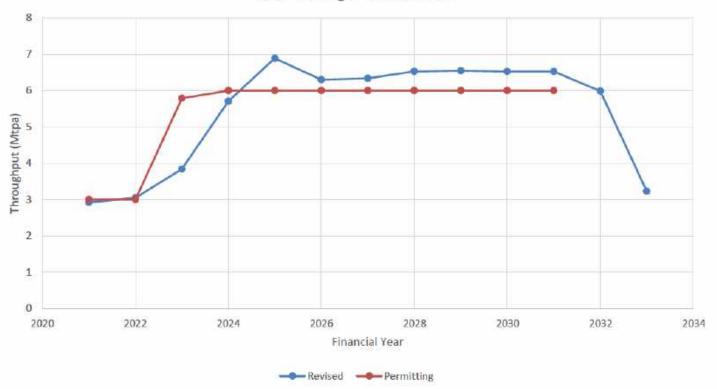
Stage Storage - Volume



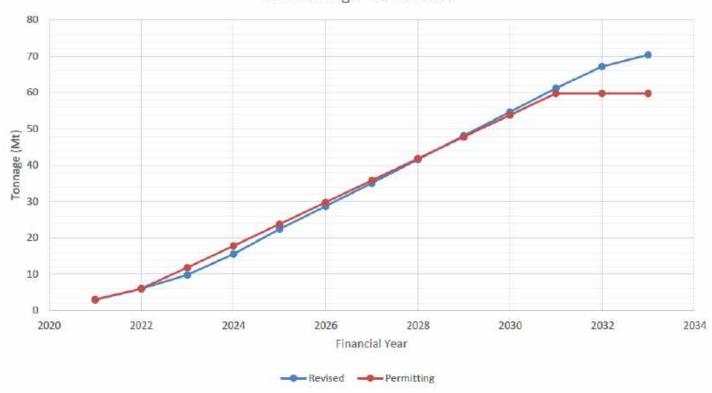
Stage Storage - Area



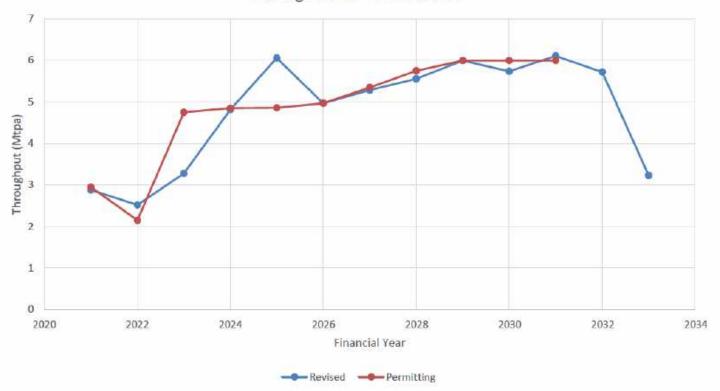
Total Tonnage - Incremental



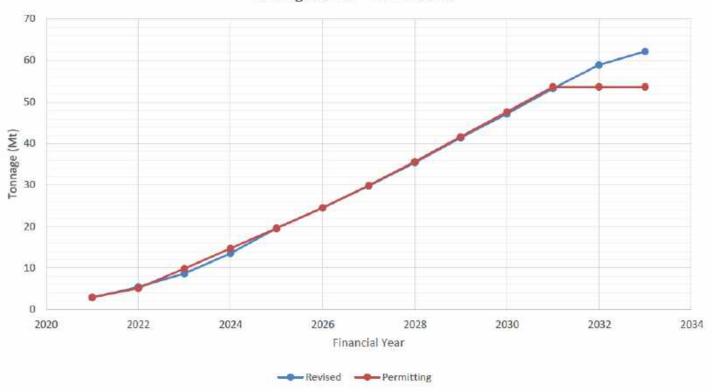
Total Tonnage - cumulative



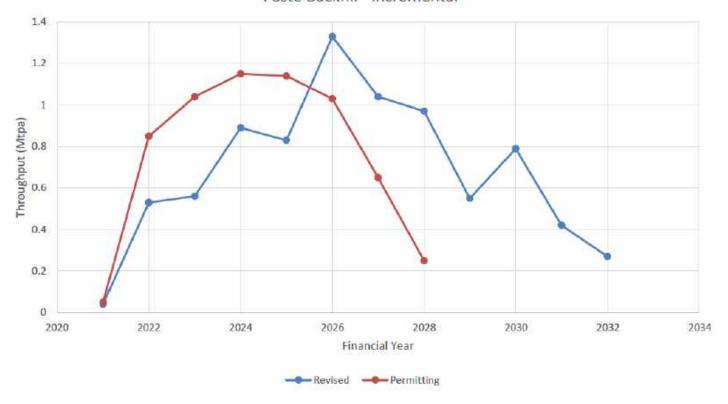
Tonnage to TSF - incremental



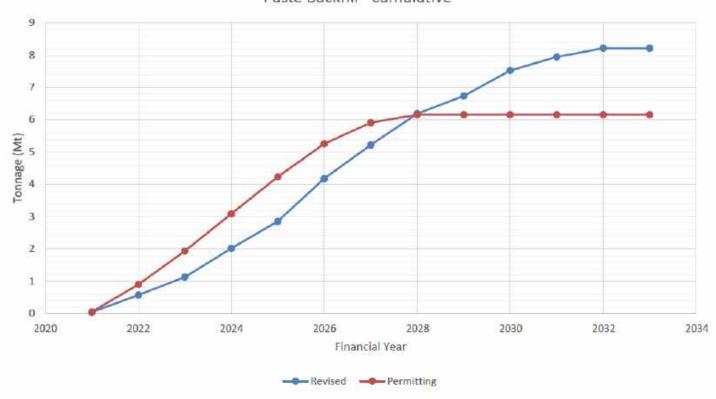
Tonnage to TSF - cumulative



Paste Backfill - incremental



Paste Backfill - cumulative





9 Thunderbox TSF Cell C & D Permitting Design (Attachment 8C)

NORTHERN STAR RESOURCES LIMITED THUNDERBOX GOLD MINE



TAILINGS STORAGE FACILITY CELL C AND CELL D PERMITTING DESIGN

PREPARED FOR:

Northern Star Resources Limited Level 1/388 Hay St, Subiaco, WA 6008

PREPARED BY:

Knight Piésold Pty Limited
Level 1 184 Adelaide Terrace



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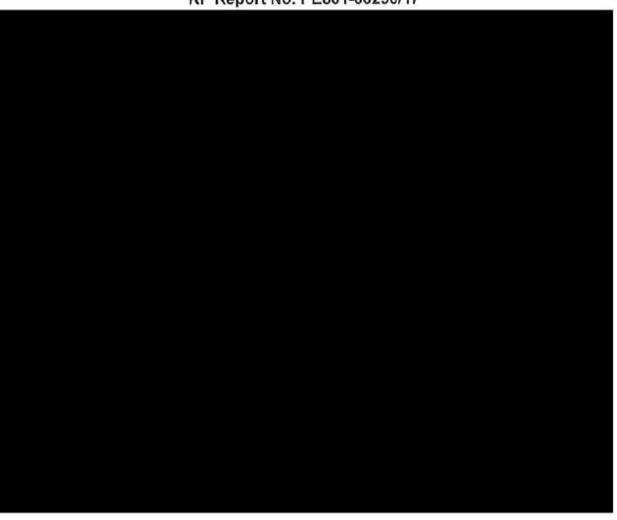
NORTHERN STAR RESOURCES LIMITED

THUNDERBOX GOLD MINE

TAILINGS STORAGE FACILITY CELL C AND CELL D PERMITTING DESIGN REPORT

KP Job No. PE801-00296/31

KP Report No. PE801-00296/17





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1. INTRODUCTION

1.1 GENERAL

The Thunderbox Gold Mine south of the town of Leinster, Western Australia is currently being operated by Northern Star Resources Limited (NSRL). Knight Piésold (KP) was engaged to undertake the engineering design for an expansion of the existing the tailings storage facility (TSF) which involves horizontal expansion integrating the Eastern Waste Dump and the existing Cell A and Cell B to form two additional cells, namely Cell C and Cell D. The site general arrangement is shown on Drawing 801-296-D3000-100.

The project is owned by NSRL and was previously on care and maintenance (since August 2007). The redevelopment of the operation commenced in 2015 following the completion of a feasibility study. The plant was re-commissioned in February 2016 and has been in continuous operation since then. The current plant throughput is 3.0 Mtpa and is ramping up to 5.79 Mtpa in FY22 and FY23 then 6.0 Mtpa in FY24. The plant incorporates a single-stage crusher, a SAG mill and a ball mill as well as conventional CIL leaching and elution circuits.

TSF Cell A Stage 9 raise was completed in January 2021 and Cell B is being used for tailings storage. With the increased plant throughput and mine reserves, it is proposed that the existing TSF will expand vertically and horizontally, further integrating the Eastern Waste Dump and the existing TSF Cell A and Cell B, by forming two additional cells (Cell C and Cell D), to reduce the overall footprint cleared, ensure improved stability by constructing a waste rock buttress for cells A and B, and achieve construction material efficiencies. Vertical expansion (Cell A and Cell B) is reported in a separate study. This report presents the mining proposal and works approval design for the proposed horizontal tailings storage expansion (Cell C and Cell D) only.

The Eastern Waste Dump profile are combined with the existing TSF to form two cells C and D, and two central decants. Utilising the two cells:

- Provides sufficient tailings storage capacity for an additional 37.7 Mt of tailings from July 2022);
- Provides efficient deposition areas to increase densities and reduce embankment levels;
- Reduces construction earthwork quantities by utilising the existing cell walls and the Eastern Waste Dump;
- Reduces the overall land take area (compared with standalone new storages);
- Provides greater flexibility in the management of tailings; and



Provides buttress requirement to the existing Cell A and Cell B.

1.2 PROJECT LOCATION AND DESCRIPTION

The Thunderbox Operations are centred on the Thunderbox Open Pit and CIL gold treatment plant, located 45 km south of the town of Leinster in Western Australia and immediately adjacent to the sealed Goldfields Highway.

The existing TSF (Cell A and Cell B) is located approximately 2 km southeast of the plant. The cells abut each other with Cell A located immediately north of Cell B. The Cell C and Cell D are proposed to be abutting the existing TSF Cell A, Cell B and the Eastern Waste Dump. The TSF (vertical expansion and horizontal expansion) is bounded to the east by the Goldfields Highway and to the immediate west by the Eastern Waste Dump. The site general arrangement is shown on Drawing 801-296-D3000-100.

The site slopes gently to the south-east at an overall slope of approximately 1%.

1.3 DESIGN OBJECTIVES

The design objectives for the TSF expansion are summarised as follows:

- Permanent and secure containment of all solid tailings materials generated over the life of the project;
- Continuous removal and reuse of free water from the facility;
- Improve the future stability of the existing cells, Cell A and Cell B;
- Provide excess storage capacity to retain design storm events;
- Incorporate safety into the design, construction and operation of the facility;
- Reduced environmental impact with rapid and effective rehabilitation; and
- Provide ease of operation.

A summary of the key policies and guidance documents used for the design of the tailing storage are summarised as follows:

- Guidelines on the Consequence Categories for Dams, ANCOLD, October 2012 (Ref. 1);
- Guidelines on Tailings Dams Planning, Design, Construction, Operation and Closure, ANCOLD, July 2019 (Ref. 2);
- Guidelines for Design of Dams and Appurtenant Structures for Earthquake, ANCOLD, May 2019 (Ref. 3);
- Department of Mines Industry Regulation and Safety (DMIRS) "Code of Practice
 Tailings Storage Facilities in Western Australia" 2013 (Ref. 4);



- Guide to Preparation of a Design Report for Tailings Storage Facilities (TSFs) -DMIRS, August 2015 (Ref. 5); and
- Guide to Departmental Requirements for the Management and Closure of Tailings Storage Facilities (TSFs) - DMIRS, August 2015 (Ref. 6).

1.4 TSF DESIGN PARAMETERS AND ASSUMPTIONS

The overall design criteria are maintained from the previous TSF permitting design (Ref.7) as summarised in Table 1.1. These design criteria are in accordance with the WA Department of Mines Industry Regulation and Safety (DMIRS) guidelines and the Australian National Committee on Large Dams (ANCOLD) guidelines.



Table 1.1: TSF design criteria

TSF Design			
Storage Capacity - Final Stage	37.7 Mt of dry tails (from July 2022).		
Freeboard	Greater of: (i) 0.50 m above maximum tailings elevation, or (ii) 0.50 m above required stormwater capacity elevation.		
Stormwater Capacity (containment) - Short duration	1% Annual Exceedance Probability (AEP) 72 hour storm event superimposed over average rainfall sequence with no release, evaporation or decant.		
- Long duration	1% AEP, 12 months wet rainfall sequence.		
Earthquake Loading - Operating - Final -Closure	Operating Basis Earthquake (OBE) – 1 in 475 year. Safety Evaluation Earthquake (SEE) - 1 in 1,000 year. Maximum Credible Earthquake (MCE) – greater than 1 in 10,000 year.		
	Operations		
Production Throughput	3.0 Mtpa, increasing to 6.0 Mtpa. 0.07 Mtpa to 1.31 Mtpa to underground paste backfill. The remaining tailings reporting to TSF cell(s).		
Production Days/Year	330 (90.4% availability).		
Slurry Characteristics	62% solids by weight. SG solids = 2.7. SG liquor = 1.006. Slurry average settled density = 1.60 t/m³. Permeability of 4 x 10 ⁻⁷ m/s.		
Deposition	Deposition from perimeter embankments towards centre of basin. Supernatant pond maintained central to the facility.		
Water Management	Vertical decant tower system for removal of supernatant solution. Return to the plant via submersible pump. Underdrainage system drains by gravity into underdrainage tower. Return to supernatant pond, via submersible pump.		



Table 1.1 (cont.): TSF design criteria

Embankment construction			
General	Cell C and Cell D downstream construction for the first two stages and upstream construction in subsequent stages		
	Downstream construction where buttressed against Eastern Waste Dump		
	Divider embankment centreline construction		
Construction Description			
-Cut off trench	Embankment upstream toe cut-off key and drain on embankment sections where upstream construction planned		
-Embankment	Compacted Oxide mine waste for Stage 1 and Stage 2, tailings will be utilised for Stage 3+ with upstream construction – Cell C and Cell D		
	Zoned embankment constructed from selected mine waste/tailings comprising an upstream low permeability zone where buttressed against Eastern Waste Dump		
-Underdrains	Slotted HDPE draincoil 160 mm diameter along toe of embankment and basin underdrainage system, reporting to an underdrainage tower		
Construction Materials			
Low permeability fill (Zone A)	Selected mine waste or tailings.		
Structural fill (Zone C/C1)	Run of mine (ROM) waste, traffic compacted.		
 Filter material (Zone F) 	Imported or screened sand.		
- Erosion protection (Zone E)	Select and/or screened from blasted rock mine waste.		
- Clean rock (Zone G)	Select and screened from blasted rock mine waste.		
- Crest wearing course	Select borrow material		
	TSF Rehabilitation		
Final Embankment Slopes	1V:3.5 H (overall) 15.9° slope		
Capping	Mine waste subsoil capillary break followed by topsoil, re-vegetation.		
Cover profile	Shaped to achieve a water harvesting structure towards decant towers.		

1.5 PROJECT HISTORY

The existing TSF comprises two adjacent operating cells (Cell A and Cell B). Both of them are circular in shape and approximately 50 ha in area. The TSF was designed by DE Cooper and Associates Pty Ltd in 2001. The TSF had an operational life of 5 years and tailings deposition ceased in 2007. The mine was recommissioned in 2015, processing and tailings deposition recommenced in January 2016. Biennial audits were conducted by Coffey Mining Pty Ltd (Coffey) from 2007 to 2013 and subsequently by Knight Piésold. The development of tailings storage at Thunderbox is summarised in Table 1.2.



 Table 1.2:
 TSF construction summary

Stage		Cell A		Cell B
	Crest RL	Construction completed	Crest RL	Construction completed
1	RL497.0 m	October 2002	RL493.5 m	October 2002
2	RL499.5 m	July 2003	RL496.0 m	September 2003
3	RL502.0 m	October 2004	RL498.5 m	October 2005
4	RL504.5 m	November 2006	RL501.0 m	March 2007
5	RL507.0 m	March 2016	RL503.5 m	April 2017
6	RL509.7 m	April 2018		
6 Expansion	RL503.5 m North Void	December 2018	RL503.5 m South Void	January 2019
7 Expansion	RL506.0 m North Void	May 2019	RL506.0 m	May 2019
8 Expansion	RL508.8 m North Void	March 2020	RL508.8 m	August 2020
9 Expansion	RL511.4 m	January 2021	RL511.4	Ongoing



2. SITE CHARACTERISTICS

2.1 GEOLOGY

The Thunderbox deposit is located within the regional north-north-west trending Thunderbox shear zone.

In addition to the major north-north-west trending ductile shear system, there is a pervasive network of relatively late north-north-east trending brittle faults, one of which separates the mineralised zones of the Thunderbox gold deposit. The shear is vertical to steeply west dipping (Ref. 8).

2.2 FLORA AND FAUNA

One species of Significant Flora has been identified in the area. This is *CSauropus* sp. Woolgorong (M. Officer s.n. 10.8.94) (Ref.9). As most populations are located well away from the proposed areas of the mine pits and associated infrastructure, it is considered unlikely that they would be affected by any mine operations.

2.3 CLIMATE

The region is in a semi-arid environment and has a dry climate with hot summers and cool winters. The average annual rainfall is about 224 mm. February and March tend to be the wettest months, September and October are the driest months. The majority of the annual average evaporation of 2,865 mm occurs from October to March. The average rate in January is 13.0 mm per day, while on a hot, windy day the evaporation can be over 20.0 mm. During the winter the average daily evaporation decreases to 3.7 mm. The relative humidity averages less than 30% during summer while the winter figures are typically around 70 %.

A detailed baseline climatology assessment for the project is provided in Appendix A with design storms for the 1% Annual Exceedance Probability (AEP) 72-hour event of 201 mm and the Probable Maximum Precipitation (PMP) 24-hour event of 560 mm.

2.4 SEISMICITY

A seismic hazard assessment has been carried out for the site based on a technical publication by Geoscience Australia 'Atlas of Seismic Hazard Maps of Australia' (Ref. 10).



Based on the Hazard Rating of 'SIGNIFICANT' Category 1 facility, the Operating Basis Earthquake (1 in 475 year peak ground acceleration) was estimated as 0.02 g, implying a low seismic hazard. The Safety Evaluation Earthquake (1 in 1,000 year peak ground acceleration) was estimated as 0.03 g. The Maximum Credible Earthquake was calculated as 0.16 g in a deterministic analysis. This was updated for the recent ANCOLD 2019 seismic guidelines.

2.5 WATER CATCHMENTS

No well-defined drainage channels were identified at the TSF site. Rainfall runoff is likely to occur as sheet-flow across the site as the flatness of the ground is unlikely to promote significant channelling of surface water.

The TSF expansion is designed to be a paddock arrangement and the catchment areas will be limited to the surface areas of the facility. However, the waste dump on the west side will need to be shaped to ensure this.

2.6 GROUNDWATER

Groundwater within the project area is generally within 28m of the natural ground level and typically flows south east across the site. Existing monitoring bore data indicates that cyclic trend can be seen where ground water levels peak during the wet season before reduce over the subsequent dry season. Groundwater within the project area is saline, and will be suitable for livestock use.



3. RISK / HAZARD CLASSIFICATION

3.1 GENERAL

The hazard/consequence category for the Thunderbox TSF has been assessed on the basis of the following two guidelines:

- Guidelines on Tailings Dams Planning, Design, Construction, Operation and Closure, ANCOLD, July 2019 (Ref. 2); and
- Department of Mines Industry Regulation and Safety (DMIRS) "Code of Practice
 Tailings Storage Facilities in Western Australia" 2013 (Ref. 4).

3.2 POPULATION AT RISK (PAR) ASSESSMENT

3.2.1 General

As part of the calculation of the hazard/consequence category, an assessment of the Population at Risk (PAR) was undertaken. A dam break analysis was undertaken and is reported in Section 7.12. The impact was assessed to determine the relevant PAR. The PAR value then was incorporated into the hazard/consequence rating, as outlined in Section 3.5.

3.2.2 PAR Calculation

Tailings release from a dam break was modelled and the impact area was determined immediately downstream of the embankment. This impact area was used to calculate the PAR for this event.

There are a number of factors impacting the PAR calculation as follows:

- The tailings solids indicate low health risk thus dust from the facility does not pose a short term or long-term health risk;
- The tailings may have some residual cyanide in solution and supernatant is saline which may impact vegetation growth;
- The mine haul road to the TSF and traffic flow/supervision during construction or stockpiling activities;
- The Goldfield Highway public road downstream of the TSF;
- Once per shift inspections by Processing Department are conducted at the TSF which requires access along the pipeline access road; and
- No notable camps, tracks, permanent offices are located within the immediate downstream.



The PAR is restricted to potential loss of life due to the physical outcomes resulting from a facility failure (i.e. dam break events) and not health or environmental issues. The PAR is estimated to be 0.51, based on the calculation of number of vehicles and time within the impact zone (including a 5% contingency), as summarized in Table 3.1. On the basis of this estimation the PAR category is defined as '<1'.



Table 3.1: Calculation of overall PAR

Population Category	Duration in Impact Zone	Frequency of Event	Calculated PAR
	(minutes)	(No per year)	
Permanent Dwellings	No dwellings or Residential structures in Impact Area	Nil	
	0	0	0
Mine Buildings	Nil (refer to note)	Nil	
	0	0	0
Goldfield Highway	Length in Impact Zone = 3,200 m	Number of people in area = 2 per car / per trip / per day	
	Car Speed = 110 km/hr	Operating days per year = 365	
		Trips per day per car = 197 (based on Main road statistics)	
	Time passing Impact Zone = 1.74 min	Number of Persons passing Impact Zone per year = 143,810	0.48
TSF Inspection	Length in Impact Zone = 500 m	No of Persons per Car = 2	
	Car Speed = 20 km/hr	Operating days per year = 365	
		Trips per day = 4	
	1.50	2920	0.008
Parks	No parks in area	Nil	
	0.00	0	0.000
Campers	No recognised camping area in impact zone	Nil	
	0.00	0	0.000
Tourists to Mine Site	No recognised public roads in impact zone	Nil	
	0.00	0	0.000
Unclassified			
(5% contingency)			0.025
		TOTAL	0.51

Note: there are currently no proposed mine infrastructure, buildings or facilities within the impact zone.



3.3 DMIRS HAZARD RATING

The TSF horizontal expansion classification was assessed by the methods set out in the DMIRS guidelines 2013 (Ref. 4) to determine the category of the facility over the life. The hazard rating was determined by investigating the potential impact of either uncontrolled release, seepage of supernatant water, or complete embankment failure and its impact on humans, stock, and environment as well as the economic loss.

The hazard rating was selected as either high, significant or low, based on the effect on humans, stock, the environment and economic loss. This considered the scenarios of uncontrolled release and embankment failure but did not consider the likelihood of occurrence. The major aspects considered are summarized in Table 3.2 below.



Table 3.2: DMIRS guidelines on hazard rating

Type of impact or damage	Hazard Rating
	Hazaru Kating
Loss of Human Life or personal injury No habitable structures or recognised camping areas are located in the immediate vicinity downstream of the embankment. Section 3.2 calculated the PAR as being '<1' for TSF failure events.	Low
Adverse human health due to direct physical impact or contamination of the environment (e.g. chemical or radiation denigration of water, soil, air) No groundwater production bores for human consumption in the downstream seepage zone of the TSF. Thus, loss of human life due to consumption of local ground water is not expected. Contamination of habitable areas is expected as the supernatant water quality from seepage or water release does not pass drinking water guidelines.	Low
Loss of assets due to direct physical impact or contamination of the environment (e.g. chemical or radioactive pollution of water, soil or air) Tailings geochemistry results indicated the sample had a moderate number of enrichments, with the level of enrichment varying from slight to high. Arsenic and Sulfur were found to be highly enriched, with antimony significantly enriched and silver, lead and selenium slightly enriched.	
Monitoring Bore data for the existing cells A and B have shown no significant elevations in heavy metals or cyanide, and are well below the DWER licence limit. It is noted that cells A and B comprise of recompacted in situ basin material and include similar seepage measures as proposed for cells C and D.	Medium
Groundwater is saline and is suitable for livestock use.	
There may be potential for temporary loss of assets of mine infrastructure located downstream of the main embankment in the event of a dam break.	
Loss of TSF storage capacity is possible however with multiple cells and repair being practicable relatively quickly addressed at a cost for earthworks.	
Damage to items of environmental, heritage or historical value due to direct physical impact or contamination of the environment (e.g. chemical or radioactive pollution of water, soil or air)	
The site does not contain any identified Threatened Ecological or Priority Ecological Communities.	Low
Temporary adverse effects on fauna and flora is possible.	
No known heritage or historical value is identified.	
Vegetation impacts of hypersaline water	



The TSF classification category is a function of the highest hazard identified above and the maximum embankment height for Cell C and D (approximately 22 m) using the matrix provided in Table 2 of DMIRS 2013. As a result, the facility is considered a "**Medium Category 1**" facility.

3.4 ANCOLD CONSEQUENCE CATEGORY

An assessment has been conducted to determine the severity level of impacts based on ANCOLD (Ref. 2). An assessment to determine the severity level of impacts from a large-scale failure of the facility "Dam Failure Severity Level" and a spillway flow or other water release "Environmental Spill Severity Level" was conducted. Similar to the DMIRS Hazard Rating, the likelihood of failure is not considered in the ANCOLD Consequence Category Assessment. The Population at Risk (PAR) as detailed in Section 3.2, being '<1' was the basis of this assessment.

Table 3.3: ANCOLD Guideline on Severity Level (Dam Failure)

Damage Type	Severity Level		
Infrastructure (dam, houses, commerce, farms, community)			
Damage to the storage itself is expected to be repairable relatively quickly but at a cost for earthworks (<\$10M). Site roads, pipework and power supply cables may be inundated but repairs are expected to be minor.	Minor		
Business Importance			
Processing operations could continue as the second TSF cell would likely be operational.			
Significant costs in recovering/replacing equipment and re-establishing working areas.	shing Major		
There would likely be severe reaction from the community and some loss of business credibility hence "crippling to business" was selected but bankruptcy unlikely.			
Public Health			
No contamination of habitable areas is expected as there are no structures or recognised camping areas located in the vicinity.	Minor		
Emergency management costs associated with response to inundation of explosives storage facility and clean-up.			
Social Dislocation			
No social impact is expected as there are no structures or public facilities located in the vicinity (<100 person or <20 business months).	Minor		
Impact Area			
The impact area would be expected to be contained within less than 20 km². Minor sediment may continue to flow downstream due to surface runoff however the site is semi-arid, therefore this is not considered to be an immediate impact area.	Major		
Impact Duration			
Released tailings can be recovered and affected areas remediated in a short period of time with no long-term environmental damage.	Minor		
Impact not expected to exceed 1 year in duration.			
Impact on Natural Environment			
Potential of tailings to inundate surface water features used by native fauna.	Medium		



A secondary assessment of an Environmental Spill consequence has been assessed for the case of water spilling during flood events, extreme wet weather or operator error. As there is no spillway, the effect would be similar to a tailings breach due to erosion of the perimeter embankment. The severity level would be considered 'MAJOR' due to Business Importance.

Hence, based on a PAR of <1 and the 'MAJOR' Dam Failure Severity Level, the facility would be rated as **SIGNIFICANT** for the Dam Failure Consequence Category. Based on a PAR of <1 and the 'MAJOR' Environmental Spill Severity Level, the facility would be rated as **SIGNIFICANT** for the Dam Spill Consequence Category.

3.5 DESIGN CATEGORY AND DESIGN CRITERIA

On the basis of the assessment provided above in sections 3.3 and 3.4, the overall minimum design criteria for this study are summarised in Table 3.4. These design criteria are in accordance with the DMIRS Code of Practice (Ref. 4) and ANCOLD guidelines (Ref. 2) where criteria are given. Other parameters were adopted on the basis of industry best practice and client requirements.



Table 3.4: DMIRS and ANCOLD design criteria summary

	HYDRAULICS				
Design Storm Storage Allowance	1% AEP, 72 hour duration storm with no release, evaporation or decant.				
(Appendix 5, DMIRS)					
Freeboard (Appendix 5, DMIRS)	Total Freeboard of at least 0.5 m or wave plus 0.3 m; Operational Freeboard to be the greater of 0.3 m or the predicted crest deformation due to the maximum seismic event.				
Operating Spillway	None.				
Closure Spillway	None.				
Diversion structures during operation	2% AEP storm				
STABILITY					
Earthquake loading (Table 7 ANCOLD 2019)	OBE 475 Year SEE 1,000 Year Post Closure MCE				
Minimum Factor of Safety	Long term drained 1.5				
(Table 8 ANCOLD 2019)	Short term undrained 1.3				
	Post Seismic 1.1				
DAM SA	DAM SAFETY / INSPECTION FREQUENCY				
Inspection Frequency	 Comprehensive inspection by Dam Designer or equivalent qualified Engineer after first year and then every 2 years. Annual intermediate inspections by Dam Engineer. Routine daily by operations personnel. 				

The above parameters have been used for the design of the facility.

The TSF has been designed to manage direct rainfall and runoff from a total catchment of approximately 140 Ha partially made up of direct rainfall on the tailings and supernatant pond surfaces. The storm capacity freeboard in the facility is maintained significantly greater than the minimum requirement of a 1% AEP 72-hour event.



4. GEOTECHNICAL SITE INVESTIGATION

4.1 GEOTECHNICAL INVESTIGATION 2017

A geotechnical investigation was undertaken in July 2017 over the TSF expansion footprint in the vicinity of cell A and B to determine the sub-surface conditions, comprising the following scope:

- Drilling of two boreholes using diamond coring techniques including in situ SPT and permeability testing;
- Excavation of 20 test pits;
- Collecting 9 No. push tube samples;
- Conducting 14 No. DCP tests; and
- Laboratory testing of samples obtained from the boreholes and the test pits.

Detailed findings are presented in the TSF expansion permitting design (Ref. 7). The following conclusions were made:

- The areas investigated are partly disturbed and relatively flat;
- The near surface profile can be excavated with conventional earthmoving equipment. The sub-surface profile generally comprises:
 - Backfill, comprising Sandy Clay/Sandy Silt/Clayey Sand if present, overlying;
 - Sandy Clay/Clayey Sand; and
 - Silty Clay/Clayey Silt.
- The in situ sandy Clay within the basin areas are generally suitable for use as soil liner if compacted to 98% SMDD. Clayey Sand with less than 35% fines may be present at some places locally within the basin.

4.2 CPT INVESTIGATION 2020

A second investigation was undertaken in September 2020 to evaluate in situ tailings material properties and hydraulic conditions within the stored tailings. The geotechnical investigation comprised the following scope:

- Eight cone penetrometer tests with dissipation tests to assess excess pore pressures within the existing TSF cells A and B;
- Collection of undisturbed samples of soils and tailings material for subsequent laboratory testing including the following:
- Particle size distribution;
 - Atterberg limits;
 - Linear shrinkage;



- Compaction tests;
- Falling head permeability tests;
- Emerson tests;
- Consolidated Undrained (CU) Triaxial tests; and
- Uniaxial compressive strength.

Detailed findings are presented in the KP technical memorandum TSF expansion geotechnical investigation included in Appendix B and the CPT investigation interpretation results are summarised below:

- The CPT investigation provided indicative depths and thicknesses of the tailings material layers. The top 3 m to 6 m comprises dilative tailings overlying contractive tailings. Tailings are generally becoming weaker with depth. The lower part of the tailings is sensitive with relatively low residual strength;
- The tailings close to the embankments are generally dilative, possibly due to traffic compaction during construction;
- The friction angle for dilative materials range from 32° to 45°, with TSF Cell A
 having a slightly higher friction angle than TSF Cell B;
- The undrained shear strengths vary significantly with state parameter, a few peak
 undrained shear strength ratios appear to be high, possibly influenced by thin
 sand lenses in the clay like layers. The undrained shear strengths need to be
 capped at the drained shear strength for stability assessment; and
- The tailings have low permeability and as such will tend to hold moisture in the tailings for a significant amount of time prior to drainage, assuming natural drainage without the aid of any underdrainage system.

The following geotechnical design parameters for in situ tailings were determined and other undrained and post seismic parameters are discussed in the technical memorandum included in Appendix B.

Table 4.1: Undrained shear strengths for tailings

CPT Reference	Elevation from	Elevation to	Peak Undrained	Residual Undrained
	(m DI)	(m DL)	strength	strength
	(m RL)	(m RL)	s _u /σ _v '	su/σ _v '
	503.30	499.40	Dilative	
CPT 05-1	499.40	495.82	0.52	0.12
	495.82	489.12	0.33	0.04
	504.20	500.80	Dilative	
CPT 06-1	500.80	497.96	0.62	0.14
	497.96	490.56	0.40	0.06



4.3 CELL C INVESTIGATION 2020

A third geotechnical investigation was undertaken in October 2020 over the TSF expansion footprint Cell C (smaller footprint) to determine the sub-surface conditions, comprising the following scope:

- Drilling of two boreholes using diamond coring techniques including in situ SPT and permeability testing;
- Excavation of 34 test pits;
- Collecting 2 No. push tube samples;
- Conducting 34 No. DCP tests; and
- Laboratory testing of samples obtained from the boreholes and the test pits.

Detailed findings are presented in the TSF expansion permitting design (Ref. 7). The following conclusions were made:

- The areas investigated are partly disturbed and relatively flat. The site should generally be suitable for the proposed TSF expansion;
- Conventional earthmoving equipment, such as large dozers and excavators should be capable of excavating the near surface materials across the site.
 However, excavation difficulties will likely be encountered where Ferricrete hardpan exists at shallow depth. It is expected that the Ferricrete hardpan materials should be excavatable using a D9 dozer or equivalent equipped with a single-tyne ripper attachment;
- Due to the existence of Ferricrete hardpan at shallow depth, the cut-off trench will be likely very shallow, less than 1 m, typically 0.5 to 1 m depending on locations;
- The ground water level is approximately 30 m below the existing ground surface, it will unlikely have any impact on the TSF design and construction;
- The permeability of top layer materials based on lab tests appear to be higher
 than would normally be expected for similar materials compacted to 98%SMDD.
 However, the in situ test results appear to be reasonable and indicate that the in
 situ undisturbed materials have lower permeability than the re-compacted same
 materials after excavation or remoulding as the cementation bond between
 particles is likely to be destroyed after excavation and compaction;
- The Emersion tests indicate that the upper layer materials and mine waste are not dispersive; and
- The in situ near surface soils to be compacted as low permeability soil liner, achieved a permeability range of 5 x 10⁻⁶ m/s.



4.4 CELL C/D INVESTIGATION 2021

Due to the increasing throughput, the TSF original concept Cell C was expanded to occupy a larger footprint. A fourth investigation was conducted in March 2021, comprising the following scope:

- Drilling three boreholes within the proposed TSF Cells C and D area using diamond coring techniques to depths of 24.1m, 27.3 m and 9m respectively;
- Standard Penetration Tests (SPT);
- Fifty-one test pits within the proposed TSF Cells C and D area; and
- Collection of soil samples for laboratory testing.

Fieldwork was supervised by a Geotechnical Engineer from KP Perth. The site investigation was undertaken in accordance with Australian Standard AS1726–2017. Detailed findings are presented in the TSF expansion Site Investigation 2021 provided in Appendix C. The following conclusions were made:

- The areas investigated are partly disturbed and relatively flat. The site should be generally suitable for the proposed TSF expansion;
- Conventional earthmoving equipment, such as large dozers and excavators should be capable of excavating the near surface materials across the site. However, excavation difficulties will likely be encountered where Laterite/Ferricrete hardpan exists at shallow depth. It is expected that the Ferricrete hardpan materials should be excavatable using a D9 dozer or equivalent equipped with a single-tyne ripper attachment;
- Due to the existence of Laterite/Ferricrete hardpan at shallow depth, the cut-off trench will be likely shallow, typically about 1.0 m, which will vary with locations;
- The ground water level is approximately 30 m below the existing ground surface, it will unlikely have any impact on the TSF design and construction;
- The permeability of top layer materials based on lab tests appear to be higher than what would normally be expected for similar materials compacted to 98% SMDD. However, the in situ test results appear to be reasonably low and indicate that the in situ undisturbed materials have lower permeability than the re-compacted same materials after excavation or remoulding as the cementation bond between particles is likely to be destroyed after excavation and compaction. The materials in the cemented laterite layer and the underlying residual soil (extremely weathered rock) will have reasonably low permeability;
- The Emersion tests indicate that the upper layer materials are not dispersive;



- The in situ near surface soils in some areas are not suitable to be compacted as low permeability soil liner; and
- The tested mine waste samples may not be suitable to be used as low permeability zone (Zone A) due to the high gravel content and low fine contents.
 Additional samples shall be collected from the stockpiles for lab testing to identify and quantify suitable low permeability mine waste as Zone A materials.



5. TAILINGS CHARACTERISTICS

5.1 GENERAL

Tailings physical testing and geochemical analysis was conducted on tailings samples in 2017 (Appendix D) during the initial life of mine (LoM) expansion design. An operational tailings sample was provided in 2018, with testing undertaken to verify material properties for Stage 7 to Stage 9 designs (Ref.11). The following physical tests were carried out on the samples:

- Classification tests to determine:
 - Particle size distribution of the tailings.
 - Supernatant liquor density.
 - Liquid and plastic limits of the tailings solids.
 - Tailings solids particle density.
- ii. Undrained and drained sedimentation tests.
- iii. Air drying tests.
- iv. Permeability tests.
- v. Consolidation tests.
- vi. Geochemistry tests.
- vii. Acid generation potential.
- viii. Multi element analysis.
- ix. Supernatant water quality.

Full details of the tests undertaken, the results and interpretation are provided in Appendix D. A summary of the design parameters and associated implications for design are provided in the following sections.

5.2 DESIGN PARAMETERS

The deposition design uses expected tailings material types and properties based on laboratory testing of the bench scale samples in 2017 for this expansion design. It is envisaged that the material properties will be similar but will vary daily and over the life of the project. Material properties which may affect the deposition methodology include particle size distribution, percent solids of discharge, slurry flow rate and specific gravity of solids and liquids. The tailings design parameters are summarised in Table 5.1.



Table 5.1: Tailings material properties

Parameter	Value
Maximum Air-Dried Density	1.62 t/m ³
Solids Particle Density	2.77 t/m ³
P ₈₀	72 μm
Beach Slope	1V: 100H (latest beach survey)
Solids Content	62%
Supernatant Release	28 – 35%
Underdrainage Release	3-8%
Vertical Permeability	4x10 ⁻⁷ m/s
Coeff. of Consolidation (Cv)	73 m²/yr
Coeff. of Volume Decrease (Mv)	0.014m ² /kN
Compression Index (Cc)	0.226

5.3 WATER BALANCE MODELLING PARAMETERS

5.3.1 General

The water management model requires a number of input parameters. The following sub-sections outline the selection of parameters used for the water management modelling.

A climatology assessment was completed as part of the original permitting design in the KP technical memorandum basic climatology included in Appendix A.

5.3.2 Rainfall and Evaporation

Daily historic climate data were obtained from the Scientific Information for Land Owners (SILO) database for use in deriving the basic climatology.



Table 5.2:	Annual s	vnthetic	climate	scenarios
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Month	Annual Climate Scenarios (mm)				
	1% AEP Dry	Average	1% AEP Wet	Pan Evaporation	
October	0.0	27.8	29.9	285	
November	2.4	0.7	3.9	333	
December	0.0	2.0	69.1	389	
January	4.0	0.0	101.2	404	
February	7.3	19.2	65.8	321	
March	0.0	17.5	173.3	289	
April	2.4	57.7	2.4	193	
May	46.2	29.2	45.4	128	
June	0.0	20.7	64.0	91	
July	0.0	13.9	8.2	98	
August	0.0	35.1	8.1	136	
September	0.0	0.6	0.0	199	
Total	62.3	224.4	571.3	2,866	

Design storms for the 1% Annual Exceedance Probability (AEP) 72-hour event is 201 mm and the Probable Maximum Precipitation (PMP) 24-hour event is 560 mm.

5.3.3 Runoff Coefficients

Runoff coefficients for undisturbed areas, active and drying tailings beaches, and pond surfaces were based on expected values for similar projects in the region. The runoff coefficients used for water balance modelling are presented in Table 6.2.

Table 5.3: Adopted runoff coefficients

Condition	Runoff Coefficient
Undisturbed Area	0.2
Drying Tailings Beach	0.8
Active Tailings Beach (Supernatant Producing Areas)	1.0
Ponds	1.0

5.3.4 Additional Modelling Parameters

The tailings slurry design parameters are provided in Section 5. The additional TSF design parameters are summarised in Table 6.3.



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TSF design	Parameters
Storage Capacity	42 Mt (From August 2017)
Life of Mine	10 years (FY2021 to FY2031)
Ore Throughput	3.0 – 6.0 Mtpa
Design Beach slope	1V: 100H (latest beach survey dated 2 nd September 2020)

5.4 IMPLICATIONS FOR DESIGN

Tailings geochemistry testing of a representative sample of tailings solids and supernatant water for the Thunderbox operation was completed in 2017 and 2018. The implications for tailings management are presented in the KP technical memorandum-Tailings Geochemical Assessment (Ref. 17). Results of the physical and geochemical testing of samples of tailings indicate the following:

- The tailings are Non-Acid Forming (NAF);
- The multi-element analyses of the solids indicate several metals to be present at
 elevated concentrations (namely arsenic, antimony, chromium, lead, sulfur and
 zinc). Arsenic and Sulfur were found to be highly enriched. As such, a cover
 system will be required on closure to isolate the tailings from the environment;
- The results of the supernatant testing have been compared to reference water quality standards for release of water from mining operations and livestock drinking water. The results indicate that the supernatant exceeds a number of reference guidelines. The laboratory was unable to determine the total cyanide concentration due to suspected interreference with thiocyanate in the sample. The concentrations of both free and WAD cyanide were high. Although the WAD cyanide level does not meet the release criteria, it is lower than the 50 mg/L target required under the International Cyanide Guideline for decant ponds. Measures are incorporated into the design to reduce seepage from the basin and release of supernatant to the environment. Monitoring bore data for the existing cells A and B show that no significant elevations of cyanide or arsenic have been registered and are well below the DWER licence conditions;
 - The supernatant and seepage are saline due to the use of saline groundwater in the processing of ore. The groundwater in the vicinity of the TSF is stock water quality. As such it is not expected that seepage from the facility will result in a significant increase in groundwater salinity. Monitoring data for the existing cells A and B indicate that Total Dissolved Solids are well below the DWER licence limit



- of 1,500 mg/L, fluctuating within a range of 290 mg/L to 750 mg/L during the most recent monitoring period;
- A cover system will need to be constructed on closure to reduce any dust generation and be resistant to erosion to prevent loss of tailings solids, also water ingress and resulting seepage. A modified store and release cover system is proposed over the tailings beach on closure to isolate the tailings solids from the environment. This cover system would limit infiltration of rainfall via a lower permeability layer and promote revegetation through provision of a water storage zone. A capillary break would also prevent the upward migration of salts; and
- In addition, the design should ensure the TSF has sufficient stormwater capacity.



6. WATER MANAGEMENT MODEL

6.1 GENERAL

The management of water within the TSF is a critical aspect of the design. In order to understand (and enable control) the flow of water around the site, a TSF water balance model was developed. The model uses the design tailings production together with estimated settled tailings densities to determine the tailings level at various stages in the facility life. The model then examined a range of extreme rainfall events to determine supernatant pond volumes and the required embankment stage crest levels. A range of extreme dry rainfall events was also analysed to determine the water shortfall that could potentially occur.

The primary objectives of the water balance modelling are summarised below:

- Estimate settled tailings densities;
- Establish the filling rate for tailings solids within the TSF;
- Determine net flows and average supernatant pond volumes within the TSF throughout operation;
- Determine supernatant pond volumes following design rainfall events and sequences;
- Determine staged embankment crest elevations, to ensure containment of tailings and prevent discharge of supernatant water for design rainfall events and sequences; and
- Determine the magnitude of recycle water shortfalls during design dry rainfall sequences.

For the water management model, the following inputs into the TSF were included:

- Water in the tailings slurry from the plant; and
- Rainfall runoff from the tailings and pond surface.

6.2 TSF WATER BALANCE

6.2.1 Model

The TSF water balance was modelled using specially developed computer software. The program is a computer model written in Visual Basic/Excel specifically for tailings storage facilities and incorporates a database of information derived from both laboratory and field data accumulated over the past 30 years by KP Australia. The program calculates tailings densities achieved in the storage, and determines the volume of water available for return to the process plant taking into account rainfall, evaporation, supernatant and underdrainage release from the tailings due to consolidation.



6.2.2 Modelling runs

The model was run with a repeating sequence of average conditions. In addition, the effects of 1% AEP wet and dry years were assessed. The effects of storm events on the TSF were also examined, as follows:

- Average climatic conditions;
- 1% AEP, 1 year wet sequence occurring in every year of the operation;
- Average climatic conditions with a 1% AEP 72 hour storm event on the TSF occurring in every year of the operation;
- Average climatic conditions with a Probable Maximum Precipitation (PMP)
 24 hour storm event on the TSF occurring in every year of the operation;
- Average climatic conditions with a Probable Maximum Precipitation (PMP)
 72 hour storm event on the TSF occurring in every year of the operation; and
- 1% AEP, 1 year dry sequence occurring in every year of the operation.

It should be noted that for all modelling runs the data were checked to ensure that no overlap of extreme events occurred. In these cases, the timing was adjusted to ensure each event considered was independent of any other event.

6.2.3 Results of modelling runs

The model was run on a monthly time-step for the duration of the operating life. Modelled flows do not represent the design duties for pumps and pipelines or peak flows for rainfall as they are averaged over the month and do not take into account efficiency and availability of the infrastructure. Six different climate scenarios were modelled, as described in the sections that follow and the results are presented in tables 6.4 to 6.9.

6.2.3.1 Average climatic conditions

The model was run with a repeating sequence of average conditions. The size of the supernatant pond is shown in Figure 6.1. Based on the modelling the following conclusions can be made:

- The TSF operates with a water deficit under average conditions. The supernatant pond remains at minimum pond size (specified in the modelling as 5,000 m³). The make-up water required is approximately 38% 83% of the water in the slurry, which ranges from 30,000 to 131,000 m³/month;
- The TSF recycle to the process plant varies from 17% to 62% of water in slurry during the operation. The average recycle is 35% of the water in slurry for the two cells, which ranges from 13,000 to 83,000 m³/month;



- The average settled dry density gradually increases as a result of exposed dry tailings beach and consolidation of the underlying tailings, achieving a final average density of approximately 1.60 t/m³; and
- Over the life of the project, the water recycled from the TSF is not sufficient to supply the water required to satisfy the process plant outputs of tailings slurry water. The total make-up water required over the life of operation is approximately 14.5 Mm³ with 7.3 Mm³ for Cell C and 8.1 Mm³ for Cell D.

6.2.3.2 1% AEP wet sequence

Effects of a 1% AEP wet year were analysed by inserting a wet year independently into each year of the model. As the pond reduces to the minimum volume during the dry season each year, the water balance impact is independent of the previous year's rainfall.

Given the prioritisation of water supply to process plant make-up, a wet year increased the TSF recycle average to as high as 85% of the water in slurry, consequently reducing the make-up requirement, which ranges from 19,000 to 127,000 m³/month.

The size of the supernatant pond is shown in Figure 6.1.

6.2.3.3 1% AEP dry sequence

The model was also run for the 1% AEP dry year simulations, to determine the process water shortfall that could potentially occur. The tables summarise the results of multiple individual modelling runs for a single 1% AEP dry event. Each year is independent, as the pond level stays at its minimum, allowing multiple individual modelling runs for the dry event to be carried out without impacting on one another.

The 1% AEP dry year precipitation is 62.3 mm. Given that the supernatant pond stays at the minimum even under average climatic conditions, the modelled dry sequence results in even lower TSF recycle.

The average recycle under 1% AEP dry year conditions is 29% of the water in slurry, which yields 16.4 Mm³ shortfall in total, ranging from 32,000 to 132,000 m³/month. It was assumed that the additional water required can be sourced from groundwater bores.

6.3 DENSITY

Tailings physical characteristics from laboratory testing in Section 5.1 were adopted to assess the impact of the TSF size on the settled dry density. It should be noted that the tailings density increases as the effective deposition area increases thus reducing the embankment rate of rise.



The deposition plan was set up to deposit into the lower cell initially for a longer period of time, so that the rate of rise in the lower TSF Cell is higher. Over time, the tailings level between the two cells will become similar, which reduces the construction materials required to raise the divider embankment. Ultimately, the final TSF embankments will be the same with tailings at approximately the same elevation.

A plot of tailings density throughout the facility life is shown in Figure 6.2. The individual layer densities are controlled in part by the minimum pond size (assumed to be 5,000 m³) and dry densities are in the range 1.49 t/m³ to 1.60 t/m³.

Table 6.1: Tailings density comparison

Nominal	Cell A		Cell B	
Throughput to TSF (Mtpa)	Layer Density Range (t/m³)	Overall Density of Tailings Mass (t/m³)	Layer Density Range (t/m³)	Overall Density of Tailings Mass (t/m³)
3.0 - 6.0	1.49 – 1.61	1.60	1.48 – 1.61	1.60

The modelled layer density reaches the maximum theoretical air-drying value in the summer months, which indicates that the facility area for TSF expansion is considered to be sufficient for a production rate in the range of 3.0-6.0 Mtpa, for the particular tailings characteristics (i.e. Hard Rock Tailings). Tailings densities are expected to improve due to the larger effective deposition areas and better supernatant pond control.

6.4 SUMMARY

The following conclusions can be made from the TSF water balance modelling:

- The water balance for the expansion is negative during the life of mine as expected. The supernatant pond stays at minimum size almost all the time, ponding of water against the embankment wall is unlikely to occur under any storm event provided the facility ponds are maintained as small as practical;
- Plant site make-up water is required throughout the operation but slightly varies between the wet and dry seasons. Due to the low rainfall the benefit of storing the wet season runoff, prior to expansion commissioning, is not significant;
- Under average climatic conditions, available water from the TSF is less than the
 required plant make-up. Decant returns range from 13,000 to 83,000 m3/month
 and the make-up requirement ranges from 30,000 to 131,000 m3/month.
 Consequently, the minimum operating pond volume is maintained throughout the
 year;



- Under dry climatic conditions, even less make-up water is available from the TSF.
 Decant returns range from 14,000 to 79,000 m³/month. This results in an increase in the external water requirement, ranging from 32,000 to 132,000 m³/month;
- Under wet climatic conditions, water available from the TSF increases and results in a temporary increase in TSF recycle to as high as 144,000 m3/month; and
- The TSF expansion footprint is considered to be sufficient for a tailings production rate of 6.0 Mtpa.



7. TAILINGS STORAGE FACILITY DESIGN

7.1 INTRODUCTION

The existing tailings storage facility (TSF) comprises of two cells, designated Cell A and Cell B, which were designed as a paddock type facility with basin underdrainage and a central pump out decant system. The TSF Cell A and Cell B Stage 1 were constructed in October 2002, utilising mine waste. Subsequent raises were constructed utilising the upstream construction technique with compacted tailings at 2.5 m per lift until April 2018. A TSF expansion integrating into the existing Eastern waste dump was designed and permitted in 2018 and the Stage 6 expansion construction was completed in December 2018, forming the north void and south void by construction of additional embankments connecting the waste dump and the existing embankment. Initial tailings deposition filled the voids. The south void deposition was completed in Q1 2019 and the north void deposition is expected to be completed in Q3 2021.

The development of tailings storage at Thunderbox is summarised in Table 7.1.

Table 7.1: TSF construction summary

Stage	C	Cell A	Cell B	
	Crest RL	Construction	Crest RL	Construction
		completed		completed
1	RL497.0 m	October 2002	RL493.5 m	October 2002
2	RL499.5 m	July 2003	RL496.0 m	September 2003
3	RL502.0 m	October 2004	RL498.5 m	October 2005
4	RL504.5 m	November 2006	RL501.0 m	March 2007
5	RL507.0 m	March 2016	RL503.5 m	April 2017
6	RL509.7 m	April 2018		
6 Expansion	RL503.5 m North Void	December 2018	RL503.5 m South Void	January 2019
7 Expansion	RL506.0 m North Void	May 2019	RL506.0 m	May 2019
8 Expansion	RL508.8 m North Void	March 2020	RL508.8 m	August 2020
9 Expansion	RL511.4 m	January 2021	RL511.4 m	Ongoing



The existing operation discharges tailings into TSF Cell B and this is expected to reach full capacity in the next several months at which time deposition will re-commence in Cell A (completed in June 2021).

With the revised mine plan to increase plant throughput, the TSF horizontal expansion (Cell C and Cell D) will develop additional area in the east, integrating the Eastern Waste Dump and the existing TSF to form two cells and two central decants. The final combined storage (Cell A, Cell B, Cell C and Cell D) will provide approximately 267 Ha in final tailings inundation area which will operate for the next 10 years. Tailings will be discharged from the perimeter embankment into the facility by sub-aerial deposition methods, via spigots spaced at regular intervals. The active tailings beach will be regularly rotated so as to maximise tailings density and evaporation of water.

The general layout of the TSF is shown on Drg. No. 801-296-D3000-110 and discussed in the following sections.

7.2 EMBANKMENT CONSTRUCTION

The Cell C and Cell D embankment will form a paddock-type storage and existing Cell A and Cell B will be raised in subsequent stages to allow construction. The staged embankment construction schedule for cells C and D only is given in Table 7.2. Over the 10-year period the rate of rise for each cell is variable but averages in the range 2.5 metres per year.

 Table 7.2:
 TSF expansion staged embankment

Cell C and D	Storage Capacity			Embankment Levels		
Stage	Cell C	Cell D	Total	Cell C	Divider	Cell D
	(Mt)	(Mt)	(Mt)	(RL m)	(RL m)	(RL m)
1	6.7	2.8	9.5	494.1	494.1	494.1
2	11.5	7.4	19	498.2	498.2	498.2
3	14	9.9	23.9	500.2	500.2	500.2
4	17	12.7	29.7	502.6	502.6	502.6
5	20	15.7	35.7	505.0	505.0	505.0
6	21	16.7	37.7	505.8	505.8	505.8

The TSF embankment will be constructed with an upstream slope of 1V:2.0H, a downstream slope of 1V:2.75H (interim) and a crest width of 6 m. The initial two stages of embankment will be raised using downstream construction technique with Run of Mine (ROM) waste from the current pit cutback and upstream construction technique using tailings in subsequent stages. The divider embankment will be raised centrally with Run of Mine (ROM) waste from the current pit cutback until Stage 2 after which the divider



will be raised with compacted dry tailings sourced from the exposed tailings beach. The embankments against the Eastern Waste Dump (of the existing Cell A and Cell B) will be raised using downstream technique with Run of Mine (ROM) in Stage 1 and compacted dry tailings from Stage 2 onwards. The typical embankment sections are shown on Drg. No. 801-296-D3000-301 and Drg. No. 801-296-D3000-302.

A waste rock buttress downstream of the external embankment of Cell A and Cell B is currently proposed to increase factors of safety. The detailed buttress requirement will be assessed in the detailed design phase for each raise based on tailings strength developed.

The TSF divider embankment will be constructed utilising Run of Mine (ROM) waste in Stage 1 and compacted dry tailings with a slope of 1V:2H for both upstream and downstream slopes. The typical embankment sections are shown on Drg. No. 801-296-D300-301.

The TSF embankment buttressed against the Eastern Waste dump will consist of 6 m compacted low permeability zone (selected mine waste in Stage 1 and tailings in subsequent stages) and will be raised using the downstream construction technique. The typical embankment sections are shown on Drg. No. 801-296-D3000-302.

The embankment against the Cell A and Cell B buttress will consist of a compacted low permeability zone (selected mine waste in Stage 1 and tailings in subsequent stages) and will be raised using the downstream construction technique. The typical embankment sections are shown on Drg. No. 801-296-D3000-303.

Tailings will be used as Zone A materials for all the subsequent upstream raises, which will be won from the previous stage tailings beach. This material will be scarified, moisture conditioned and compacted by an earthworks contractor.

7.3 CELL A AND CELL B BUTTRESS

A buttress downstream of cells A and B is currently undergoing construction in order to increase the embankment factor of safety on the east side. This buttress is being constructed utilising oxide mine waste material, won directly from the current pit cut back, placed and traffic compacted by the mining fleet. The current design intent is for Cells C and D to further buttress Cell A and Cell B during operation as the tailings level within the facility rises. It is noted that the buttressing has been taken into account in when determining the capacity of cells C and D.



7.4 CUT-OFF TRENCH AND LOW PERMEABILITY UPSTREAM MATERIAL

A cut-off trench is proposed around the entire facility including the east waste dump embankment, Cell A and Cell B embankment and the external embankment. It is proposed that low permeability Zone A material will be placed on the east waste dump, Cell A and Cell B downstream buttress face in addition to the external embankment to limit seepage from the facility.

7.5 GEOTEXTILE/FILTER ZONE

Based on available laboratory testing completed on samples taken from oxide mine waste and the eastern waste dump and no geotextile/filter zone will be required. It is noted that variations within the material may be encountered during construction and therefore a geotextile/filter zone may be required where materials are not considered compatible. This will be assessed in the field during construction.

7.6 SEEPAGE CONTROL

In order to reduce seepage losses through the TSF expansion basin and to increase the settled densities of deposited tailings, a number of seepage control and underdrainage collection features are integrated into the design. Monitoring bore data from the existing cells A and B have been reviewed to confirm the performance of the seepage control measures implemented. It is noted that groundwater quality with respect to pH, water level, total dissolved solids, WAD CN and Arsenic are all within the DWER licence requirements (Ref. 16). Therefore, the seepage control measures adopted for cells C and D have been based on the those utilised within cells A and B. A summary of the Monitoring Borehole water quality data from the 2018 Audit is provided below in Table 7.3.

Table 7.3: Summary of Monitoring Borehole water quality data - 2018

Parameter	DWER licence	Gro	undwater	
	limit	Max	Min	Average
рН	6.0 - 9.0	7.9	7.5	7.7
Standing water level	> 4.0	24.6	12.8	16.6
Total dissolved solids	< 1,500 mg/L	750	290	521
Weak acid dissociable cyanide (WAD CN)	< 0.5 mg/L	0.096	0.005	0.014
Arsenic (As)	< 0.5 mg/L	0.063	0.001	0.017



The seepage control and underdrainage collection systems will consist of the components as listed below:

- i. Low permeability basin liner;
- ii. Basin underdrainage collection system;
- iii. Underdrainage collection tower; and
- iv. Embankment upstream toe drain.
- v. Cut-off trench

Primary seepage control from the expanded TSF footprint Cell C and Cell D will comprise the construction of a cut-off trench excavated in the foundation soils, backfilled with low permeability fill to reduce seepage loss through the embankment foundations.

The cut-off trench will be located beneath the upstream toe of the embankments and will be cut to a depth of approximately 0.5-1 m. The cut-off trench will be constructed continuously along the upstream toe of the external embankment and the embankment section against the waste dump to the full deposition elevation to limit potential seepage at any level. If the cut material is suitable as fill, it may be replaced in the excavation in compacted layers; alternatively, suitable low permeability material will be won, conditioned, placed and compacted in the trench.

The details of the embankment cut-off trench are shown on Drg. No. 801-296-D3000-302.

i. Low permeability compacted soil liner (CSL)

A compacted soil liner (CSL) will be constructed within the entire expanded TSF basin area, comprising either reworked in situ material or imported Zone A material.

The construction methodology for the compacted soil liner will be as follows:

- The TSF basin area will be cleared, grubbed and topsoil stripped;
- Borrow areas within the TSF basin will be shaped to facilitate full gravity drainage into the TSF;
- Where in situ material is suitable (classified as Zone A material) scarify, moisture condition and compact in situ material to achieve target density; and
- Where in situ material is unsuitable (too coarse for classification as Zone A
 material or low plasticity to non-plastic) win suitable material from borrow, place,
 spread, moisture condition and compact to achieve target density.



The CSL surface shall drain positively and be finished with a smooth drum vibratory roller or other approved means. The low permeability liner properties (permeability, moisture content, dry density ratio and thickness) will be preserved after compaction. A trial pad will be constructed to assess the linear shrinkage during drying. If significant cracking of the CSL is observed then a protection layer will be constructed over the liner, to prevent cracking and erosion during operation prior to inundation of the CSL with tailings.

Monitoring bore data have been reviewed to confirm the performance of the seepage control measures implemented for existing cells A and B. It is noted that the existing cells A and B comprise a CSL located 300m radially around the decant towers with the remaining basin untreated. Additionally, no central underdrainage was installed within cells A or B. Groundwater quality with respect to pH, water level, total dissolved solids, WAD CN and Arsenic for cells A and B are all within the DWER licence requirements (Ref. 16). Therefore, it is proposed that reworked in situ material and imported material (where required) will be utilised to construct the CSL in cells C and D. Laboratory testing on the basin material indicated that a permeability in the range of 6 x 10⁻⁷ m/s would typically be achievable. Areas of unsuitable material will be removed and replaced with low permeability material won from borrow. It is noted that the Ferricrete / Laterite hardpan underlying the facility will work in conjunction with the CSL and other seepage control measures to provide adequate seepage control for the facility. The seepage control measures proposed for cells C and D generally exceed those utilised for cells A and B.

ii. Basin underdrainage collection system

The TSF underdrainage collection system is designed to reduce the phreatic surface on the tailings basin area under the decant pond. The underdrainage has several benefits:

- Reduces seepage through the basin which is beneficial to the environment; and
- Drains the tailings mass, thus increasing the density of the tailings and provides a more efficient facility in terms of storage.

The design of the underdrainage system takes advantage of the natural fall of the ground, and thus minimal reshaping of the basin will be required. The underdrainage system will be constructed over the CSL under the decant pond to reduce seepage further from the supernatant pond.

The underdrainage system will consist of finger and collector drains in both Cell C and Cell D. The finger drains will consist of 100 mm diameter drain coil pipe embedded within a 400 mm sand (zone F) layer wrapped in geotextile. The collector drains will consist of



160 mm diameter drain coil pipe embedded within a 400 mm sand (zone F) layer wrapped in geotextile. Erosion mitigation measures have been incorporated into the design to protect the underdrainage system for the period between end of construction and tailings deposition.

The branch drain pipes will feed directly into the underdrainage tower located at the upstream toe of the embankment. The extent of the underdrainage system is designed to cover the maximum pond volume resulting from a 1 in 50-year 24-hour storm event. The underdrainage system will be constructed in Stage 1 and will cover approximately 35% of the total basin area.

iii. Underdrainage tower

On the upstream face of the embankment, two underdrainage towers will be constructed. Each tower will collect solution from the toe drains and underdrainage system. The underdrainage tower will consist of the following components:

- A solid concrete pipe of 900 mm internal diameter, Class 3, reinforced concrete with spigot and socket joints;
- A submersible pump (designed by others); and
- A hoist and pulley to raise and lower the pump (designed by others).

The typical sections and details are shown on Drg. No. 801-296-D3000-452.

iv. Embankment upstream toe drain

In addition to the expanded basin drainage system, a toe drain will be constructed along the upstream toe of the embankment on sections where the embankment is designed to be raised upstream. The toe drain has two purposes. The main purpose is to increase the stability of the embankment during the initial and future stages by providing drainage of tailings at the embankment. The second purpose of the toe drain is to act as an underdrainage collection pipe.

The toe drain will comprise of two 160 mm drain coil pipes laid at the base of the external embankment toe within 300 mm of drainage material (Zone F) wrapped in geotextile. The toe drain will flow into the underdrainage collection tower for recycling back into the facility. Erosion mitigation measures have been incorporated into the design to protect the underdrainage system for the period between end of construction and tailings deposition.

Details of the embankment toe drain are shown in Drg. No. 801-296-D300-302.



All seepage control works will be completed as part of the Stage 1 construction. The underdrainage tower will be extended during future raises.

v. Cut-off Trench

One of the primary seepage controls from the TSF will comprise a cut-off trench excavated in the foundation soils and backfilled with low permeability fill (Zone A) to reduce seepage loss under the embankment foundations. The cut-off trench will be located beneath the upstream low permeability zone to a depth of 1 to 2 m (depending on ground conditions). The cut-off trench will be constructed continuously along the embankment to the full deposition elevation in order to reduce potential seepage at any level. Low permeability material (Zone A) will be used to backfill the cut-off trench. If the cut material is suitable as fill, it may be placed in the TSF embankment in compacted layers (or reused as cut-off trench backfill).

The location and details of the embankment cut-off trench are shown on Drg. Nos 801-296-D3000-201 to 801-296-D3000-303.

7.7 DECANT AND RETURN WATER POND SYSTEM

The deposition is to push the supernatant pond towards the centre of the TSF expansion Cell C and Cell D. The decant pump will be raised on a regular basis with successive embankment lifts to ensure that no tailings enters the pump intake. The decant system will comprise of the following components:

- An access causeway constructed of Zone D material;
- A decant tower, consisting of an 1,800 mm diameter slotted concrete pipe surrounded by clean waste rock (Zone G);
- A submersible pump and pipework (designed by others); and
- A hoist and pulley to raise and lower the pump (designed by others).

The location of the decant tower is shown on Drg. No. 801-296-D3000-500 with relevant sections and details on Drg. No. 801-296-D3000-551 and Drg. No. 801-296-D3000-552.

7.8 SEEPAGE MODELLING

Finite element seepage analyses, using GeoStudio (Ref. 12) Seep/W program, were conducted to evaluate seepage through the embankment and foundation under normal operating conditions and to approximate the phreatic surface and pore water pressures in the TSF embankment.

The seepage analysis results were used to estimate the total seepage losses from the TSF. The outputs from the Seep/W program were used to model the phreatic surface



and pore water pressures for the final embankment configurations in a Slope/W slope stability assessment. A detailed seepage assessment was completed and is presented in Appendix E. The results summary is provided below:

- In the steady state condition, the effectiveness of the basin and toe drains appear not to have significant impact on the phreatic levels and seepage to the embankment due to the low ground water table and pond being far away from the embankment. However, these drains will be effective to lower the phreatic levels in the tailings during operation; and
- Based on the seepage modelling for the steady state condition, the estimated phreatic surfaces will be under the embankment, and there will be very little seepage into the embankment. During operation it is recommended to carry out the following further work:
 - Transient modelling to estimate the phreatic levels during operation after confirmation of the detailed schedules of construction and tailings deposition.
 - Geotechnical investigation including CPT tests to confirm the following:
 - o Properties of the constructed embankment.
 - o Tailings properties.
 - o Phreatic levels.
 - Update of this seepage analysis after the geotechnical investigation.

7.9 STABILITY ASSESSMENT

GeoStudio Slope/W program was used to determine the factor of safety of the TSF in a slope stability assessment for the design of the TSF expansion and to confirm the validity of the existing facility design. "SLOPE/W", developed by GEO-SLOPE International Ltd. in Canada (Ref. 13), is a limit equilibrium program and was performed using the modified Morgenstern-Price method. The detailed stability assessment is included in Appendix F.

The results summary is provided below:

- The results of the stability analyses demonstrate an adequate factor of safety under the static loading condition. For the revised ANCOLD Guidelines (Ref. 2), it is recommended to assume that a flow liquefaction event can occur. For this post-liquefaction loading condition the factor of safety 1.48;
- KP recommends CPT testing should continue to be undertaken in the tailings before future raises are constructed to increase the tailings strength database, and to confirm the properties of the constructed materials, tailings properties and phreatic levels; and



The modelled material parameters should be verified and confirmed during the
construction and operation phases. It is also crucial that the facility be operated
correctly and the supernatant pond be kept under control at all times to ensure
maximum distance from the embankment to the operational pond is maintained.

7.10 FREEBOARD

The TSF is designed to operate without the need for a spillway. A minimum Total Freeboard (water) of 500 mm should be maintained on the facility after a storm event together with a minimum Operational Freeboard (solids) of 300 mm, as outlined in the DMIRS Code of Practice (Ref. 4).

Freeboard allowances are set such that a PMP event can be retained on the facility without breach of the perimeter embankments. Following high rainfall events, increased decant return to the Plant will be used to lower the pond level and increase available storm storage capacity.

Definitions for freeboard are provided below:

Operational Freeboard (for solids) is the vertical height between the lowest elevation of the perimeter embankment and the tailings beach immediately inside the embankment. The operational freeboard varies over the course of a deposition cycle as the storage is raised and fills with tailings. The operational freeboard becomes critically important at the end of a deposition cycle, particularly to reduce the potential for back flow and overtopping as a result of mounding of tailings at discharge points. The Operational Freeboard is 300 mm.

Total Freeboard (for water) is the vertical height between the lowest elevation of the perimeter embankment and the pond level **AFTER** a 1% AEP 72 hour storm event assuming no decant or evaporation. As a result, it is difficult to visualise operationally if the storm event hasn't taken place. Hence the maximum pond level before the design event is determined mathematically to account for the potential inflow. The Total Freeboard after the event is then 500 mm.

Beach Freeboard is the vertical height between the normal operating pond level and the point on the beach where the operational freeboard is measured i.e. the exposed beach extent. The Beach Freeboard can vary significantly during the life of the storage and depends upon beach length, slurry/tailings characteristics and deposition methodology.



7.11 DAM BREAK ASSESSMENT

A tailings inundation exercise was conducted for the dam break scenario of the TSF expansion. The modelled inundation was conducted to assist in determining the areas and Population at Risk (PAR) due to a dam break.

A significant failure of the embankment will result in a release of tailings and/or water depending on the location of the breach, its size and the cause. For the analysis, the end condition was assessed as a tailings discharge resulting from a final embankment height breach when the facility is at its maximum tailings storage volume in a relatively dry condition. For this case the failure was modelled at both Cell C and Cell D embankment of the facility.

An embankment breach resulting in a release of slurry/solids was analysed from the ultimate facility height. A number of methods of analysing the flow of slurry and run out from a breached tailings dam have been presented over the years. These have been based on reviews of past failures and attempts to predict the flow slides using analytical means and simple correlations with facility geometry. The correlations proposed by M. Rico et al. (Ref. 14) and more recently Larrauri & Lall (Ref. 15) were used to correlate outflow volume from stored volume followed by correlating runout estimates from the outflow volume. Then a tailings deposition model (Rift TD) was used to generate potential inundation maps.

Rico (2007) proposed a correlation between stored tailings volume at the time of failure with tailings outflow volume from a historical data set of 29 dam failures of varying embankment configurations, tailings consistency and stored volumes. An empirical relationship with a good correlation ($r^2 = 0.86$) was found that determined, on average, about one-third of the tailings and water are released. It is expected that downstream embankments would typically be below this regression line and upstream embankments or high-water retaining facilities would be above this regression line.

The outflow volume was then correlated against recorded run out distances of tailings dam failures. A regression line is provided by Rico ($r^2 = 0.56$) however records varied significantly based on tailings water content, tailings viscosity, downstream obstacles, topographic restrictions, barriers etc. Therefore, run out distance estimates can vary significantly.

Additional information about TSF failures since Rico is available. The original liner regressions by Rico were updated by Larrauri & Lall with the results using the updated dataset. A new model for the calculation of Dmax was proposed introducing the predictor (Hf). This variable was introduced to consider that the potential energy associated with



the release volume may be better related to the fractional volume released as opposed to the total volume of the TSF.

In addition to the methodology above, another methodology developed by KP Australia was applied to the Thunderbox TSF. The logic behind the methodology is that the stronger the tailings mass, the steeper the resulting tailings breach will be. If the breach slope is very steep, then at a certain elevation the released tailings will block the outlet preventing further tailings from being released. An example of this concept is illustrated on Figure 7.1.

It is noted that both methodologies described above are considered to be compliant with the requirements of ANCOLD (Ref. 1 and Ref. 2) and are recognised as being appropriate by industry practitioners such as the Canadian Dam Association.

The calculation for the tailings/solids breach was based on the following assumptions:

- The dam breach could occur on either Cell D in the North or Cell C in the South.
 Therefore, both cases were considered in this analysis;
- Due to cells A and B upstream of cells C and D a cascading failure mechanism was considered:
 - Northern Failure solids release from Cell D and from Cell A.
 - Eastern Failure solids release from Cell C and Cell D only.
- Cells C and D, were assumed to have tailings storage volume at the final stage of approximately 23.6 Mm³ (approximately 13.1 Mm³ for Cell C and 10.5 Mm³ for Cell D) with a final embankment height of 20 to 22 m;
- Cells A and B, were assumed to have tailings storage volume at the final stage of approximately 26 Mm³ (for the purposes of this assessment this was assumed as approximately 13 Mm³ for Cell A and 13 Mm³ for Cell B) with a final embankment height of 23 to 29 m;
- An estimated outflow volume as a result of embankment failure of 3.8 Mm³ for cells A and B, 3.8 Mm³ for Cell C, 3.1 Mm³ for Cell D based on the recent Larrauri & Lall regression;
- The localised detailed topography around the immediate vicinity of the TSF Expansion was assessed for the case of a dry condition embankment slump type failure. Inundation areas based on beach slopes of 1V:25H, 1V:50H, 1V:75H, 1V:100H, 1V:125H, 1V:150H and 1V:175H were overlayed on Figure 7.2 for the breach occurring on the northern embankment and Figure 7.3 for the breach occurring at the divider between Cell C and Cell D; and



 For the breach occurring on the northern embankment, the area of impact is approximately 524 ha and the runout distance is about 3.6 km. For the breach occurring on the divider embankment, the area of impact is approximately 596 ha and the runout distance is about 3.5 km.



8. MONITORING AND MAINTENANCE

8.1 MONITORING PROGRAMME

As part of the TSF operation, extensive monitoring of all aspects of the operation will be undertaken to ensure the TSF design intent is achieved. This monitoring will comprise three basic types. These are:

Operation monitoring.

This will include items such as spigot offtake location, whether pipe joints are leaking, etc. This level of monitoring is directed at ensuring that the facility is operating smoothly.

Compliance monitoring.

This includes items such as checking survey pins for movement, monitoring bores for contamination, etc. This monitoring is required to ensure that the project is meeting all its commitments in regard to a safe and secure operation.

Performance monitoring.

This will include items such as tailings level surveys and water flow measurements, etc. This monitoring is necessary to assess the performance of the facility and refine future embankment lift levels and final extents.

The monitoring requirements for the facilities are summarised in Table 8.1.



Table 8.1: Monitoring programme

Area	Monitoring Requirement	Frequency	Method
Section 1:	Operation monitoring		
Tailings Facility	Pipeline integrity, visual inspection of corridor	Daily	Manual
	Offtake location	Daily	Manual
	Blockage or damage of discharge	Daily	Manual
	Visual check on tailings and water levels versus embankment crest and marker guide posts	Daily	Manual
	Visual integrity assessment of facility	Daily	Manual
Decant	Size of decant pond	Daily	Manual
	Pond level and associated freeboard	Daily	Manual
	Location of decant pond	Daily	Manual
Section 2:	Compliance Monitoring		
Embankment	Survey pins	Monthly	Manual
	Water volume estimate within facility	Monthly	Manual
	Tailings level and volume	Monthly	Manual
	Aerial topographic survey	Annual	Manual
	General inspection by suitably qualified engineer	Annually	Manual
Monitoring Bores	Water level	Monthly	Manual
	Water quality – conductivity	Quarterly	Manual
	Water quality – major component analysis	Quarterly	Manual
Piezometers	Water level	Monthly	Manual
Section 3:	Performance Monitoring		
Climatic	Precipitation	Hourly	Automatic
Tailings	Tailings solids (tonnes)	Daily	Automatic
	Water in tailings (tonnes or m³)	Daily	Automatic
	Average tailings flow (m ³ /s)	Daily	Automatic
Decant and	Outflow from decant (m³)	Daily	Automatic
underdrainage	Outflow from underdrainage (m³)	Daily	Automatic
Water Balance	Assessment of all water within the TSF to determine the need for additional water management	Quarterly	Manual
Technical Audit	Independent geotechnical engineer.	Annually	Manual

A general layout of the required key monitoring locations is presented on Drg. No. 801-296-D3000-900 with recommended monitoring/verification sections presented. The technical specification for installation of complex monitoring equipment will be in accordance with the manufacturer's specifications, with supervision by the equipment supplier recommended in order to ensure long term function of the equipment. More generic monitoring equipment will be installed by an appropriate company with experience in drilling the required locations and providing the required materials.



8.2 MAINTENANCE PROGRAMME

Inspection and maintenance of the TSF is largely aimed at mitigating potential problems by dealing with them before they develop into major problems.

Some aspects of the maintenance programme such as inspections can be integrated into the monitoring programme. If problems are detected then the problem is either to be corrected immediately or is to be noted and a maintenance request form filled in. The form will allow for different levels of urgency depending on how quickly the maintenance is required. The assessment of urgency should be based on the potential for the problem to affect the operation or integrity of the facility. The maintenance will then be integrated with the overall site maintenance programme.

Table 8.2 outlines the maintenance requirements for each area of the TSF. Modifications to the maintenance programmes as a result of emergency situations or annual reviews should be conducted regularly. Checklists for the Production Service Technician will be provided in the operating manual so that all inspections and faults are documented.



Table 8.2: Maintenance programme

Area	Monitoring Requirement	Frequency
Pipeline	Inspect pipeline for pipe bursts, leaking joints.	Daily
	Inspect offtake(s) for blockages, failure etc. Repair and/or replace as necessary. Re-locate offtake(s) to new location and check that flow at new location is acceptable.	Daily
	Inspect the pipeline corridor for signs of potential leakage.	Daily
	Carry out maintenance on valves and fittings as recommended by suppliers.	As recommended
Tailings	Inspect level of tailings versus embankment crest level.	Daily
	Inspect tailings for general gradient.	Daily
	Inspect discharge location for erosion or damage. Repair as appropriate.	Daily
Decant	Inspect decant pond and location. Adjust discharge location to maintain pond around decant tower.	Daily
	Inspect decant for damage or debris.	Weekly or after significant rain events
Embankment	Check general structural integrity and visual signs of seepage through embankment.	Daily
	Visual inspection for slips, erosion problems including around survey pins, tension cracks etc. Problems to be referred to qualified geotechnical engineer for assessment.	Weekly
Underdrainage System	Covered by tailings.	N/A
Toe Drains	Covered by tailings.	N/A
Instrumentation	Inspect all instrumentation and repair / replace as required.	Frequency as per instrumentation instructions
General	General inspection of tailings facility and all structures.	Annually
	General assessment of dust emissions.	Daily
	Geotechnical audit and report.	Annually
·	·	·



8.3 GEOTECHNICAL AUDIT REPORTS

It is important that suitably qualified geotechnical advice is available during the operating life of the facility. NSRL should keep the Geotechnical Engineer informed of the operating performance of the facility and raise any questions immediately.

As the TSF expansion is currently in a normal operation phase, annual audits should be conducted, provided that the operation and performance of the facility continues to be satisfactory.

A comprehensive Geotechnical Audit Report should be prepared and submitted to NSRL and DMIRS following each annual audit site visit.

8.4 INCIDENTS AND REPORTING

In the operation of the TSF there are many potential incidents that should be recorded, the reporting of such incidents can be spilt into two categories:

- Reportable internally these are incidents that may affect the day to day operations of the facility and are generally relatively minor. Knight Piésold would not normally be notified of these incidents unless specialist advice is sought; and
- Reportable externally these are serious incidents that impact the environment and the health and safety of employees and the public. These are significant incidents and are reportable by law to the relevant government agency. Knight Piésold should be notified and specialist advice sought.

As externally reportable incidents are extremely wide ranging, reference should be made to suitable published documents to determine which incidents may be applicable to the facility. Additionally, it is considered that should the licensed conditions be breached then it is likely to be an incident that is externally reportable.

Regardless of the reportable status of an incident, the incident can be split into three general categories:

- Environmental;
- · Structural; and
- Health and Safety (Personnel or Public).



Reporting of incidents is an extremely important task as part of the safe operation of a TSF. However, the procedure for reporting should not be one that is prescribed but one that is developed by NSRL within the organisation. NSRL has the following incident reporting procedures and a register managed via INX:

- If an incident occurs, an Incident Report Form is filled in, this is signed off by relevant managers and entered into INX (actions can be tracked until the incident is closed);
- If the incident is classified as major and requires follow-up (including externally reportable), an Investigation Report Form is completed, and this concludes with a Managers Review Meeting; and
- If it is a maintenance issue (i.e. no spill), a work order is raised on PRONTO and work assigned and conducted.



9. OPERATING AND EMERGENCY ACTION PLAN

9.1 GENERAL

A detailed emergency action plan will be developed as part of the operating manual for the TSF to manage emergency situations and unforeseen natural disasters. Potential consequences and remedial procedures will be presented in the operating manual for the following events:

- Power failure;
- Earthquake events;
- Extreme rainfall;
- Dam break/overtopping; and
- Excessive dust.

The consequences of the various situations and some remedial procedures are provided.

9.2 POWER FAILURE

The tailings pumps are connected to the process plant power distribution system and the decant return system is run utilising a genset located on the decant causeway. In the event of a total loss of power, all systems including all pumps and automatic valves not connected to standby generators will cease operations. Systems affected could include:

- Removal of water in the decant pond back to the plant;
- · Underdrainage pumping system; and
- Tailings pumping system.

Shutdown of the decant pump will mean that the level of water in the decant pond will slowly begin to rise if tailings continue to be discharged. Visual inspection of the supernatant pond level will be required. The available storage volume is very large, however if power cannot be reinstated within 3 days, temporary power facilities should be installed.

Shutdown of the underdrainage pumps will mean that the head of water overlying the underdrainage system will increase, and consequently the potential for seepage will also increase. If the power outage is likely to be for more than two days then an emergency plan should be initiated to supply temporary power to the underdrainage pumps.

Upon loss of power, the tailings pumping system will shut down. It is expected that, regardless of the length of stoppage, the tailings will be re-mobilised readily on re-commencing pumping. However, in the event that the tailings cannot be re-mobilised,



the pipeline(s) will need to be drained by manual valves into sumps constructed at the pipeline(s) low points.

Discontinuing tailings discharge will likely prevent decant water return from being required at the plant. A similar result to no water removal may occur.

9.3 EARTHQUAKE EVENTS

Recorded earthquake events have occurred in the Eastern-Goldfields region and accordingly the TSF has been designed to stringent stability criteria. However, the stability of the facility embankments should be checked annually as part of the review of the facility.

Contingency procedures to be adopted in the event of an earthquake are dependent on the intensity of the earthquake event at the site and are outlined below:

9.3.1 Major Earthquake Event (Modified Mercalli Intensity VI or greater)

Recognition: Difficulty standing, hanging objects quiver, masonry cracks, waves on ponds, some minor injuries (Ref. 3).

Response:

- Immediately terminate deposition of tailings into the TSF and pumping from the decant;
- ii. Immediate inspection of facility embankments and decants for obvious deformation or movement;
- iii. Immediate inspection of all pipes including residue pipeline(s) for rupture or leakage;
- iv. Arrange for immediate inspection and report of storage by a suitably qualified geotechnical engineer;
- v. Deposition can be recommenced if no major damage to facility and pipelines has occurred;
- vi. Survey pins and monitoring bores are to be read immediately after the event and all instrumentation to be read daily; and
- vii. A detailed daily inspection of the facility is to be undertaken until completion of the geotechnical engineer's report.



9.3.2 Minor Earthquake Event (Modified Mercalli Intensity V or less)

Recognition: Felt outdoors as well as indoors, liquid disturbed, small objects displaced, doors swing open or closed, pictures move (Ref. 3).

Response:

- i. Inspect pipelines for rupture or leakage. If required by damage to the lines, stop pumping until repairs are complete;
- ii. Inspect embankments and decants for obvious deformation or movement;
- iii. Monitor all instrumentation immediately after the event and weekly thereafter until readings return to normal; and
- iv. If any damage or leakage is observed, immediately arrange for an inspection by a suitably qualified geotechnical engineer.

9.4 EXTREME RAINFALL EVENTS

The tailings storage facility has a minimum flood storage capacity equivalent to a 1% AEP storm event. For the most part, the flood storage capacity will be well in excess of the design capacity. On this basis it is not expected that overtopping of the embankment will occur.

Major storm events (typically cyclonic) can cause serious erosion problems as well as other damage. A system of inspections should be carried out to mitigate any potential adverse consequences of extreme rainfall.

9.5 DAM FAILURE / OVERTOPPING

9.5.1 General

The TSF embankments are designed for defined operating capacities and specific storm event return periods. The Production Service Technicians working within the facility area will be trained in the correct method and approach to be used for the deposition of tailings.

An assessment has been made, for the facility, of the potential consequences of a release of contaminants (either tailings or water). Typical emergency action plans are provided for each type of release with specific details provided.

The analysis is based on an assessment of potential failure mechanisms, the consequences of each type of failure, and remedial measures that should be undertaken for each situation.



9.5.2 Emergency Action Plan – Contaminated Seepage

All tailings facilities incur seepage to varying degrees. The quantity and quality of the seepage depends on many factors including:

The underdrainage system is expected to intercept a proportion of the seepage. However, good operating practices will assist in reducing seepage, including:

- Reducing the decant pond size at all times by optimising management of the process plant;
- Maximising drying time of successive tailings layers to improve densification and therefore reduce permeability of the tailings;
- Continual operation of the underdrainage system; and
- Maximising the percent solids and density of the tailings.

Seepage from the facility and detection in the monitoring system will be handled as part of the general maintenance and monitoring programme. Monitoring boreholes will be located around the perimeter of the facility and the levels and chemical constituents of the groundwater in the bores will be monitored on a regular basis in accordance with the lease operating conditions.

If unacceptable changes in the concentration of groundwater chemical constituents or groundwater level are detected in the monitoring bores, environmental protocols need to be initiated, which address these issues.

9.5.3 Emergency Action Plan – Embankment Overtopping

As discussed, the risk of embankment overtopping is very low. Regular inspections will ensure that the volume of material which may potentially escape will be small. However, in the event of this situation occurring, the following procedures should be initiated:

- The deposition point should be re-located so that the loss of tailings out of the facility stops;
- An inspection should be made of the extent of the release;
- The Department of Mines Industry Regulation and Safety (DMIRS), the Department of Water and Environment Regulation (DWER) and other relevant authorities should be contacted / notified;
- The tailings after drying should be picked up and stored in the facility;
- Any environmental or other damage should be made good; and
- The area of the crest at which the spill occurred should be flagged off to indicate
 to the Production Service Technicians that no further deposition is to occur from
 this area until after the next embankment lift.



9.5.4 Emergency Action Plan – Dam Failure

This is the most critical type of emergency situation with the greatest hazard potential for damage. If such a dam failure occurred the following steps should be taken:

- Shut down the plant and cease deposition into the facility;
- Immediately on determining that a dam break has occurred, check on the location and safety of any personnel known to be in the area. If it is possible that someone has been caught in the flow slide, inform the emergency services (police rescue, etc.) and follow their recommendations;
- Any available suitable earthmoving equipment shall immediately be mobilised to a designated assembly area adjacent to the facility;
- If safe and practical to do so, the flow of supernatant / water from the facility shall be controlled as far as possible, utilising earthmoving equipment to form bunds with the intent of minimising flow into natural watercourses;
- Report the incident to the DMIRS, DWER and other relevant authorities;
- Inform Knight Piésold of the failure as soon as practicable;
- Inspect the flow area, determine the extent of the flow slide and record the damage it has caused;
- Inspect the facility to determine the area of the embankment which has been damaged;
- If the tailings line has been buried or damaged, disassemble the line, remove or repair the damaged sections and re-lay the pipeline into another area of the facility (if practicable) to allow continuation of tailings deposition when safe to do so;
- Fix any damage to the decant structure, pipelines, access roads, etc. as quickly as practicable;
- The breach and the overall facility should be inspected by a competent geotechnical engineer and a repair plan developed;
- Affected areas outside the facility should be inspected and a plan developed to recover the tailings;
- After the tailings has dried and the breach is stable, the tailings should be picked up and placed into the facility;
- The breach should be repaired and all damage to the environment made good;
- The conditions before the failure should be determined and the operating procedure modified so that the same situation does not occur again; and
- Re-commence tailings deposition when safe to do so.



9.6 EXCESSIVE DUST

Dust is a hazard both to site personnel and the surrounding environment and therefore should be controlled throughout the life of the TSF.

There are a number of distinct operations and effects associated with the TSF that have the potential to create dust emissions. The potential sources of dust are:

- Wind blow;
- Movement of vehicles around the site:
- Material excavation, handling and tipping operations; and
- Restoration works.

The following dust measures are considered to be adequate for the control of dust emissions from the TSF. These are:

- Regular and adequate employment of water spraying on unsurfaced areas to control dust emissions. The need for this control will be assessed continually, based upon on-site observations, weather conditions and the potential for dust emission across the site. In particular, dry conditions with high wind speeds would be a cause for action; and
- Progressive restoration and establishment of vegetative cover on the site as soon as possible following construction.

If it is considered essential that operations susceptible to dust are undertaken and personnel are to be exposed to elevated levels of dust, appropriate precautions should be undertaken which can include but are not limited to the following:

- Small localised dust suppression, including mist spray;
- Supply and wearing of appropriate dust masks;
- Provision of change and wash down facility for personnel;
- Personnel dust monitors; and
- Restriction of working periods in elevated dust environments.

The Contractor and Superintendent shall make random audits of dust emissions, and the Superintendent shall have the authority to stop work if, in his opinion, dust emissions are excessive or have the potential to create a safety or environmental problem.

The Contractor has their own safety procedures for tailings pipeline management as part of their contract. The Contractor is responsible for the implementation of proper construction, mining practices and safety precautions.



10. SAFETY IN DESIGN

10.1 GENERAL

As part of the requirements for the design of the TSF facility it is the responsibility of everyone involved to reduce risks associated with the facility as much as practicable.

The TSF has gone through a number of phases in its life cycle for the existing cells as outlined below and these will be repeated for the Cell C and Cell D design:

Phase 1: Initial concept development and design.

Phase 2: Initial construction including installation of equipment, pipework etc.

Phase 3: Discharge of tailings into facility and operation of water recovery and

control systems.

Phase 4: Review of operating data and design of Stage 2.

Phase 5: Construction of Stage 2.

Phase 6: Discharge of tailings.

Phases 7 to 21: Repeat of Phases 4 to 6 for subsequent stages.

Phase 22: Review of operating data and final design of facility closure.

Phase 23: Construction of closure components.

Phase 24: Monitoring and reporting on compliance with closure criteria

requirements.

10.2 DELINEATION OF RESPONSIBILITY

Each phase has separate requirements and may be under the control of different stakeholders. Table 10.1 provides a list of responsible parties for each of the current phases and an outline of the technical requirements for the responsible parties for future phases.



Table 10.1:	Designate	ed responsible	e parties	
Storio		Dooign		_

Stage	Design	Construction	Operation		
Cells C and D Expansion	Designer: Knight Piésold Pty Ltd	Clients Supervisor: Knight Piésold Pty Ltd	Operator: Northern Star Resources		
Stage 1	Owner: Northern Star Resources Limited	Contractor: T.B.A Owner: Northern Star Resources Limited	Limited		
Stages 2 to 6	To be designated (refer to Note 1)	To be designated (refer to Note 2)	Operator: (refer to Note 3)		
Closure	To be designated (refer to Notes 2 and 4)	To be designated (refer to Note 2)	Operator: (refer to Note 3)		

- Note 1: Review of operating parameters and design of facility stages and closure requires selection of an appropriate engineering firm with experience in tailings management (tailings) and civil engineering fields.
- Note 2: Client Supervisor responsibility requires selection of an appropriate engineering firm with experience in supervision of civil engineering/earthworks construction.
 - Contractor responsibility requires selection of an appropriate construction company with experience in construction of civil engineering facilities particularly construction of dams and interaction with mining fleet operations.
- Note 3: The current operator of the mine is Northern Star Resources Limited, and it is assumed that their responsibility will be extended to include the operation of the TSF.
- Note 4: The designer of the closure phase will also need to include experience in the design of TSF shutdown and closure / rehabilitation requirements.

It should be noted that as the current designer, KP will provide a design for all stages of the facility based on projected throughput and other data. This does not remove the need for design review and detailed construction documentation development for each future stage.

10.3 INCORPORATION OF SAFETY COMPONENTS IN DESIGN STAGE

10.3.1 General

There is a hierarchy of control measures to eliminate or reduce risks as follows:

- i. Elimination;
- ii. Substitution;
- iii. Isolation;
- iv. Engineering controls;
- v. Administrative controls; and
- vi. Personal protective equipment.

For the storage of tailings there are certain minimum requirements involving the need for a storage facility including embankment and methods for delivery of the tailings and removal of surplus water. As a result, many of the hazards associated with the facility cannot be eliminated and engineering and administrative control systems are required.



10.3.2 Engineering Controls

The design has incorporated a number of design aspects to reduce risks associated with the facility as follows:

- TSF Expansion integrating the development of the Eastern Waste Dump increases safety of the embankment for seismic loading and settlement;
- The embankment has been designed with zones to reduce seepage flow and improve stability by lowering the phreatic surface;
- A cut-off trench was constructed under the embankment to reduce seepage flows through the foundation material;
- Drains were constructed at the facility upstream toe to increase tailings densities near the embankment, reduce the phreatic surface and reduce seepage losses;
- Conservative parameters and safety factors have been used in the design;
- Design includes safety berms for all embankment crest areas; and
- The TSF Cell A and Cell B Expansion has been operated for several years and the initial TSF Cell C and Cell D Stage 1 is designed for an operating period of 2 years which will allow sufficient time to allow verification and confirmation of tailings beach slopes and achieved densities, well before construction of Stage 2 is required. The timing or height of Stage 2 can then be adjusted if required.

10.3.3 Knowledge Transfer to Downstream Responsible Parties

As the designer it is necessary to provide a clear understanding of the responsible parties for the construction and operational phases regarding the assumptions made for how the facility is to be constructed and operated. Primarily knowledge transfer is achieved by providing various documents outlining construction requirements and management procedures. These include:

Construction

- Provision of construction drawings for TSF Expansion Stages 2 to 6;
- A technical specification covering all aspects of construction of the facility including QA/QC requirements; and
- Specification for QA testing separate from the Contractor.

Operation

 An operating manual outlining deposition concepts and targets, minimum monitoring, maintenance requirements, required inspection levels and frequency.



10.3.4 Ongoing Review and Design Re-assessment

In order to ensure that the original design intent is met or alternatively that changes to the intent are backed up with detailed design work, the life cycle incorporates a series of regular checks on the operation and design. These include the following:

- Annual audits of the facility by a competent geotechnical engineer to comply with the requirements of the DMIRS guidelines (Ref.4);
- A design phase for each subsequent stage (stages at about 3 year frequency) involving review of operating parameters to date and any required modifications to the design based on forward projections for the next stage; and
- Testing of rehabilitation and closure requirements during the operating phase with ongoing monitoring after closure (as outlined in the closure documentation).

The ongoing review of the operation and design will ensure that the facility can be modified as required for any changes to the conditions and operation of the mine site.



11. CONSTRUCTION

Construction for each stage will be conducted by a dedicated earthworks contractor and will utilise mining operations for material source and delivery. A dedicated construction management team, technical supervision and quality assurance/quality control (QA/QC) is required for each stage of construction to ensure the facility is constructed to the design intent. At the completion of each subsequent embankment raise, an "as built" drawing set and construction report will be prepared. This will be completed so the report can be submitted to regulators within timeframes stipulated within the works approval. The report will collate the QA/QC records, document changes to the design to suit site conditions and reasons for the changes and verification of the material parameters actually used with an updated seepage and stability model if required.

11.1 SCOPE OF WORK

The scope of works for the contractor of each stage includes supply of all necessary personnel, equipment and materials, and undertaking of all works necessary to construct the embankment in accordance with the Drawings and Specifications. In summary, each stage construction works comprises the following elements:

- Mobilisation to site;
- Initial survey pickup for DTM;
- Construction of the TSF zoned embankment to the design levels and grades, including, erosion protection, wearing course and safety bunds;
- Construct concrete underdrainage tower base and towers with waste surround, access causeway and inlet excavation;
- Construct concrete decant base and slotted towers with rock surround, access causeway and inlet excavation;
- Supply and installation of monitoring boreholes and assist with the installation of piezometers and other instrumentation;
- Final and intermediate survey pickup for DTM; and
- Demobilisation from site.

11.2 TECHNICAL SPECIFICATION

11.2.1 Embankment Zones Placement

The embankment material shall be placed uniformly without abrupt changes in material type, quality or size for each zone. The following zones are required for the construction of the TSF.



Zone A

Zone A layer shall comprise low permeability fill meeting grading requirements outlined in Table 11.1. The material shall have a minimum plasticity index of 8.

Table 11.1: Zone A material grading requirements

Sieve Size (mm)	% Passing
19	100
9.5	50 – 100
2.36	30 - 100
0.075	30 - 90

Zone A material shall be material sourced from the tailings beach or Waste Dump. The Contractor shall condition the material to obtain moisture content suitable for placement and compaction. The method by which the contractor conditions the material shall be subject to the approval of the Engineer. The layer shall be compacted to density limits of at least 98% SMDD.

The material shall be placed and compacted in successive horizontal layers of loose material so that the maximum compacted layer depth is not greater than 300 mm.

Compaction of Zone A material shall be completed with vibratory pad foot compactor and sealed off using a smooth drum compactor in the case of wet weather. During the compaction process the embankment foundation shall be lightly watered or dried as required to maintain specified moisture content.

Zone C

Zone C is the structural fill zone. The material shall be sourced from borrow areas, existing stockpiles, excavations associated with the works, or other areas as directed by the Engineer.

The material shall have a Plasticity Index of at least 5% and shall have a grading conforming to the limits as given in the Table 11.2.

Table 11.2: Zone C gradation limits

Sieve Size	Percent	Passing
(mm)	Minimum	Maximum
300	100	100
37.5	70	100
19	30	100
4.75	10	100
0.425	0	70
0.075	0	30



Zone C1

Zone C1 construction shall be carried out in accordance with the following method specification:

- Any areas where proof rolling results in excessive heaving or rutting (to the point where the foundation becomes un-trafficable), shall be removed and replaced with approved Zone C1 fill.
- ii. The footprint area of each Zone C1 layer shall be proof rolled and nominally compacted by trafficking loaded haul trucks uniformly across the full bench width prior to commencing the next layer of filling. It is noted that clearing/stripping operations will effectively complete this task prior to placement of the first layer of Zone C1 fill.
- iii. The compacted layer thickness for oxidised waste material in Zone C1 shall be 0.5 0.7 m (nominally 0.7 1.0 m layer uncompacted). This layer thickness should be confirmed (and may be modified) with the Engineer once the material properties and vehicle type performing traffic compaction are confirmed.
- iv. The compacted layer thickness for fresh rock material in Zone C1 shall be 1.5 2.5 m.
- v. Any boulders (greater than 1.3 m diameter) within the Zone C1 areas shall be pushed to the downstream batter of Zone C1 (away from any Zone B / Zone C1 interface).
- vi. When paddock dumping waste during the wet season, piles of material shall be spread (and compacted) as soon as practicable, to reduce the risk of large volumes of uncompacted fill becoming saturated within the embankment profile. Saturated fill, caused by rainfall onto dumped piles left in place by the Contractor, shall be removed from the TSF embankment extents.
- vii. The material shall be placed and spread in a manner that ensures even trafficking across the entire platform by the hauling fleet.
- viii. Any completed fill areas that rut or significantly deform under haul truck trafficking, as designated by the Engineer, shall be excavated and the spoil hauled to a stockpile remote from the TSF embankment, as nominated by the Company.



- ix. Each Zone C1 layer shall be constructed in a manner that remains free draining and so that water sheds away from the completed embankment/previous layer.
- x. If required, trimming of the Zone C1 upstream face will be completed as part of the Zone A / Zone B raise construction. Zone C1 material shall be placed to ensure there is minimal surplus material requiring removal to achieve the final trimmed profile. This can be achieved by ensuring (via survey set out) the toe of each fill layer is located on the final Zone B / Zone C1 interface line.

Zone D

Zone D is random fill to be constructed from general fill material sourced from various locations around the site. Proposed materials may be sourced from borrow areas, excavations associated with the works, or other areas as directed by the Engineer.

Zone E

Zone E is a riprap erosion protection zone to be sourced from pit borrow or existing mine waste dumps by the Contractor. Proposed materials may also be sourced from any other excavation providing that the material is to the satisfaction of the Engineer.

The riprap shall be free of wood, steel, organics and other deleterious materials. Zone E material shall consist of hard, dense and durable rock and shall be well graded within the following limits, except as otherwise approved by the Engineer in writing:

- Maximum rock size shall be not greater than that which can be encompassed in the layer thickness shown on the Drawings.
- ii. Not less than 25 percent, by weight, of the material shall consist of rock having an average dimension greater than 200 mm.
- iii. Not more than 15 percent, by weight, of the material shall consist of rock having an average dimension of less than 75 mm.
- iv. A maximum fines content (<0.075 mm) of 5%.

The Zone E material need not be hand placed but may be dumped and spread by approved methods to a layer thickness as shown on the Drawings in such a manner as to ensure that the finer material is generally near the surface on which the rip rap is placed; that the larger rocks are generally near the outer surface; that the completed rip rap is stable without any tendency to slide and that there are no large unfilled voids. Hand working of the outer surface may be necessary to meet this requirement.



Zone G

Zone G decant surround material shall be obtained from waste dumps by the Contractor as directed by the Principal. Zone G material shall comprise sound, durable, clean, sub-angular to angular rock fragments and be free of wood, steel, organics and other deleterious material. Due to the intended purpose of the zone, the placed material shall comprise clean rock-fill and have a general particle size of between 100 mm and 300 mm and little or no fines (less than 5% passing 0.075 mm sieve).

Zone F

Zone F is classified as a free draining sand/gravel drainage medium. The drainage medium will be sourced from the nominated borrow as directed by the Principal. The target grading envelope for the material is given in Table 11.3. To achieve the specified grading requirements screening may be required.

Table 11.3: Zone F grading requirements

Sieve Size (mm)	% Passing
19	100
2.36	40 - 100
0.425	0 - 30
0.075	0 - 5

Wearing Course

The wearing course shall consist of durable, select laterite/gravel or other suitable material approved by the Engineer. Wearing course material shall be sourced from borrow areas, existing stockpiles, excavations associated with the works, or other areas as directed by the Engineer.

The wearing course shall be free from cobbles, stumps, roots, sticks, vegetable matter and other deleterious matter. It shall be placed to the lines and grades as indicated on the Drawings. Wearing course material shall have a gradation as specified in Table 11.4.



AS test sieve	Percentage by mass of total aggregate passing test sieve								
	Nominal maximum particle size								
	37.5 mm	20 mm							
63.0	100	-							
37.5	80 – 100	100							
19.0	60 – 80	80 – 100							
9.5	45 – 65	55 – 80							
4.75	30 – 50	40 – 60							
2.36	20 – 40	30 – 50							
0.425	10 – 25	12 – 27							
0.075	5 – 15	5 – 15							

In addition, the wearing course material should have a plasticity index of between 0 and 10%, and a liquid limit of between 0 and 25%.

Topsoil

Topsoil is defined as soil of any gradation or degree of plasticity which contains significant quantities of visually identifiable vegetation, sod, roots or humus. Topsoil material will be won from stripping operations and kept in stockpile, until being utilised for rehabilitation of the site.

11.2.2 Quality Assurance

This section details all quality control and acceptance testing associated with earthworks items that is to be carried out by the Engineer. The extent of testing shall be at the discretion of the Company Representative and generally as detailed herein.

Tests shall be carried out as the work progresses to ensure that construction conforms to the requirements of this Specification. All tests shall be carried out in accordance with AS1289 (Methods of Testing Soils for Engineering Purposes) unless stated otherwise.

Testing frequencies for quality control shall be as advised by the Engineer. Table 11.5 summarises the minimum amount of testing required.



Table 11.5: Quality control and record testing

Type of Test	Frequency (Once per)
Zone A (Tailings)	
Atterberg Limits	5,000 m ³
Particle Size Distribution (PSD)	5,000 m ³
Moisture – Density Relationship	5,000 m ³
Field density and moisture content	500 m³
Zone A (Mine waste)	
Atterberg Limits	2,500 m ³
Particle Size Distribution (PSD)	2,500 m ³
Moisture – Density Relationship	2,500 m ³
Field density and moisture content	500 m³
Zones E & G	
Particle Size Distribution (PSD)	5,000 m³
Thickness Verification	
Survey Instrument grid Point	100 m²
Direct measurement via auger/excavation hole or other approved method	5,000 m ²



12. REHABILITATION

12.1 EMBANKMENT PROFILE

At the end of the TSF operation, the final stage downstream face of the TSF embankment will have an overall slope of 1V: 3.5 H (15.9°). The adopted downstream profile will be inherently stable under both normal and seismic loading conditions, will provide a stable drainage system, and will allow for re-vegetation.

With the proposed upstream construction, rehabilitation can commence a number of years before planned closure to allow the performance to be monitored and modified as necessary. The embankment downstream face will be rehabilitated with topsoil which has been stockpiled from construction activities.

12.2 CAPPING

Based on the geochemistry results available, the tailings are assessed to be enriched with metals, and high arsenic concentration. Closure of the facility will require a cover system including a capillary break to prevent upward migration of salinity, a low permeability layer to reduce ingress of water and oxygen and an erosion protection layer overlain. This will be reviewed in the detailed rehabilitation phase.

The main focus of the rehabilitation programme will be re-vegetation, erosion control and stormwater management. Establishing a surface cover of vegetation will reduce the potential for adverse environmental impact such as dust generation and rainfall erosion, as well as improving aesthetics.

Rehabilitation of the tailings surface will commence upon termination of tailings deposition. After removal of the pond around the decant area, drying of the tailings is expected to take several months. The underdrainage system will need to continue to operate for a number of years after completion of capping and re-vegetation to drain excess water from the tailings deposit.

The final profile of the tailings surface will form a water harvesting structure towards decant towers to collect rainfall for vegetation growth. A more detailed decommissioning/closure plan will be developed for the tailings storage facility during the final years of operation.

12.3 INFRASTRUCTURE CLOSURE

The removal of infrastructure at closure will be staged and will proceed through the following steps:

 The final supernatant pond will progressively evaporate after tailings deposition has ceased:



- The tailings surface will be allowed to dry until it has sufficient strength to provide access;
- The tailings distribution pipelines will be removed from the embankment;
- The decant pump and associated electrical and mechanical infrastructure will be removed:
- The decant tower will be backfilled with coarse rock to the surface of the causeway;
- The underdrainage system will continue to be operated intermittently until the quantity of water reduces. No set duration can be specified, but typically this will be in the order of 6 – 12 months. After the flow reduces, the underdrainage pump will be removed and the underdrainage tower will be backfilled with coarse rock;
- The embankment external faces will be rehabilitated if not already done so as part of the final embankment stage works earlier;
- The cover design will be placed over the tailings surface, with topsoil and revegetation; and
- The monitoring bores will continue to be monitored as set out by any permitting requirements.



13. REFERENCES

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- Guidelines on Tailings Dams Planning, Design, Construction, Operation and Closure, ANCOLD, Revision 1, July 2019.
- Guidelines for Design of Dams and Appurtenant Structures for Earthquake, ANCOLD, May 2019.
- Department of Mines, Industry Regulation and Safety (DMIRS) "Code of Practice – Tailings storage facilities in Western Australia" 2013.
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- 6. Department of Mines, Industry Regulation and Safety (DMIRS) "Guide to Departmental Requirements for the Management and Closure of Tailings Storage Facilities (TSFs)", August 2015.
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- 10. The 2018 National Seismic Hazard Assessment for Australia, Geoscience Australia: Canberra.
- 11. Knight Piésold Pty Ltd: Thunderbox Operation, TSF Expansion Stage 7-9 permitting design Rev 0, Sep 2018, Ref: PE801-00296/11.
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- 16. Knight Piésold Pty Ltd: Tailings Management Facility Audit, Nov 2018, Ref: PE801-00296/13.



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TABLES



Year	Month	Rainfall	Evaporation	Cumulative Tonnage	Tailings Level	Tailings Layers	density Mass	Water in Slurry	Supernatant Runoff	Rainfall Runoff	Evaporation Losses	Pond Volume	Consolidation Volume	Available Recycle		Discharge	Make Up Requirement
		mm	mm	t	RL m	t/m3	t/m3	mª	m*	m*	m*	m ^a	m*	m²	%	m*	m ^a
	Aug-22 Sep-22	35.1	136 198.9	279,517 559,033	486.36 487.24	1,35 1,41	1.35	170,295	54,254 51,859	6,734 173	1,666 3,202	5,251	727	2 .	0.0%	16	170,295 170,295
27	Oct-22	27.8	285.2	838,550	487.63	1.54	1.43	170,295	49,316	9,722	4,776	5,000	199		0.0%		170,295
1	Nov-22	0.7	332.8	1,118,067	488.00	1.51	1.47	170,295	47,443	279	5,715	5,000		5,779	3.4%		164,516
-	Dec-22 Jan-23	0	388.8 404.1	1,397,584	488.36 488.72	1.61 1.61	1.51	170,295	45,998 45,468	887	6,883 7,204	5,000	1,529	2,689 1,926	1.6%		167,606 168,369
	Feb-23	19.2	320.8	1,956,617	489.06	1.61	1.55	170,295	46,935	10,192	5,462	5,000	798	1,020	0.6%	1	169,275
	Mar-23 Apr-23	17.5 57.7	288.7 192.9	2,236,134 2,515,651	489.31 489.57	1.61	1.56	170,295 170,295	48,975 51,737	9,616 31,877	4,817 3,036	5,000	493 340	3,375	2.0%		166,920 170,295
	May-23	29.2	128.3	2,795,167	489.84	1.50	1.56	170,295	53,932	16,205	1,923	5,000	207	882	0.5%	E	169,413
	Jun-23 Jul-23	20.7 13.9	90.5 98.4	3,074,684	490.13	1.44	1,55	170,295	55,007	11,555	1,317 1,439	5,000	178 164	-	0.0%	168	170,295
	Aug-23	35.1	136	3,354,201 3,633,718	490.41	1,46 1.53	1.54	170,295 170,295	54,856 53,692	7,796 19,788	2,051	5,000	144		0.0%	, F.	170,295
	Sep-23	0.6	198.9	3,913,234	490.92	1.61	1,55	170,295	51,539	340	3,127	5,000	595		0.0%	1.0	170,295
	Oct-23 Nov-23	27.8	285.2 332.8	4,192,751 4,472,268	491.16 491.41	1.61	1.56	170,295 170,295	49,072 47,254	15,810	4,751 5,693	5,000	1,849 2,157	5,779	3.4%	(R)	170,295 164,516
2	Dec-23	2	388.8	4,751,784	491.65	1.61	1.57	170,295	45,863	1,144	6,864	5,000	1,199	2,689	1.6%		167,606
1	Jan-24 Feb-24	19.2	404.1 320.8	5,031,301 5,310,818	491.89 492.14	1.51	1.58	170,295	45,388 46,902	11,056	7,193 5,458	5,000	824 621	1,926 1,020	0.6%		168,369 169,275
1	Mar-24	17.5	288.7	5,590,335	492.38	1.61	1.58	170,295	48,959	10,087	4,815	5,000	483	59,070	34.7%		111,225
	Apr-24	57.7	192.9	5,869,851	492.63	1.61	1.58	170,295	51,726	33,363	3,035	5,000	509	49,082	28.8%	- 2	121,213
4	May-24 Jun-24	29.2 20.7	128.3 90.5	6,149,368 6,428,885	492.89 493.16	1.52	1.58	170,295	53,926 55,003	16,921 12,034	1,923 1,316	5,000	433 389	54,263 42,006	31.9%	727	116,032 128,289
	Jul-24	13.9	98.4	6,708,402	493.42	1.46	1.57	170,295	54,852	8,099	1,439	5,000	358	40,002	23.5%		130,293
	Aug-24	35.1	136	6,708,402	493.42	-	1.57		-	20,451	2,045	5,000	315	39,792	0.0%		
	Sep-24 Oct-24	0.6 27.8	198.9 285.2	6,708,402 6,708,402	493.42 493.42		1.57		7	350 16,197	3,114 4,726	5,000	273 50	52,463 54,267	0.0%	-	
23	Nov-24	0.7	332.8	6,708,402	493.42	- 8	1.57	- 4	-	408	5,658	5,000	10	80,918	0.0%	161	
3	Dec-24 Jan-25	0	388.8 404.1	6,708,402 6,708,402	493.42 493.42		1.57	9		1,165	6,820 7,145	5,000	0	65,421 65,423	0.0%	N. T.	
	Feb-25	19.2	320.8	6,909,587	493.59	1.61	1,57	122,572	33,754	11,177	5,449	5,000	0	61,377	50.1%		61,195
	Mar-25 Apr-25	17.5 57.7	288.7 192.9	7,110,773 7,311,959	493.76 493.93	1.61	1.58	122,572	35,236 37,229	10,192 33,672	4,808 3,032	5,000	947	71,573 49,347	58.4% 40.3%	(8)	50,999 73,224
8	May-25	29.2	128.3	7,513,144	494.10	1.61	1.58	122,572	38,813	17,062	1,921	5,000	1,948	61,979	50.6%		60,593
	Jun-25	20.7	90.5	7,714,330	494.28	1.52	1.58	122,572	39,588	12,121	1,316	5,000	1,282	44,118	36.0%		78,454
*	Jul-25 Aug-25	13.9 35.1	98.4 136	7,915,516 8,116,701	494.45 494.62	1.55	1.58 1.58	122,572	39,479 38,641	8,150 20,616	1,438 2,049	5,000	959 784	41,341 39,019	33.7% 31.8%	-	81,231 83,553
	Sep-25	0.6	198.9	8,317,887	494.79	1,61	1.58	122,572	37,089	353	3,123	5,000	625	53,120	43.3%	4.00	69,452
	Oct-25 Nov-25	27.8 0.7	285.2 332.8	8,519,073 8,720,259	494.96 495.13	1.61	1.59	122,572	35,311 34,001	16,382 413	4,743 5,682	5,000	499 452	54,714 82,563	44.6% 57.4%	72	67,858 40,009
4	Dec-25	2	388.8	8,921,444	495.30	1,61	1.59	122,572	32,999	1,182	6,850	5,000	488	69,358	56.6%		53,214
	Jan-26	0	404.1	9,122,630	495.47	1.61	1.59	122,572	32,657	11 207	7,178	5,000	406	66,111	53.9%		56,461
	Feb-26 Mar-26	19.2 17.5	320.8 288.7	9,323,816 9,525,001	495.63 495.80	1.61	1.59	122,572	33,749 35,232	11,397 10,392	5,448 4,807	5,000	355 318	61,869 18,720	50.5% 15.3%	127	50,703 103,852
	Apr-26	57.7	192.9	9,726,187	495.97	1.61	1.59	122,572	37,226	34,327	3,032	5,000	289		0.0%	18	122,572
	May-26 Jun-26	29.2	128.3 90.5	9,927,373	496.14 496.31	1.51	1.59	122,572	38,811 39,587	17,392 12,352	1,921	5,000	266 262	11,521	9.4%	5.50	111,050
	Jul-26	13.9	98.4		496.49	1,56	1.59	122,572	39,478	8,305	1,438	5,000	256		0.0%		122,572
	Aug-26	35.1	136	10,530,930	496.66	1.61	1.59	122,572	38,639	21,003	2,049	5,000	252	- 20.402	0.0%	100	122,572
1 3	Sep-26 Oct-26	0.6 27.8	198.9 285.2	10,732,116	496.82 496.99	1.61	1,59	122,572	37,086 35,308	360 16,684	3,122 4,743	5,000	234	39,482 40,620	32.2%		83,090 81,951
	Nov-26	0.7	332.8	11,134,487	497.15	1.61	1.59	122,572	33,997	421	5,681	5,000	190	68,816	56.1%	-	53,756
5	Dec-26 Jan-27	0	388.8 404.1	11,335,673	497.31	1.61	1.59	122,572	32,995 32,653	1,201	6,850 7,178	5,000	158 153	55,902 51,676	45.6% 42.2%	2.40	66,670 70,896
	Feb-27	19.2	320.8	11,536,858	497.47	-	1.60	-	-	11,528	5,424	5,000	211	47,150	0.0%		
	Mar-27	17.5	288.7	11,536,858	497.47	2	1.60		2	10,507	4,790	5,000	174	57,991	0.0%		-
3	Apr-27 May-27	57.7 29.2	192.9 128.3	11,536,858 11,536,858	497.47	3	1,60		-	34,644 17,532	3,023 1,917	5,000	46 19	34,944 47,450	0.0%		-
	Jun-27	20.7	90.5	and the second second second second	497.47		1.60			12,429	1,314	5,000	9	29,185	0.0%	-	
	Jul-27 Aug-27	13.9 35.1	98.4 136	11,536,858	497.47	1,61	1.60	126,790	39,968	8,346 21,079	1,435 2,049	5,000	5 2	27,819 25,885	20.4%		100,904
1	Sep-27	0.6	198.9		497.81	1.61	1.60	126,790	38,362	360	3,123	5,000	1	40,053	31.6%	(6)	86,737
	Oct-27	27.8	285.2 332.8	12,161,184	497.98	1.61	1.60	126,790	36,523	16,689	4,743	5,000	136 144	41,134	32,4%	3.5	85,655
6	Nov-27 Dec-27	0,7	388.8	12,369,293	498.15 498.31	1.61	1.60	126,790 126,790	35,167 34,130	1,200	5,682 6,851	5,000	122	68,810 54,547	54.3% 43.0%		57,980 72,243
-	Jan-28	0	404.1	12,785,510	498.48	1.61	1.60	126,790	33,777	-	7,179	5,000	115	50,886	40.1%	161	75,903
1	Feb-28 Mar-28	19.2 17.5	320.8 288.7	12,993,618	498.65 498.82	1.61	1.60	126,790 126,790	34,907 36,442	11,532 10,497	5,448 4,808	5,000	109 95	46,601 57,845	36.8% 45.5%		80,189 68,945
	Apr-28	57.7	192.9	13,409,835	498.99	1.61	1.60	126,790	38,506	34,615	3,032	5,000	91	34,558	27.3%	727	92,232
- 8	May-28 Jun-28	29.2	128.3 90.5	13,617,944	499.15 499.33	1.61	1.60	126,790 126,790	40,146	17,508 12,413	1,921	5,000	85 87	47,461 28,927	37.4% 22.8%	(E	79,329 97,863
1	Jul-28	13.9	98.4		499.50	1,55	1.60	126,790	40,836	8,331	1,438	5,000	86	27,504	21.7%	-	99,286
1	Aug-28	35.1	136	14,034,161	499.50		1.60			21,036	2,045	5,000	86	25,629	0.0%	-	
>	Sep-28 Oct-28	0.6 27.8	198,9 285.2	14,034,161	499.50 499.50	-	1.60			360 16,661	3,114 4,726	5,000	83	6,315 5,892	0.0%		
	Nov-28	0.7	332.8	14,034,161	499.50	-	1.60	94	-	420	5,658	5,000	20	31,666	0.0%	- 1	-
7	Dec-28 Jan-29	0	388.8 404.1	14,034,161	499.50 499.50		1.60	84		1,199	6,820 7,145	5,000	11 6	15,634 11,124	0.0%	127	-
3	Feb-29	19.2	320.8	14,278,335	499.69	1.61	1.60	148,763	40,957	11,541	5,453	5,000	3	6,914	4.5%	i Ri	141,848
3	Mar-29 Apr-29	17.5 57.7	288.7 192.9	14,522,509	499.88 500.08	1.61	1.60	148,763 148,763	42,758 45,179	10,502 34,634	4,811 3,033	5,000	42	59,000 35,601	39.7% 23.9%		89,763 113,161
	May-29	29,2	128.3	15,010,858	500.08	1.58	1.60	148,763	47,103	17,515	1,922	5,000	167	48,605	32.7%	- 6	100,158
	Jun-29	20.7	90.5	15,255,032	500.48	1.48	1,60	148,763	48,046	12,418	1,316	5,000	250	30,050	20.2%	16	118,713
	Jul-29 Aug-29	13.9 35.1	98.4 136	15,499,206 15,743,380	500.69	1.50	1.60	148,763 148,763	47,913 46,895	8,333 21,037	1,439 2,050	5,000	205 179	28,602 26,713	19.2%	-	120,161
	Sep-29	0.6	198.9	15,987,555	501.08	1.61	1.60	148,763	45,011	360	3,124	5,000	155	41,099	27.6%	720	107,664
	Oct-29 Nov-29	27.8 0.7	285.2 332.8	16,231,729 16,475,903	501.27 501.46	1.61 1.61	1.60 1.60	148,763 148,763	42,853 41,263	16,653 419	4,746 5,686	5,000	1,012 1,228	42,227 70,180	28.4% 47.2%		106,536 78,583
8	Dec-29	2	332.8	16,475,903	501.46	1.61	1.60	148,763	40,047	1,197	6,856	5,000	739	55,819	37.5%		92,944
	Jan-30	0	404.1	16,964,251	501.84	1.61	1.60	148,763	39,632	200	7,185	5,000	510	52,133	35.0%	-	96,629
>	Feb-30 Mar-30	19.2 17.5	320.8 288.7	16,964,251 16,964,251	501.84	-	1.60	5		11,493	5,424 4,790	5,000	400 331	47,815 19,078	0.0%	2.49 (R)	
1	Apr-30	57.7	192.9	16,964,251	501.84		1.60	9	= 2	34,539	3,023	5,000	42		0.0%	E .	181
	May-30 Jun-30			16,964,251 16,964,251	501.84 501.84		1.60 1.60			17,479 12,391	1,917 1,314	_	22 14	11,974	0.0%	3	
3	Jul-30	13.9		16,964,251	501.84	-	1.60	-	-	8,320	1,436		9	-	0.0%	161	
1	Aug-30	35.1	136	17,215,707	502.04	1.59	1.60	153,199	48,293	21,013	2,050	5,000	6	-	0.0%	5%	153,199
	Sep-30 Oct-30	0.6 27.8		17,467,162	502.24 502.44	1.61	2000000	153,199 153,199	46,354 44,131	359 16,634	3,125 4,747	THE RESERVE THE PERSON NAMED IN	293	47,049 48,451	30.7%	120	106,150
100 30	Nov-30	0.7	332.8	17,970,073	502.64	1.61	1,60	153,199	42,494	419	5,687	5,000	255	76,821	50.1%	16	76,378
9	Dec-30	2	388.8	18,221,528	502.84	1.61	1,60	153,199	41,241	1,196	6,857	5,000	215	62,864	41.0%	-	90,335

TABLE 6.4: TSF CELL C WATER BALANCE - AVERAGE CONDITIONS

Year	Month	Rainfall mm	Evaporation	Cumulative Tonnage	Tailings Level	Tailings density		Water in Slurry	Supernatant Runoff	Rainfall Runoff	Evaporation Pond		Consolidation Volume	Available	TSF	Discharge	Make Up
rear	Month		mm		Level	Layers	Mass	ma	m*	Aunun	Losses	Volume	volume	Recycle		0 0 22 1	Requirement
				t	RL m	t/m3	t/m3			m*	m*	mª	m ^a	mª.	96	m*	m*
	Jan-31	. 0	404.1	18,472,984	503.04	1.51	1.60	153,199	40,815	20	7,185	5,000	189	59,398	38.8%	1.61	93,801
	Feb-31	19.2	320,8	18,724,439	503.24	1.61	1.60	153,199	42,180	11,492	5,454	5,000	171	55,013	35.9%	(8)	98,185
	Mar-31	17.5	288.7	18,975,894	503.44	1.61	1.60	153,199	44,034	10,457	4,811	5,000	152	66,061	43.1%		87,137
	Apr-31	57.7	192.9	19,227,350	503.65	1.51	1.60	153,199	46,527	34,485	3,034	5,000	143	42,401	27.7%	-	110,797
	May-31	29.2	128.3	19,478,805	503.85	1.57	1.60	153,199	48,509	17,439	1,922	5,000	136	55,772	36.4%	727	97,427
	Jun-31	20.7	90.5	19,730,260	504.07	1.47	1.60	153,199	49,479	12,365	1,316	5,000	121	37,224	24.3%		115,975
	Jul-31	13.9	98.4	19,981,716	504.29	1.49	1.60	153,199	49,343	8,297	1,439	5,000	119	35,127	22.9%	18	118,071
	Aug-31	35.1	136	19,981,716	504.29		1.60	J 20 3	-	20,951	2,045	5,000	106	32,958	0.0%		151
	Sep-31	0.6	198,9	19,981,716	504.29	- 1	1.60	- 3		358	3,114	5,000	99	6,469	0.0%		-
	Oct-31	27.8	285.2	19,981,716	504.29	121	1.60	. 6	. Fa	16,594	4,726	5,000	35	6,017	0.0%	E .	
	Nov-31	0.7	332.8	19,981,716	504.29	9.	1,60	19	- 4	418	5,658	5,000	14	31,558	0.0%	193	18
10	Dec-31	2	388.8	19,981,716	504.29	H1 11	1.60	25	e 2	1,194	6,820	5,000	7	15,583	0.0%	-	
	Jan-32	0	404.1	19,981,716	504.29		1.60	2 . 3		70	7,145	5,000	4	11,091	0.0%	4.74	
	Feb-32	19.2	320.8	20,232,235	504.49	1.61	1.60	152,628	42,023	11,473	5,454	5,000	2	6,894	4.5%	-	145,734
	Mar-32	17.5	288.7	20,482,754	504.68	1.51	1.60	152,628	43,870	10,440	4,811	5,000	2	67,263	44.1%	161	85,366
	Apr-32	57.7	192.9	20,733,273	504.88	1.61	1.60	152,628	46,354	34,427	3,034	5,000	896	43,592	28.6%	5.50	109,036
11	May-32	29.2	128.3	20,983,791	505.08	1.57	1.60	152,628	48,328	17,410	1,922	5,000	1,452	56,311	36.9%		96,317

Year	Month	Rainfall	Evaporation	Cumulative Tonnage	Tailings Level	Tailings		Water in Slurry	Supernatant Runoff	Rainfall Runoff	Evaporation Losses	Pond Volume	Consolidation Volume	Available		Discharge	Make Up Requirement
		mm	mm	t	RLm	Layers t/m3	Mass t/m3	mª	m².	m ^a	m*	m*	m ⁴	Recycle m ^a	%	m²	m*
	Aug-22	35,1	136	124,355	488.27	1.35	1.35	75,763	24,445	5,488	1,728	5,194	721	28,011	37.0%	127	47,752
	Sep-22 Oct-22	0.6 27.8	198.9 285.2	248,710 373,065	489.18 489.55	1.42	1.39	75,763 75,763	23,272	125 6,653	3,185 4,769	5,000		20,406	26.9% 31.7%	1 2	55,357 51,739
1	Nov-22	0.7	332.8	497,420	489.91	1.59	1.48	75,763	21,310	184	5,708	5,000		15,786	20.8%	3	59,977
1	Dec-22	0	388.8	621,775	490.27	1.60	1.51 1.53	75,763	20,663	569	6,875	5,000		14,357	18.9% 18.3%	5	61,406 61,934
į.	Jan-23 Feb-23	19.2	404.1 320.8	746,130 870,485	490.62 490.97	1.60	1.54	75,763 75,763	20,410	6,319	7,197 5,458	5,000	617 343	13,829 22,227	29.3%	6	53,536
	Mar-23	17.5	288.7	994,840	491.15	1.60	1,55	75,763	21,888	6,099	4,812	5,000	212	23,387	30.9%	7	52,376
1	Apr-23 May-23	57.7 29.2	192.9 128.3	1,119,195 1,243,550	491.32 491.49	1.60	1.56 1.57	75,763 75,763	23,076 24,026	21,123 11,188	3,033 1,922	5,000	154 71	41,320 33,364	54.5% 44.0%	8	34,443 42,399
Ů.	Jun-23	20.7	90.5	1,367,906	491.67	1.51	1.56	75,763	24,492	8,301	1,316	5,000	55	31,533	41.6%	10	44,230
	Jul-23	13.9	98.4	1,492,261	491.85	1,56	1,56	75,763	24,422	5,823	1,438	5,000	50	28,857	38.1%	11	46,906
l l	Aug-23 Sep-23	35.1 0.6	136 198.9	1,616,616	492.02	1.61	1.57	75,763 75,763	23,907 22,955	15,315 272	2,049 3,121	5,000	47	37,221 20,147	49.1% 26.6%	12	38,542 55,616
	Oct-23	27.8	285.2	1,865,326	492.35	1.61	1.58	75,763	21,860	13,073	4,740	5,000	67	30,260	39.9%	14	45,503
25	Nov-23	0.7	332.8	1,989,681	492.52	1.61	1,58	75,763	21,050	341	5,677	5,000	201	15,916	21.0%	15	59,847
2	Dec-23 Jan-24	0	388.8 404.1	2,114,036 2,238,391	492.69 492.85	1.61	1.58	75,763 75,763	20,427	1,009	6,843 7,169	5,000	132 102	14,725	19.4%	16 17	61,038 52,621
	Feb-24	19.2	320.8	2,362,746	493.01	1.61	1.59	75,763	20,876	10,330	5,441	5,000	83	25,848	34.1%	18	49,915
	Mar-24	17.5	288.7	2,487,101	493.13	1.61	1.59	75,763	21,788	9,558	4,802	5,000	71	26,615	35.1%	19	49,149
	Apr-24 May-24	57.7 29.2	192.9 128.3	2,611,456 2,735,811	493.24 493.35	1.61	1.59	75,763 75,763	23,017 23,994	31,630 16,058	3,029 1,920	5,000	59 40	51,677 38,173	58.2% 50.4%	20 21	24,086 37,591
1	Jun-24	20.7	90.5	2,860,166	493.47	1.61	1.59	75,763	24,472	11,426	1,315	5,000	36	34,619	45.7%	22	41,144
	Jul-24	13.9 35.1	98.4 136	2,984,521	493.58 493.58	1.61	1.59	75,763	24,405	7,697 19,435	1,437 2,045	5,000	35 34	30,699 17,425	40.5%	23 24	45,064
3	Aug-24 Sep-24	0.6	198.9	2,984,521 2,984,521	493.58	-	1.59		-	332	3,114	5,000	32	- 17,423	0.0%	25	
	Oct-24	27.8	285.2	2,984,521	493.58	-	1.59	- 4	-	15,393	4,726	5,000	7	10,674	0.0%	26	-
3	Nov-24	0.7	332.8 388.8	2,984,521	493.58 493.58	-	1.59 1.59	- 94	-	388 1,107	5,658	5,000	2	-	0.0%	27 28	9
3	Jan-25	0	404.1	2,984,521 2,984,521	493.58	-	1.59			1,107	5,820 7,145	5,000	0		0.0%	28	
3	Feb-25	19.2	320.8	3,176,291	493.76	1.61	1.59	116,835	32,184	10,729	5,449	5,000	0	37,465	32.1%	30	79,370
	Mar-25 Apr-25	17.5 57.7	288.7 192.9	3,368,060 3,559,829	493.93 494.11	1.61	1.60	116,835 116,835	33,594 35,491	9,819 32,555	4,808 3,032	5,000	0 21	38,605 65,035	33.0% 55.7%	31 32	78,230 51,800
1	May-25	29.2	128.3	3,751,599	494.28	1.61	1.60	116,835	36,999	16,554	1,921	5,000	20	51,652	44.2%	33	65,183
	Jun-25	20.7	90.5	3,943,368	494.47	1.53	1.59	116,835	37,737	11,801	1,315	5,000	20	48,243	41.3%	34	58,592
6	Jul-25 Aug-25	13.9 35.1	98.4 136	4,135,138	494.65 494.82	1.55	1.59	116,835 116,835	37,633 36,834	7,964	1,438 2,049	5,000	19 18	44,178 55,018	37.8% 47.1%	35 36	72,657 61,817
1	Sep-25	0.6	198.9	4,518,677	495.00	1.61	1.60	116,835	35,355	347	3,122	5,000	17	32,597	27.9%	37	84,238
,	Oct-25	27.8	285.2	4,710,446	495.16	1.61	1.60	116,835	33,661	16,148	4,742	5,000	15	45,081	38.6%	38	71,754
4	Nov-25 Dec-25	0.7	332.8 388.8	4,902,216 5,093,985	495.33 495.50	1.61	1.60	116,835	32,413 31,457	1,167	5,681 6,849	5,000	13	27,152 25,785	23.2%	39 40	89,683 91,050
	Jan-26	0	404.1	5,285,755	495.67	1.61	1,60	116,835	31,131	-	7,177	5,000	86	24,040	20.6%	41	92,795
	Feb-26	19.2	320.8	5,477,524	495.84	1.51	1.60	116,835	32,171	11,257	5,447	5,000	84	38,065	32.5%	42	78,770
3	Mar-26 Apr-26	17.5 57.7	288.7 192.9	5,669,293 5,861,063	496,00 496,17	1.61	1.60	116,835 116,835	33,584 35,485	10,270 33,944	4,807 3,031	5,000	70 60	39,118 66,457	33.5% 56.9%	43 44	77,717 50,378
8	May-26	29.2	128.3	6,052,832	496.34	1.61	1.60	116,835	36,995	17,207	1,921	5,000	52	52,333	44.8%	45	64,502
1	Jun-26	20.7	90.5	6,244,602	496.52	1.54	1.60	116,835	37,735	12,228	1,316	5,000	49	48,696	41.7%	46	68,139
1	Jul-26 Aug-26	13.9 35.1	98.4 136	6,436,371 6,628,141	496.69 496.86	1.57	1.60	116,835	37,531 36,831	8,226 20,814	1,438 2,049	5,000	45 44	44,464 55,639	38.1% 47.5%	48	72,371 61,196
	Sep-26	0.6	198.9	6,819,910	497.02	1.61	1.60	116,835	35,351	357	3,122	5,000	38	32,624	27.9%	49	84,211
3	Oct-26 Nov-26	27.8	285.2 332.8	7,011,680	497.18	1.61	1.60	116,835 116,835	33,656 32,408	16,518 416	4,742 5,680	5,000	33 29	45,466 27,172	38.9%	50 51	71,369 89,663
5	Dec-26	2	388.8	7,395,219	497.50	1.61	1.60	116,835	31,453	1,186	6,849	5,000	24	25,814	22.1%	52	91,021
	Jan-27	0	404.1	7,586,988	497.66	1.61	1.60	116,835	31,128	- 180	7,176	5,000	22	23,974	20.5%	53	92,862
3	Feb-27 Mar-27	19.2 17.5	320.8 288.7	7,586,988 7,586,988	497.66 497.66		1.60			11,370 10,363	5,424 4,790	5,000	21 20	5,967 5,593	0.0%	54 55	
1	Apr-27	57.7	192.9	7,586,988	497.66	- 4	1.60	1	121	34,170	3,023	5,000	6	31,152	0.0%	56	(A)
	May-27	29.2	128,3	7,586,988	497.66	-	1.60	- 2	*	17,292	1,917	5,000	2	15,377	0.0%	57	
1 3	Jun-27 Jul-27	20.7 13.9	90.5 98.4	7,586,988 7,586,988	497.66 497.66		1,60	- 1		12,258 8,231	1,314 1,435	5,000	0	10,946 6,796	0.0%	58 59	-
	Aug-27	35.1	136	7,790,120	497.83	1.61	1.60	123,758	39,013	20,774	2,049	5,000	0	57,738	46.7%	60	66,019
5	Sep-27	0.6 27.8	198.9 285.2	7,993,252	498.00 498.17	1.61	1.61	123,758	37,447	355	3,123	5,000	0 16	34,679	28.0%	61 62	89,079 76,419
	Oct-27 Nov-27	0.7	332.8	8,196,384 8,399,517	498,34	1.61	1.61	123,758	35,652 34,330	16,414 413	4,743 5,682	5,000	18	47,339 29,079	23.5%	63	94,679
- 6	Dec-27	2	388.8	8,602,649	498.51	1.51	1.61	123,758	33,318	1,178	6,851	5,000	17	27,653	22.4%	64	96,095
3	Jan-28 Feb-28	19.2	404.1 320.8	8,805,781 9,008,913	498.68 498.85	1.61	1.61	123,758	32,974 34,077	11,295	7,179 5,448	5,000	16 12	25,812 39,935	20.9%	65 66	97,946 83,823
	Mar-28	17.5	288.7	9,212,045	498.83	1.51	1.61	123,758	35,574	10,272	4,808	5,000	11	41,049	33.2%	67	82,709
s	Apr-28	57.7	192.9	9,415,177	499.18	1.61	1.61	123,758	37,588	33,862	3,032	5,000	10	68,428	55.3%	68	55,330
3	May-28 Jun-28	29.2	128.3 90.5	9,618,310	499.35 499.52	1.61	1.61	123,758	39,188 39,971	17,131 12,149	1,921 1,316	5,000	10	54,407 50,815	44.0%	69 70	69,350 72,943
	Jul-28	13.9	98.4	10,024,574	499.69	1,55	1.60	123,758	39,861	8,156	1,438	5,000	10	46,589	37.6%	71	77,169
8	Aug-28	35.1	136	10,024,574	499.69	2	1.60		2 0	20,595	2,045	5,000	10	18,560	0.0%	72	
	Sep-28 Oct-28	0.6 27.8	198.9 285.2	10,024,574	499.69 499.69	-	1.60		-	352 16,312	3,114 4,726	5,000	10 5	11,591	0.0%	73	
- V	Nov-28	0.7	332.8	10,024,574	499.69	· ·	1,60	-	-	411	5,658	5,000	2	18	0.0%	75	
7	Dec-28	2 0	388.8	10,024,574	499.69	2 1	1.60	- 2	2 3	1,174	6,820	5,000	1	-	0.0%	76	
0	Jan-29 Feb-29	19.2	404.1 320.8	10,024,574	499.69 499.89	1.61	1.60	146,109	40,232	11,303	7,145 5,453	5,000	0	46,082	0.0% 31.5%	77 78	100,027
5	Mar-29	17.5	288.7	10,504,212	500.08	1.61	1.61	146,109	41,999	10,289	4,811	5,000	0	47,478	32.5%	79	98,631
1	Apr-29	57.7 29.2	192.9 128.3	10,744,032	500.27 500.47	1.61	1.61	146,109 146,109	44,376 46,265	33,942 17,171	3,033 1,922	5,000	5 105	75,289 61,619	51.5% 42.2%	80 81	70,820 84,490
1	May-29 Jun-29	20.7	90.5	11,223,670	500.47	1.48	1.60	146,109	46,265	12,178	1,316	5,000	190	58,242	39.9%	81	87,867
	Jul-29	13.9	98.4	11,463,489	500.89	1.49	1.60	146,109	47,060	8,175	1,439	5,000	153	53,949	36.9%	83	92,160
1	Aug-29 Sep-29	35.1 0.6	136 198.9	11,703,308	501.09 501.28	1.60	1.60	146,109 146,109	46,060 44,211	20,645 353	2,050 3,124	5,000	124 100	64,779 41,540	44.3% 28.4%	84 85	81,330 104,569
1	Oct-29	27.8	285.2	12,182,947	501.47	1.61	1.60	146,109	42,092	16,353	4,746	5,000	947	54,646	37.4%	86	91,464
- 19 - 19	Nov-29	0.7	332.8	12,422,766	501.66	1.61	1.60	146,109	40,531	412	5,686	5,000	1,180	36,437	24.9%	87	109,672
8	Dec-29 Jan-30	0	388.8 404.1	12,662,585	501.85 502.04	1.61	1.60	146,109	39,337 38,930	1,177	6,856 7 185	5,000	684 461	34,341 32,207	23.5%	88 89	111,768 113,902
1	Jan-30 Feb-30	19.2	320.8	12,902,405	502.04	- 1.61	1.61	140,109	38,930	11,297	7,185 5,424	5,000	355	6,227	0.0%	90	113,902
	Mar-30	17.5	288.7	12,902,405	502.04	(4)	1.61	- 24	9	10,296	4,790	5,000	289	5,795	0.0%	91	(4)
	Apr-30 May-30			12,902,405		= 1	1,61			33,948 17,180	3,023 1,917		17 5	30,942 15,267	0.0%	92 93	
1	Jun-30			12,902,405	502.04	-	1.61		1	17,180	1,314		2	10,868	0.0%	94	1
	Jul-30	13.9	98.4	12,902,405	502.04		1.61			8,178	1,436	5,000	1	6,744	0.0%	95	
	Aug-30 5ep-30			13,150,680 13,398,956	502.25 502.45	1,59 1,61		151,261 151,261	47,684 45,770	20,662 353	2,050 3,125	5,000	1	66,297 42,999	43.8%	96 97	84,964 108,262
	Oct-30			13,647,232	502.65	1.61	- W.	151,261	43,576	16,367	4,747	5,000	273	55,470	36.7%	98	95,792
1 to 1	Nov-30	0.7	332.8	13,895,508	502.85	1.61	1.61	151,261	41,960	412	5,687	5,000	228	36,914	24.4%	99	114,348
9	Dec-30	2	388.8	14,143,784	503.05	1.61	1.61	151,261	40,724	1,178	6,858	5,000	186	35,230	23.3%	100	116,031

TABLE 6.5: TSF CELL D WATER BALANCE - AVERAGE CONDITIONS

Year	Month	Rainfall	Evaporation	Cumulative	Tailings	Tailings	density	Water in Slurry	Supernatant Runoff	Rainfall Runoff	Evaporation	Pond Volume	Consolidation	Available	TSF	Discharge	Make Up
rear	Month		mm	Tonnage	Level	Layers	Mass	Siurry	Runon	Runon	Losses	volume	Volume -	Recycle	e		Requirement
		mm		t	RL m	t/m3	t/m3	mª	m ^a	mª	m*	mª	mª	mª.	96	m²	m ^a
	Jan-31	0	404.1	14,392,060	503.25	1.61	1.61	151,261	40,303	23	7,186	5,000	158	33,275	22.0%	101	117,987
	Feb-31	19.2	320.8	14,640,335	503.45	1.61	1.61	151,261	41,650	11,323	5,454	5,000	140	47,659	31.5%	102	103,603
	Mar-31	17.5	288.7	14,888,611	503.65	1.61	1.61	151,261	43,480	10,307	4,811	5,000	127	49,102	32.5%	103	102,159
	Apr-31	57.7	192.9	15,136,887	503.85	1.61	1.61	151,261	45,941	33,999	3,034	5,000	118	77,024	50.9%	104	74,237
	May-31	29.2	128.3	15,385,163	504.06	1.57	1.61	151,261	47,896	17,200	1,922	5,000	110	63,284	41.8%	105	87,977
	Jun-31	20.7	90.5	15,633,439	504.28	1.47	1.60	151,261	48,854	12,199	1,316	5,000	108	59,845	39.6%	106	91,416
	Jul-31	13.9	98.4	15,881,714	504.50	1.49	1.60	151,261	48,720	8,189	1,439	5,000	105	55,575	36.7%	107	95,686
	Aug-31	35.1	136	15,881,714	504.49		1.60	J 53 14		20,678	2,045	5,000	97	18,731	0.0%	108	
	5ep-31	0.6	198.9	15,881,714	504.49	- 1	1.60			353	3,114	5,000	91	-	0.0%	109	,
	Oct-31	27.8	285.2	15,881,714	504.49	i=1 /	1.60	[2]	121	16,378	4,725	5,000	30	11,681	0.0%	110	
	Nov-31	0.7	332.8	15,881,714	504.49	·	1.60	39	# 1	412	5,658	5,000	11		0.0%	111	9
10	Dec-31	2	388.8	15,881,714	504.49	+ /	1,60	25	F 1	1,178	6,820	5,000	4	**	0.0%	112	1.51
	Jan-32	0	404.1	15,881,714	504.49	- 4	1.60	2 2			7,145	5,000	2		0.0%	113	
	Feb-32	19.2	320.8	16,129,989	504.70	1.61	1.60	151,260	41,649	11,328	5,454	5,000	1	47,525	31,4%	114	103,736
	Mar-32	17.5	288,7	16,378,263	504.89	1.61	1.60	151,260	43,480	10,311	4,811	5,000	1	48,980	32.4%	115	102,281
	Apr-32	57.7	192.9	16,626,537	505.09	1.61	1.60	151,260	45,940	34,014	3,034	5,000	886	77,807	51.4%	116	73,454
11	May-32	29.2	128.3	16,874,811	505.29	1.57	1.60	151,260	47,896	17,207	1,922	5,000	1,431	64,612	42.7%	117	86,649

Year	Month	Rainfall	Evaporation	Cumulative Tonnage	Tailings Level	Tailings	density	Water in	Supernatant Runoff	Rainfall Runoff	Evaporation Losses	Pond Volume	Consolidation Volume	Available	TSF	Discharge	Make Up Requirement
real	Month	mm	mm	t	RLm	Layers t/m3	Mass t/m3	ma	m²	ma	m ^a	ma	m ^a	Recycle m ^a	96	m²	m ^a
	Aug-22	35,1	136	279,517	486.36	1.35	1.35	170,295	54,252	- ""	1,662	5,215		28,011	16.4%	- "	142,284
	Sep-22	0.6	198.9	559,033	487.24	1.41	1.38	170,295	51,858	*	3,195	5,000	185	20,406	12.0%	1	149,889
-	Oct-22 Nov-22	27.8	285.2 332.8	838,550 1,118,067	487.63 488.00	1.54 1.61	1.43	170,295 170,295	49,316 47,443	955	4,766 5,716	5,000	3598	24,024 15,786	14.1% 9.3%	3	146,271 154,509
1	Dec-22	2	388.8	1,397,584	488.36	1.61	1.51	170,295	45,998	-	6,882	5,000		14,357	8.4%	4	155,938
	Jan-23	0	404.1	1,677,100	488.72	1.61	1.53	170,295	45,468	1,956	7,206	5,000	1,529	13,829	8.1%	5	156,465
	Feb-23 Mar-23	19.2 17.5	320.8 288.7	1,956,617 2,236,134	489.05 489.31	1.61	1.55	170,295	46,935 48,975	3,875	5,457 4,810	5,000	798 493	22,227	13.1%	5 7	148,068 146,908
	Apr-23	57.7	192.9	2,515,651	489.57	1.61	1.57	170,295	51,737	1,326	3,021	5,000	340	41,320	24.3%	8	128,975
	May-23	29.2	128,3	2,795,167	489.84	1.50	1.56	170,295	53,932	25,639	1,925	5,000	207	33,364	19.6%	9	136,931
1	Jun-23 Jul-23	20.7	90.5 98.4	3,074,684	490.13 490.41	1,44	1.55	170,295	55,007 54,856	F 100	1,314	5,000	178 164	31,533 28,857	18.5%	10 11	138,762 141,438
	Aug-23	35,1	136	3,633,718	490,67	1.53	1.54	170,295	53,692	-	2,044	5,000	144	37,221	21,9%	12	133,074
	Sep-23	0.6	198.9	3,913,234	490,92	1.61	1.55	170,295	51,539	28	3,127	5,000	595	20,147	11.8%	13	150,148
	Oct-23 Nov-23	27.8	285.2 332.8	4,192,751 4,472,268	491.15 491.41	1.61	1.56	170,295	49,072 47,254	1,370	4,740 5,693	5,000	1,849 2,157	30,260 15,916	17.8% 9.3%	14 15	140,035 154,379
2	Dec-23	2	388.8	4,751,784	491.65	1.61	1.57	170,295	45,863	1,370	6,863	5,000	1,199	14,725	8.6%	16	155,570
9	Jan-24	. 0	404.1	5,031,301	491.89	1.61	1.58	170,295	45,388	2,294	7,195	5,000	824	13,142	7.7%	17	157,153
9	Feb-24 Mar-24	19.2	320.8 288.7	5,310,818 5,590,335	492.14 492.38	1.61	1.58	170,295	46,902 48,959	4,203	5,454 4,808	5,000	621 483	25,848 26,615	15.2% 15.6%	18 19	144,447
	Apr-24	57.7	192.9	5,869,851	492.63	1.61	1.58	170,295	51,726	1,388	3,020	5,000	509	51,677	30.3%	20	118,618
	May-24	29.2	128.3	6,149,368	492.89	1.52	1.58	170,295	53,926	26,772	1,926	5,000	433	38,173	22.4%	21	132,122
	Jun-24 Jul-24	13.9	90.5 98.4	6,708,402	493.15 493.42	1.45	1.58	170,295 170,295	55,003 54,852		1,314 1,437	5,000	389 358	34,619 30,699	20,3%	22	135,676 139,596
	Aug-24	35.1	136	6,708,402	493.42	1.40	1.57	170,233	34,032		2,038	5,000	31.5	17,425	0.0%	24	139,336
	Sep-24	0.6	198.9	6,708,402	493.42	-	1.57	2 , 3	-		3,113	5,000	273	-	0.0%	25	-
	Oct-24	27.8	285.2	6,708,402	493.42	- 1	1.57			1 300	4,715	5,000	50	10,674	0.0%	26	(F)
3	Nov-24 Dec-24	0.7	332.8 388.8	6,708,402 6,708,402	493.42 493.42		1.57	39	-	1,398	5,659 6,819	5,000	10	-	0.0%	27 28	9
	Jan-25	0	404.1	6,708,402	493.42		1.57			2,331	7,147	5,000	0		0.0%	29	
	Feb-25	19.2	320.8	6,909,587	493,59	1.61	1,57	122,572	33,754	4,249	5,444	5,000	0	37,465	30.6%	30	85,107
	Mar-25 Apr-25	17.5 57.7	288.7 192.9	7,110,773 7,311,959	493.75 493.93	1.61	1.58	122,572	35,236 37,229	1,401	4,801 3,017	5,000	947	38,605 65,035	31.5% 53.1%	31	83,966 57,537
	May-25	29.2	128.3	7,513,144	494.10	1.61	1.58	122,572	38,813	26,996	1,924	5,000	1,948	51,652	42.1%	33	70,920
	Jun-25	20.7	90.5	7,714,330	494.28	1.52	1.58	122,572	39,588		1,313	5,000	1,282	48,243	39.4%	34	74,329
	Jul-25 Aug-25	13.9 35.1	98.4 136	7,915,516 8,116,701	494.45 494.62	1.55	1.58	122,572	39,479 38,641	94 95	1,436 2,042	5,000	959 784	44,178 55,018	36.0% 44.9%	35 36	78,394 67,554
	Sep-25	0.6	198.9	8,317,887	494.79	1.61	1.58	122,572	37,089	-	3,123	5,000	625	32,597	26.6%	37	89,975
	Oct-25	27.8	285.2	8,519,073	494.96	1.61	1.59	122,572	35,311	- 2	4,732	5,000	499	45,081	36.8%	38	77,491
4	Nov-25 Dec-25	0.7	332.8 388.8	8,720,259 8,921,444	495.13 495.30	1.61	1,59	122,572	34,001 32,999	1,417	5,683 6,849	5,000	452 488	27,152 25,785	22.2%	39 40	95,420 96,787
	Jan-26	0	404.1	9,122,630	495.47	1.61	1,59	122,572	32,657	2,368	7,180	5,000	406	24,040	19.6%	41	98,532
	Feb-26		320.8	9,323,816	495.53	1.51	1.59	122,572	33,749	4,333	5,443	5,000	355	38,065	31.1%	42	84,507
	Mar-26 Apr-26	17.5 57.7	288.7 192.9	9,525,001	495.80 495.97	1.61	1.59 1.59	122,572	35,232 37,226	1,428	4,800 3,017	5,000	318 289	39,118 66,457	31,9% 54.2%	43	83,454 56,115
8	May-26	29.2	128.3	9,927,373	496.14	1.61	1.59	122,572	38,811	27,517	1,924	5,000	266	52,333	42.7%	45	70,239
	Jun-26	20.7	90.5		496,31	1.53	1.59	122,572	39,587		1,313	5,000	262	48,696	39.7%	46	73,876
	Jul-26	The second second	98.4	10,329,744	496.49	1.56	1.59 1.59	122,572	39,478	- B)	1,435	5,000	256	44,464	36.3% 45.4%	47	78,108
	Aug-26 Sep-26	35.1	136 198.9	10,530,930	496.66 496.82	1.61	1.59	122,572	38,639 37,086		2,042 3,122	5,000	252 234	55,639 32,624	26.6%	48 49	66,933 89,948
	Oct-26	27.8	285.2	10,933,301	496.99	1.51	1.59	122,572	35,308	51.0	4,732	5,000	212	45,466	37.1%	50	77,106
	Nov-26		332.8	11,134,487	497.15	1.61	1.59	122,572	33,997	1,442	5,682	5,000	190	27,172	22.2%	51	95,400
5	Dec-26 Jan-27	0	388.8 404.1	11,335,673 11,536,858	497.31	1.61	1.59	122,572	32,995 32,653	2,402	6,849 7,180	5,000	158 153	25,814 23,974	21.1% 19.5%	52 53	96,758 98,598
	Feb-27	19.2	320.8	11,536,858	497.47	-	1.60		-	4,383	5,419	5,000	211	5,967	0.0%	54	
- 8	Mar-27	17.5 57.7	288.7	11,536,858	497.47	-	1.60	-		1.667	4,783	5,000	174 45	5,593	0.0%	55 56	
1	Apr-27 May-27	29.2	192.9 128.3	11,536,858	497.47	-	1.60		-	1,441 27,739	3,008 1,921	5,000	19	31,152 15,377	0.0%	57	
	Jun-27	20.7	90.5		497.47	1-1	1,60	2 2	-	-	1,311	5,000	9	10,946	0.0%	58	-
	Jul-27	13.9 35.1	98.4	11,536,858	497.47	1.61	1.60	126 700	20.000	21	1,434	5,000	5 2	6,796	0.0% 45.5%	59	50 0F1
9	Aug-27 Sep-27	0.6	136 198.9		497.64	1.61	1.60	126,790 126,790	39,968 38,362	- 2	2,042 3,123	5,000	1	57,738 34,679	27.4%	60	69,051 92,111
i i	Oct-27	27.8	285.2	12,161,184	497.98	1.61	1.60	126,790	36,523	-	4,732	5,000	136	47,339	37.3%	62	79,450
	Nov-27	0.7	332.8	12,369,293	498.15	1.61	1.60	126,790	35,167	1,441	5,683	5,000	144	29,079	22.9%	63	97,711
- 6	Dec-27 Jan-28	0	388.8 404.1	12,577,401	498.31 498.48	1.61	1.60	126,790 126,790	34,130 33,777	2,400	6,850 7,181	5,000	122 115	27,653 25,812	21.8%	65	99,127 100,978
	Feb-28	19.2	320.8	12,993,618	498.65	1.61	1.60	126,790	34,907	4,384	5,443	5,000	109	39,935	31.5%	66	86,855
	Mar-28	17.5 57.7	288.7 192.9	13,201,727	498.82	1.51	1.60	126,790	36,442	2.440	4,801	5,000	96	41,049	32.4% 54.0%	67	85,740
1	Apr-28 May-28	29.2	192.9	13,409,835	498.99 499.15	1.61	1.60	126,790 126,790	38,506 40,146	1,440 27,701	3,017 1,924	5,000	91 86	68,428 54,407	42.9%	68 69	58,362 72,382
1	Jun-28	20.7	90.5	13,826,052	499.33	1.52	1.60	126,790	40,949		1,313	5,000	87	50,815	40.1%	70	75,975
	Jul-28	13.9	98.4	14,034,161	499.50	1,55	1.60	126,790	40,836	-	1,436	5,000	86	46,589	36.7%	71	80,201
	Aug-28 Sep-28	35.1 0.6	136 198.9	AND DESCRIPTION OF THE PARTY OF	499.50 499.50	-	1.60	14			2,038 3,113	5,000	85 83	18,560	0.0%	72	-
	Oct-28	27.8	285.2	14,034,161	499.50	*	1.60	*		-	4,715	5,000	39	11,591	0.0%	74	8
	Nov-28	0.7	332.8	14,034,161	499.50		1,60	-		1,438	5,659	5,000	20		0.0%	75	
7	Dec-28 Jan-29	0	388.8 404.1	14,034,161	499.50 499.50		1.60	- 4	-	2,397	6,819 7,148	5,000	11 6	- 2	0.0%	76	-
	Feb-29	19.2	320.8	14,278,335	499.69	1.61	1.60	148,763	40,957	4,388	5,448	5,000	3	46,082	31.0%	78	102,680
	Mar-29	17.5	288.7	14,522,509	499.88	1.61	1.60	148,763	42,758	1 1	4,804	5,000	2	47,478	31.9%	79	101,285
	Apr-29 May-29	57.7 29.2	192.9 128.3	14,766,683	500.08 500.27	1.61	1.60	148,763 148,763	45,179 47,103	1,441 27,712	3,018 1,925	5,000	42 167	75,289 61,619	50.6%	80 81	73,473 87,143
1	Jun-29	20.7	90.5	15,255,032	500.48	1.48	1.60	148,763	48,046	ZIJIAZ.	1,313	5,000	250	58,242	39.2%	82	90,521
	Jul-29	13.9	98.4	15,499,206	500.69	1.50	1.60	148,763	47,913	80	1,437	5,000	205	53,949	36.3%	83	94,813
	Aug-29 Sep-29	35.1	136 198.9	15,743,380 15,987,555	500.89	1.60	1.60	148,763 148,763	46,895 45,011		2,043 3,124	5,000	179 155	64,779 41,540	43.5%	84	83,983 107,223
	Oct-29		285.2	16,231,729	501.27	1.61	1.60	148,763	42,853	+	4,736	5,000	1,012	54,646	36.7%	86	94,117
2	Nov-29	0.7	332.8	16,475,903	501.46	1.61	1.60	148,763	41,263	1,438	5,687	5,000	1,228	36,437	24.5%	87	112,326
8	Dec-29 Jan-30	0	388.8 404.1	16,720,077	501.65 501.84	1.61	1.60	148,763	40,047 39,632	2,394	6,855 7,187	5,000	739 510	34,341 32,207	23.1%	88 89	114,421 116,556
1	Feb-30	19.2	320.8	16,964,251	501.84	1.01	1.60		39,032	4,370	5,419	5,000	400	6,227	0.0%	90	110,330
	Mar-30		288.7	16,964,251	501.84	(4)	1.60	24	9		4,783	5,000	331	5,795	0.0%	91	9
	Apr-30 May-30	-	1,10,000,000	16,964,251 16,964,251	501.84 501.84	= 1	1.60	8		1,437 27,655	3,008 1,920	5,000	42 22	30,942 15,267	0.0%	92	
	Jun-30			16,964,251	501.84	-	1.60	- 4	1	27,033	1,311	5,000	14	10,868	0.0%	94	
	Jul-30	13.9	98.4	16,964,251	501.84		1.60		-	9	1,434	5,000	9	6,744	0.0%	95	
	Aug-30 5ep-30			17,215,707	502.04 502.24	1,59		153,199 153,199	48,293 46,354		2,043 3,125	5,000	6	66,297 42,999	43.3%	96 97	85,901 110,200
	Oct-30		THE PERSON NAMED IN COLUMN 1	17,718,618	502.44	1.61	- North Children	153,199	44,131	- 3	4,736	5,000	293	55,470	36.2%	98	97,729
	Nov-30	0.7	332.8	17,970,073	502.64	1.61	1.60	153,199	42,494	1,436	5,688	5,000	255	36,914	24.1%	99	116,285
9	Dec-30	2	388.8	18,221,528	502.84	1.61	1.60	153,199	41,241		6,856	5,000	215	35,230	23.0%	100	117,968

TABLE 6.6: TSF CELL C WATER BALANCE - DRY CONDITIONS

Year	Month	Rainfall	Evaporation	Cumulative	Tailings Level	Tailings (density	Water in Slurry	Supernatant Runoff	Rainfall Runoff	Evaporation	Pond Volume	Consolidation Volume	Available	TSF	Discharge	Make Up
Tear	Month		501	Tonnage	Level	Layers	Mass	Siurry	Kullon	Aution	Losses	votume	volume	Recycle	8	52.0	Requirement
		mm	mm	t	RLm	t/m3	t/m3	mª	m²	m ^a	m*	mª	m ^a	mª	96	m²	m ^a
	Jan-31	. 0	404.1	18,472,984	503.04	1.51	1.60	153,199	40,815	2,391	7,188	5,000	189	33,275	21.7%	101	119,924
	Feb-31	19.2	320.8	18,724,439	503.24	1.61	1.60	153,199	42,180	4,369	5,449	5,000	171	47,659	31.1%	102	105,540
	Mar-31	17.5	288.7	18,975,894	503.44	1.61	1.60	153,199	44,034		4,805	5,000	152	49,102	32.1%	103	104,097
	Apr-31	57.7	192.9	19,227,350	503.65	1.61	1.60	153,199	46,527	1,434	3,019	5,000	143	77,024	50.3%	104	76,174
	May-31	29.2	128.3	19,478,805	503.85	1.57	1.60	153,199	48,509	27,592	1,925	5,000	136	63,284	41.3%	105	89,914
	Jun-31	20.7	90.5	19,730,260	504.07	1.47	1.60	153,199	49,479	-	1,313	5,000	121	59,845	39.1%	106	93,354
	Jul-31	13.9	98.4	19,981,716	504.29	1.49	1.60	153,199	49,343	- 83	1,437	5,000	119	55,575	36.3%	107	97,623
	Aug-31	35.1	136	19,981,716	504.29		1.60	S 55 1-		F2)	2,038	5,000	106	18,731	0.0%	108	
	Sep-31	0.6	198.9	19,981,716	504.29	- 1	1.60			-	3,113	5,000	99	-	0.0%	109	-
	Oct-31	27.8	285.2	19,981,716	504.29	F	1.60	. 64	121		4,715	5,000	35	11,681	0.0%	110	. 3
	Nov-31	0.7	332.8	19,981,716	504.29	E	1.60	39	9	1,433	5,659	5,000	14		0.0%	111	36
10	Dec-31	2	388.8	19,981,716	504.29	+ 7	1,60	25	7 8 7		6,819	5,000	7	*	0.0%	112	
	Jan-32	0	404.1	19,981,716	504.29	- 4	1.60	2 2	0.0	2,388	7,147	5,000	4		0.0%	113	
	Feb-32	19.2	320.8	20,232,235	504.49	1.61	1.60	152,628	42,023	4,362	5,449	5,000	2	47,525	31.1%	114	105,104
	Mar-32	17.5	288,7	20,482,754	504.68	1.61	1.60	152,628	43,870	-	4,805	5,000	2	48,980	32.1%	115	103,648
	Apr-32	57.7	192.9	20,733,273	504.88	1.61	1.60	152,628	46,354	1,432	3,019	5,000	896	77,807	51.0%	116	74,822
11	May-32	29.2	128,3	20,983,791	505.08	1.57	1.60	152,628	48,328	27,546	1,925	5,000	1,452	64,612	42.3%	117	88,016

Year	Month	Rainfall	Evaporation	Cumulative Tonnage	Tailings Level	Tailings		Water in	Supernatant Runoff	Rainfall Runoff	Evaporation Losses	Pond Volume	Consolidation Volume	Available		Discharge	Make Up Requirement
		mm	mm	t	RLm	Layers t/m3	Mass t/m3	mª.	m _s	m ^a	m*	m*	m ⁴	Recycle m*	%	m²	m*
	Aug-22	35,1	135	124,355	488.27	1.35	1.35	75,763	24,442	- 1	1,723	5,155	727	22,565	29.8%	- 21	53,198
- 0	Sep-22 Oct-22	0.6 27.8	198.9 285.2	248,710 373,065	489.18 489.55	1.42	1.39	75,763 75,763	23,271	*	3,177 4,759	5,000	(4)	20,249 17,381	26.7%	1 2	55,514 58,382
	Nov-22	0.7	332.8	497,420	489.91	1.59	1.48	75,763	21,310	630	5,709	5,000		16,231	21.4%	3	59,532
1	Dec-22	2	388.8	621,775	490.27	1.60	1.51	75,763	20,663	- 2	6,874	5,000	1991	13,789	18.2%	4	61,974
- 5	Jan-23 Feb-23	19.2	404.1 320.8	746,130 870,485	490.62 490.97	1.60	1.53	75,763 75,763	20,410	1,226 2,403	7,199 5,453	5,000	516 342	15,054 18,315	19.9%	5	60,709 57,448
	Mar-23	17.5	288.7	994,840	491.15	1.60	1.55	75,763	21,888	2,403	4,805	5,000	212	17,295	22.8%	7	58,468
	Apr-23	57.7	192.9	1,119,195	491.32	1.60	1.56	75,763	23,076	879	3,018	5,000	154	21,091	27.8%	8	54,672
- 69	May-23 Jun-23	29.2	128.3 90.5	1,243,550 1,367,906	491.49 491.67	1.50	1.57	75,763 75,763	24,026	17,702	1,925	5,000	71 55	39,874 23,234	52.6% 30.7%	9	35,889 52,529
- 1	Jul-23	13.9	98.4	1,492,261	491.85	1,56	1,56	75,763	24,422		1,436	5,000	50	23,036	30.4%	11	52,727
	Aug-23	35,1	136	1,616,616	492.02	1.61	1.57	75,763	23,907	-	2,042	5,000	47	21,913	28.9%	12	53,850
1	Sep-23 Oct-23	27.8	198.9 285.2	1,740,971 1,865,326	492.18 492.35	1.61	1.57 1.58	75,763 75,763	22,955 21,860	- S	3,121	5,000	42 67	19,875 17,198	26.2%	13 14	55,888 58,565
	Nov-23	0.7	332.8	1,989,681	492.52	1.61	1.58	75,763	21,050	1,170	4,729 5,677	5,000	201	16,744	22.1%	15	59,020
2	Dec-23	2	388.8	2,114,036	492.69	1.61	1.58	75,763	20,427	-	6,842	5,000	132	13,717	18.1%	16	62,046
9	Jan-24 Feb-24	19.2	404.1 320.8	2,238,391	492.85 493.01	1.61	1.58	75,763	20,210	2,086	7,171	5,000	102 83	15,226	20.1%	17	50,537
9	Mar-24	17.5	288.7	2,362,746	493.01	1.61	1.59	75,763 75,763	21,788	3,928	5,436 4,795	5,000	71	19,451 17,063	22,5%	19	56,312 58,700
	Apr-24	57.7	192.9	2,611,456	493.24	1.61	1.59	75,763	23,017	1,316	3,014	5,000	59	21,378	28,2%	20	54,385
	May-24	29.2	128.3	2,735,811	493.35	1.61	1.59	75,763	23,994	25,407	1,923	5,000	40	47,518	62.7%	21	28,245
	Jun-24 Jul-24	20.7	90.5 98.4	2,860,166 2,984,521	493.47 493.58	1.61	1.59 1.59	75,763 75,763	24,472	-	1,312 1,435	5,000	35 35	23,196 23,005	30.5%	22	52,567 52,758
	Aug-24	35.1	136	2,984,521	493.58	-	1.59		-		2,038	5,000	34		0.0%	24	
	Sep-24	0.6	198.9	2,984,521	493.58		1.59				3,113	5,000	32		0.0%	25	
1	Oct-24 Nov-24	27.8	285.2 332.8	2,984,521 2,984,521	493.58	-	1.59	. 14	-	1,329	4,715 5,659	5,000	7 2	-	0.0%	26 27	
3	Dec-24	2	388.8	2,984,521	493.58	+ 7	1,59	2	-	1,525	5,819	5,000	1	+	0.0%	28	-
	Jan-25	10.0	404.1	2,984,521	493.58	- 1.61	1.59	440.000		2,215	7,148	5,000	0	20.020	0.0%	29	05.015
- 1	Feb-25 Mar-25	19.2 17.5	320.8 288.7	3,176,291 3,368,060	493.76 493.93	1.61	1.59	116,835 116,835	32,184 33,594	4,079	5,443 4,801	5,000	0	30,820 28,793	26.4%	30 31	86,015 88,042
	Apr-25	57.7	192.9	3,559,829	494.11	1.61	1.60	116,835	35,491	1,354	3,017	5,000	21	33,850	29.0%	32	82,986
	May-25	29.2	128,3	3,751,599	494.28	1.61	1.60	116,835	36,999	26,192	1,924	5,000	20	61,286	52.5%	33	55,549
1	Jun-25 Jul-25	20.7 13.9	90.5 98.4	3,943,368 4,135,138	494.47 494.65	1.53	1.59	116,835	37,737 37,633		1,313	5,000	20 19	36,444 36,216	31.2%	34 35	80,391 80,619
- 6	Aug-25	35.1	136	4,326,907	494.82	1.61	1.59	116,835	36,834	- 80	2,042	5,000	18	34,810	29.8%	36	82,025
	Sep-25	0.6	198.9	4,518,677	495.00	1.61	1.60	116,835	35,355	-	3,122	5,000	17	32,249	27.6%	37	84,586
	Oct-25 Nov-25	27.8	285.2 332.8	4,710,446	495.16 495.33	1.61	1.60	116,835 116,835	33,661 32,413	1,397	4,732 5,682	5,000	15 13	28,944 28,141	24.8%	38	87,891 88,694
4	Dec-25	2	388.8	5,093,985	495.50	1.61	1.60	116,835	31,457	1,334	6,848	5,000	11	24,620	21.1%	40	92,215
	Jan-26	0	404.1	5,285,755	495.67	1.61	1,60	116,835	31,131	2,338	7,179	5,000	86	26,376	22.6%	41	90,459
	Feb-26 Mar-26	19.2 17.5	320.8 288.7	5,477,524 5,669,293	495.84 496.00	1.61	1.60	116,835	32,171 33,584	4,280	5,442 4,800	5,000	70	31,093 28,854	26.5%	42	85,742 87,981
3	Apr-26	57.7	192.9	5,861,063	496.17	1.61	1.60	116,835	35,485	1,412	3,016	5,000	60	33,940	29.0%	44	82,895
	May-26	29.2	128.3	6,052,832	496.34	1.61	1.60	116,835	36,995	27,225	1,924	5,000	52	62,348	53.4%	45	54,487
	Jun-26	20.7	90.5	6,244,602	496.52	1.54	1.60	116,835	37,735		1,313	5,000	49	36,471	31.2%	46	80,364
	Jul-26 Aug-26	13.9 35.1	98.4 136	6,436,371 6,628,141	496.69 496.86	1.57	1.60	116,835	37,531 36,831		1,435 2,042	5,000	45 44	36,241 34,832	31.0%	47	80,594 82,003
	Sep-26	0.6	198.9	6,819,910	497.02	1.61	1.60	116,835	35,351		3,122	5,000	38	32,268	27.6%	49	84,567
	Oct-26	27.8	285.2	7,011,680	497.18	1.61	1.60	116,835	33,656	1 490	4,731	5,000	33	28,959	24.8%	50	87,877
5	Nov-26 Dec-26	0.7	332.8 388.8	7,203,449 7,395,219	497.34	1.61	1.60	116,835	32,408 31,453	1,425	5,681 6,848	5,000	29	28,181	24.1%	51 52	88,654 92,206
	Jan-27	0	404.1	7,586,988	497.66	1.61	1.60	116,835	31,128	2,369	7,179	5,000	22	26,340	22,5%	53	90,495
	Feb-27	19.2	320.8	7,586,988	497.66		1.60		-	4,323	5,419	5,000	21		0.0%	54	-
1	Mar-27 Apr-27	17.5 57.7	288.7 192.9	7,586,988 7,586,988	497.66 497.66	-	1.60	-	-	1,421	4,783 3,008	5,000	20 5	-	0.0%	55 56	-
	May-27	29.2	128,3	7,586,988	497.66	*	1.60	- 38	-	27,359	1,921	5,000	2	25,441	0.0%	57	
	Jun-27	20.7	90.5	7,586,988	497.66	-	1,60	-		-	1,311	5,000	1		0,0%	58	-
1	Jul-27 Aug-27	13.9 35.1	98.4 136	7,586,988	497.55	1.61	1.60	123,758	39,013	2	1,434 2,042	5,000	0	36,971	0.0%	59 60	86,786
	Sep-27	0.6	198.9	7,993,252	498.00	1.61	1.61	123,758	37,447	*	3,123	5,000	0	34,324	27.7%	61	89,433
- 0	Oct-27	27.8	285.2	8,196,384	498.17	1.61	1.61	123,758	35,652		4,732	5,000	16 18	30,936	25.0%	62	92,822
6	Nov-27 Dec-27	0.7	332.8 388.8	8,399,517 8,602,649	498,34 498,51	1.61	1.61	123,758	34,330 33,318	1,416	5,683 6,850	5,000	17	30,081 26,486	24.3%	63 64	93,677 97,272
-	Jan-28	0	404.1	8,805,781	498.68	1.61	1.61	123,758	32,974	2,353	7,181	5,000	16	28,163	22.8%	65	95,595
	Feb-28	19.2	320.8	9,008,913	498.85	1.61	1.61	123,758	34,077	4,295	5,443	5,000	12	32,939	26.6%	66	90,818
	Mar-28 Apr-28	17.5 57.7	288.7 192.9	9,212,045	499.02 499.18	1.61	1.61	123,758	35,574 37,588	1,408	4,801 3,017	5,000	11	30,784 35,990	24.9%	68	92,973 87,768
	May-28	29.2	128.3	9,618,310	499.35	1.61	1.61	123,758	39,188	27,105	1,924	5,000	10	64,378	52.0%	69	59,380
	Jun-28	20.7	90.5 98.4	9,821,442	499.52	1.52	1.61	123,758	39,971	-	1,313	5,000	10 10	38,668	31.2%	70 71	85,090 85,333
	Jul-28 Aug-28	35,1	136	10,024,574	499.69 499.69	1,55	1.60	123,758	39,861	-	1,436 2,038	5,000	10	38,435	31.1%	72	85,323
	Sep-28	0.6	198.9	10,024,574	499.69		1.60	19	ų.		3,113	5,000	10	- £1	0.0%	73	
	Oct-28	27.8	285.2	10,024,574	499.69	* 1	1.60	- 88		1 409	4,715	5,000	5	±2	0.0%	74	
7	Nov-28 Dec-28	0.7	332.8 388.8	10,024,574	499.69 499.69	-	1.60		- 3	1,408	5,659 6,819	5,000	2		0.0%	75 76	
- 0	Jan-29	0	404.1	10,024,574	499.69	2	1.60	[<u>8</u>]		2,347	7,148	5,000	0	21	0.0%	77	- 2
	Feb-29	19.2	320.8	10,264,393	499.89	1.61	1.60	146,109	40,232	4,298	5,448	5,000	0	39,082	26.7%	78	107,028
	Mar-29 Apr-29	17.5 57.7	288.7 192.9	10,504,212	500.08 500.27	1.61	1.61	146,109	41,999 44,376	1,412	4,804 3,018	5,000	5	37,195 42,774	25.5%	79 80	108,914
	May-29	29.2	128.3	10,983,851	500.47	1.58	1.61	146,109	46,265	27,167	1,925	5,000	105	71,613	49.0%	81	74,497
	Jun-29	20.7	90.5	11,223,670	500.68	1.48	1.60	146,109	47,190	-	1,313	5,000	190	46,066	31.5%	82	100,043
	Jul-29 Aug-29	13.9 35.1	98.4 136	11,463,489	500.89 501.09	1.49	1.60	146,109	47,060 46,060	80	1,437 2,043	5,000	153 124	45,776 44,142	31.3%	83 84	100,333 101,968
	Sep-29	0.6	198.9	11,943,128	501.28	1.61	1.60	146,109	44,211	- 1	3,124	5,000	100	41,187	28.2%	85	104,922
	Oct-29	27.8	285.2	12,182,947	501.47	1.61	1.60	146,109	42,092		4,736	5,000	947	38,303	26.2%	86	107,806
8	Nov-29 Dec-29	0.7	332.8 388.8	12,422,766	501.66 501.85	1.61	1.60	146,109	40,531 39,337	1,412	5,687 6,855	5,000	1,180 684	37,437 33,166	25.6%	87 88	108,673 112,944
	Jan-30	0	404.1	12,902,405	502.04	1.61	1.60	146,109	38,930	2,353	7,187	5,000	461	34,558	23.7%	89	111,551
	Feb-30	19.2	320.8	12,902,405	502.04	-	1.61	-	77	4,295	5,419	5,000	355	-	0.0%	90	-
	Mar-30	17.5	288.7	12,902,405	502.04	+	1.61	34	-	1.612	4,783 3,008	5,000	289		0.0%	91	(4)
	Apr-30 May-30		100000000000000000000000000000000000000	12,902,405	502.04	-	1.61			1,412 27,182	3,008 1,920	5,000	17	25,266	0.0%	92	-
8	Jun-30	20.7	90.5	12,902,405	502,04		1.61	[#]	1	54	1,311	5,000	2	2	0.0%	94	-
5	Jul-30			12,902,405	502.04 502.25	1,59	1.61	151,261	47,684	91	1,434	5,000	1 1	45 642	0.0%	95 96	105.619
	Aug-30 5ep-30			13,150,680	502.45	1.59		151,261	45,770	# # P	2,043 3,125	5,000	1	45,642 42,646	28.2%	96	105,619 108,616
	Oct-30	27.8	285.2	13,647,232	502.65	1.61	1.61	151,261	43,576	29]	4,736	5,000	273	39,114	25.9%	98	112,148
	Nov-30			13,895,508	502.85	1.61		151,261	41,960	1,414	5,688	5,000	228	37,914	25.1%		113,347
9	Dec-30	2	388.8	14,143,784	503.05	1.61	1,61	151,261	40,724		6,857	5,000	185	34,054	22.5%	100	117,208

TABLE 6.7: TSF CELL D WATER BALANCE - DRY CONDITIONS

Year	Month	Rainfall	Evaporation	Cumulative	Tailings Level	Tailings	density	Water in Slurry	Supernatant Runoff	Rainfall Runoff	Evaporation	Pond Volume	Consolidation Volume	Available	TSF	Discharge	Make Up
rear	Month		101	Tonnage	rever	Layers	Mass	Siurry	Kullon	Runon	Losses	volume	volume	Recycle	28	1220	Requirement
		mm	mm	t	RLm	t/m3	t/m3	mª	m²	m ^a	m*	mª	m ^a	m ^a	%	m²	m ^a
	Jan-31	0	404.1	14,392,060	503.25	1.61	1.61	151,261	40,303	2,355	7,188	5,000	158	35,628	23.6%	101	115,633
	Feb-31	19.2	320.8	14,640,335	503.45	1.61	1.61	151,261	41,650	4,305	5,449	5,000	140	40,646	26.9%	102	110,615
	Mar-31	17.5	288.7	14,888,611	503.65	1.61	1.61	151,261	43,480		4,805	5,000	127	38,802	25.7%	103	112,459
	Apr-31	57.7	192.9	15,136,887	503.85	1.61	1.61	151,261	45,941	1,414	3,019	5,000	118	44,454	29.4%	104	106,808
	May-31	29.2	128.3	15,385,163	504.06	1.57	1.61	151,261	47,896	27,213	1,925	5,000	110	73,295	48.5%	105	77,967
	Jun-31	20.7	90.5	15,633,439	504.28	1.47	1.60	151,261	48,854		1,313	5,000	108	47,649	31.5%	106	103,613
	Jul-31	13.9	98.4	15,881,714	504.50	1.49	1.60	151,261	48,720	- 83	1,437	5,000	105	47,388	31.3%	107	103,873
	Aug-31	35.1	136	15,881,714	504.49	- a	1.60	J 55 L	- 6		2,038	5,000	97		0.0%	108	
	Sep-31	0.6	198.9	15,881,714	504.49	- 1	1.60	-		-	3,113	5,000	91	-	0.0%	109	-
	Oct-31	27.8	285.2	15,881,714	504.49	i÷: //	1.60	. 9	121	-	4,715	5,000	30	# .	0.0%	110	. 9
	Nov-31	0.7	332.8	15,881,714	504.49	× 1	1.60	39	- 4	1,414	5,659	5,000	11	95	0.0%	111	9
10	Dec-31	2	388.8	15,881,714	504.49	+ //	1,60	E# 1	T = 1		6,819	5,000	4	**	0.0%	112	
	Jan-32	0	404.1	15,881,714	504.49	- 4	1.60	2 8	0.0	2,356	7,147	5,000	2		0.0%	113	
	Feb-32	19.2	320.8	16,129,989	504.70	1.61	1.60	151,260	41,649	4,307	5,449	5,000	1	40,509	26.8%	114	110,752
	Mar-32	17.5	288,7	16,378,263	504.89	1.61	1.60	151,260	43,480		4,805	5,000	1	38,676	25.6%	115	112,585
	Apr-32	57.7	192.9	16,626,537	505.09	1.61	1.60	151,260	45,940	1,415	3,019	5,000	886	45,222	29.9%	116	106,038
11	May-32	29.2	128,3	16,874,811	505.29	1.57	1.60	151,260	47,896	27,225	1,925	5,000	1,431	74,627	49.3%	117	76,634

Year	Month	Rainfall	Evaporation	Cumulative Tonnage	Tailings Level	Tailings		Water in	Supernatant Runoff	Rainfall Runoff	Evaporation Losses	Pond Volume	Consolidation Volume	Available		Discharge	Make Up Requirement
	10 10 000 100	mm	mm	t	RL m	Layers t/m3	Mass t/m3	mª	mª	m²	m*	mª	m ⁴	Recycle m ^a	%	m²	m ^a
	Aug-22	35.1	136	279,517	486.36	1.35	1.35	170,295	54,252	1,554	1,663	5,224	120	53,920	31.7%		116,375
9	Sep-22 Oct-22	27.8	198.9 285.2	559,033 838,550	487.24 487.63	1.41	1.38	170,295	51,858 49,316	10,457	3,196 4,776	5,000	193	48,886 54,996	28.7% 32.3%	1 2	121,409 115,299
	Nov-22	0.7	332.8	1,118,067	488.00	1.51	1.47	170,295	47,443	1,552	5,717	5,000	9	43,279	25.4%	3	127,016
1	Dec-22 Jan-23	0	388.8 404.1	1,397,584	488.36 488.72	1.61	1.51	170,295	45,998 45,468	30,632 49,492	6,915 7,255	5,000	1,529	69,715 89,234	40.9% 52.4%	5	100,580 81,061
9	Feb-23	19.2	320.8	1,956,617	489.06	1.61	1.55	170,295	46,935	34,928	5,481	5,000	798	77,179	45.3%	6	93,115
	Mar-23	17.5	288.7	2,236,134	489.31	1.61	1.56	170,295	48,975	95,226	4,876	5,000	493	139,819	82.1%	7	30,476
1	Apr-23 May-23	57.7 29.2	192.9 128.3	2,515,651 2,795,167	489.57 489.84	1.61	1.57 1.56	170,295 170,295	51,737 53,932	1,326 25,195	3,021 1,925	5,000	340 207	50,381 77,408	29.6% 45.5%	8	119,913 92,887
	Jun-23	20.7	90.5	3,074,684	490.13	1,44	1.55	170,295	55,008	35,725	1,322	5,050	178	89,538	52.6%	10	80,757
	Jul-23 Aug-23	13.9 35.1	98.4 136	3,354,201 3,633,718	490,41	1.46	1.54	170,295	54,856 53,692	4,599 4,567	1,443 2,046	5,000	164 144	58,226 56,356	34.2%	11	112,069 113,939
	Sep-23	0.6	198.9	3,913,234	490,92	1.61	1.55	170,295	51,539	2	3,127	5,000	595	49,008	28.8%	13	121,287
8	Oct-23	27.8	285.2 332.8	4,192,751	491.15	1.61	1.56	170,295	49,072 47,254	17,004	4,752	5,000	1,849	63,173	37.1%	14 15	107,122
2	Nov-23 Dec-23	2	388.8	4,472,268	491.41 491.65	1.61	1.57	170,295	45,863	2,226 39,516	5,694 6,897	5,000	2,157 1,199	45,943 79,681	27.0% 46.8%	16	124,352 90,614
5	Jan-24	. 0	404.1	5,031,301	491.89	1.51	1.58	170,295	45,388	58,026	7,244	5,000	824	96,993	57.0%	17	73,302
8	Feb-24 Mar-24	19.2 17.5	320.8 288.7	5,310,818 5,590,335	492.14 492.38	1.61	1.58	170,295	46,902 48,963	37,889 99,900	5,478 5,076	5,000	621 483	79,934 144,060	46.9% 84.6%	18 19	90,361 26,234
1	Apr-24	57.7	192.9	5,869,851	492.63	1.61	1.58	170,295	51,729	1,388	3,062	5,000	509	50,773	29.8%	20	119,522
- 5	May-24 Jun-24	29.2	128.3 90.5	6,149,368 6,428,885	492.89 493.16	1.52	1.58 1.58	170,295	53,926 55,004	26,309 37,209	1,926 1,322	5,000	433 390	78,742 91,231	46.2% 53.5%	21	91,553 79,063
	Jul-24	13.9	98,4	6,708,402	493.42	1.46	1.57	170,295	54,852	4,778	1,443	5,000	358	58,593	34.4%	23	111,702
1	Aug-24	35.1	136	6,708,402	493.42	-	1.57		-	4,719	2,039	5,000	31.5	2,995	0.0%	24	-
8	Sep-24 Oct-24	0.6 27.8	198.9 285.2	6,708,402 6,708,402	493.42 493.42	-	1.57	-		17,421	3,113 4,726	5,000	273 50	12,744	0.0%	25 26	-
ag (1	Nov-24	0.7	332.8	6,708,402	493.42	*	1.57	34	4	2,272	5,659	5,000	10	-	0.0%	27	9.
3	Dec-24 Jan-25	2	388.8 404.1	6,708,402 6,708,402	493.42 493.42	-	1.57	2.5		40,260 58,963	5,853 7,197	5,000	2 0	33,409 51,766	0.0%	28 29	
1	Feb-25	19.2	320.8	6,909,587	493,59	1.61	1.57	122,572	33,754	38,303	5,468	5,000	0	66,589	54.3%	30	55,983
1	Mar-25 Apr-25	17.5 57.7	288.7 192.9	7,110,773 7,311,959	493.76 493.93	1.61 1.59	1.58 1.58	122,572 122,572	35,421 37,356	101,630 1,410	9,417 6,253	28,447 5,000	947	104,186 56,907	85.0% 46.4%	31 32	18,386 65,665
- 8	May-25	29.2	192.9	7,513,144	494.10	1.59	1.58	122,572	38,813	26,529	1,924	5,000	1,852	65,270	53.3%	33	57,302
-	Jun-25	20.7	90.5	7,714,330	494.28	1.52	1.58	122,572	39,589	37,476	1,322	5,040	1,384	77,087	52.9%	34	45,485
6	Jul-25 Aug-25	13.9 35.1	98.4 136	7,915,516 8,116,701	494.46 494.63	1.55	1.58	122,572	39,480 38,641	4,808 4,758	1,441 2,044	5,000	960 782	43,846 42,137	35.8% 34.4%	35 36	78,726 80,435
1	Sep-25	0.6	198.9	8,317,887	494.80	1.61	1.58	122,572	37,089		3,123	5,000	524	34,590	28.2%	37	87,982
	Oct-25 Nov-25	27.8	285.2 332.8	8,519,073 8,720,259	494.97 495.13	1.61	1.58	122,572	35,311 34,001	17,620 2,303	4,744 5,683	5,000	499 451	48,686 31,072	39.7% 25.4%	38 39	73,886 91,500
4	Dec-25	2	388.8	8,921,444	495.30	1.61	1.59	122,572	32,999	40,848	6,884	5,000	487	67,450	55.0%	40	55,122
- 3	Jan-26	0	404.1	9,122,630	495.47	1.61	1,59	122,572	32,657	59,914	7,231	5,000	406	85,746	70.0%	41	36,826
	Feb-26 Mar-26	19.2 17.5	320.8 288.7	9,323,816	495.63 495.80	1.61	1.59	122,572	33,749 35,424	39,058 103,655	5,468 9,723	5,000	355 318	67,694 104,186	55.2% 85.0%	42	54,878 18,386
	Apr-26	57.7	192.9	9,726,187	495.97	1.59	1.59	122,572	37,359	1,438	6,474	5,000	289	58,099	47.4%	44	64,473
	May-26 Jun-26	29.2	128.3 90.5	9,927,373	496.14 496.32	1.61	1.59	122,572	38,811 39,587	27,041 38,193	1,924	5,000	249 278	64,177 76,697	52.4% 62.6%	45 46	58,395 45,875
	Jul-26	13.9	98.4	10,329,744	496.49	1.56	1.59	122,572	39,478	4,899	1,441	5,000	259	43,234	35.3%	47	79,338
	Aug-26	35.1	136	10,530,930	496.66 496.83	1.61	1.59 1.59	122,572	38,639 37,086	4,847	2,044	5,000	253 234	41,695	34.0% 27.9%	48 49	80,877 88,373
3	Sep-26 Oct-26	0.6 27.8	198.9 285.2	10,732,116	496.99	1.61	1.59	122,572	35,308	17,945	3,122 4,743	5,000	212	34,198 48,720	39.7%	50	73,852
	Nov-26	0.7	332.8	11,134,487	497.15	1.61	1.59	122,572	33,997	2,343	5,683	5,000	190	30,848	25.2%	51	91,724
5	Dec-26 Jan-27	0	388.8 404.1	11,335,673 11,536,858	497.31	1.61	1.59	122,572	32,995 32,653	41,495 60,762	6,883 7,230	5,000	157 152	67,764 85,337	55.3% 70.4%	52 53	54,808 36,234
1	Feb-27	19.2	320.8	11,536,858	497.48	-	1.59		-	39,507	5,444	5,000	211	34,274	0.0%	54	
1	Mar-27 Apr-27	17.5 57.7	288.7 192.9	11,536,858 11,536,858	497.47	2	1.59		9	1,441	4,851 3,008	5,000	174 45	99,375	0.0%	55 56	
1	May-27	29.2	128,3	11,536,858	497.47	*	1.59	-	8	27,259	1,920	5,000	19	25,357	0.0%	57	
1	Jun-27	20.7	90.5	11,536,858	497,47	2	1,59			38,427	1,320	5,014	9	37,102	0.0%	58	
	Jul-27 Aug-27	13.9 35.1	98.4 136	11,536,858	497.47	1.61	1.59	126,790	39,968	4,923	1,435 2,044	5,000	5 2	3,506 42,791	0.0%	59 60	83,999
5	Sep-27	0.6	198.9	11,953,076	497.81	1.61	1.60	126,790	38,362		3,123	5,000	1	35,241	27.8%	61	91,549
	Oct-27 Nov-27	27.8	285.2 332.8	12,161,184	497.98 498.15	1.61	1.60	126,790 126,790	36,523 35,167	17,950 2,342	4,744 5,683	5,000	136 144	49,864 31,969	39,3% 25,2%	62 63	76,925 94,820
- 6	Dec-27	2	388.8	12,577,401	498.32	1.61	1.60	126,790	34,130	41,468	6,884	5,000	122	68,836	54.3%	64	57,954
3	Jan-28 Feb-28	19.2	404.1 320.8	12,785,510 12,993,618	498.48 498.65	1.61	1.60	126,790 126,790	33,777 34,907	60,721 39,519	7,231 5,468	5,000	115 109	87,382 69,067	68.9% 54.5%	65 66	39,407 57,723
1 8	Mar-28	17.5	288.7	13,201,727	498.82	1.51	1.60	126,790	36,630	104,659	9,516	29,098	96	107,771	85.0%	67	19,018
š	Apr-28	57.7 29.2	192.9 128.3	13,409,835	498.99 499.15	1.59 1.61	1.60	126,790	38,636 40,146	1,450	6,324	5,000	91 81	57,950 65,524	45.7% 51.7%	68 69	68,840 61,266
3	May-28 Jun-28	29.2	90.5	13,617,944	499.15	1.52	1.60	126,790 126,790	40,146	27,221 38,379	1,924	5,040	90	78,057	61.6%	70	48,732
	Jul-28	13.9	98.4	14,034,161	499.50	1,55	1.60	126,790	40,836	4,914	1,441	5,000	87	44,437	35.0%	71	82,353
3	Aug-28 Sep-28	35.1 0.6	136 198.9	14,034,161	499.50 499.50	-	1.60	14	2	4,854	2,039 3,113	5,000	87 83	2,902	0.0%	72	
1	Oct-28	27.8	285.2	14,034,161	499.50	-	1.60	28		17,920	4,727	5,000	39	13,232	0.0%	74	
7	Nov-28 Dec-28	0.7	332.8 388.8	14,034,161	499.50 499.50	# 1	1.60		3	2,337 41,413	5,659 6,853	5,000	20 11	34,570	0.0%	75 76	- 1
25	Jan-29	0	404.1	14,034,161	499,50	8	1.60	[W]	- 4	60,651	7,198	5,000	6	53,459	0.0%	77	
5	Feb-29 Mar-29	19.2 17.5	320.8 288.7	14,278,335	499.69 499.89	1.61	1.60	148,763 148,763	40,957 42,892	39,552 104,433	5,472 7,742	5,000	3 2	75,040 126,448	50.4% 85.0%	78 79	73,723 22,314
- 1	Mar-29 Apr-29	57.7	192.9	14,766,683	500.08	1.60	1.60	148,763	45,272	1,447	5,037	5,000	42	54,861	36.9%	80	93,901
	May-29	29.2	128.3	15,010,858	500.28	1.58	1.60	148,763	47,103	27,232	1,925	5,000	155	72,566	48.8%	81	76,196
0	Jun-29 Jul-29	20.7 13.9	90.5	15,255,032	500.49	1.48	1.60	148,763	48,046 47,914	38,395 4,916	1,322	5,043	261 206	85,337 51,637	57.4% 34.7%	82	63,426 97,126
- 1	Aug-29	35.1	136	15,743,380	500.89	1.50	1.60	148,753	46,895	4,855	2,045	5,000	179	49,884	33.5%	84	98,878
1	Sep-29 Oct-29	0.6 27.8	198.9 285.2	15,987,555 16,231,729	501.08 501.27	1.61	1.60	148,763 148,763	45,011 42,853	17,911	3,124 4,747	5,000	155 1,012	42,042 57,029	28.3% 38.3%	85 86	106,721 91,734
19	Nov-29	0.7	332.8	16,475,903	501.46	1.61	1.60	148,763	41,263	2,337	5,688	5,000	1,012	39,140	26.3%	87	109,623
8	Dec-29	2	388.8	16,720,077	501.65	1.61	1.60	148,763	40,047	41,373	6,890	5,000	740	75,269	50.6%	88	73,493
1	Jan-30 Feb-30	19.2	404.1 320.8	16,964,251 16,964,251	501.85 501.84	1.61	1.60	148,763	39,632	60,577 39,387	7,237 5,444	5,000	510 400	93,483 34,343	62.8%	89 90	55,280
	Mar-30	17.5	288.7	16,964,251	501.84	-	1.60	- 4	-	103,735	4,850	5,000	331	99,216	0.0%	91	(4)
	Apr-30 May-30	57.7 29.2		16,964,251 16,964,251	501.84 501.84	=	1.60			1,437 27,176	3,008 1,920		42 22	25,277	0.0%	92 93	
	Jun-30	20.7	90.5	16,964,251	501.84		1.60		1	38,310	1,320	5,013	14	36,991	0.0%	94	
5	Jul-30	13.9 35.1		16,964,251	501.84	150	1.60	153,199	48,293	4,908	1,436	5,000	9	3,495	0.0%	95 96	102.095
8	Aug-30 5ep-30	0,6		17,215,707	502.05 502.25	1.59 1.61		153,199	48,293	4,849	2,045 3,125	5,000	4	51,104 43,233	28.2%	97	102,095
	Oct-30	27.8	285.2	17,718,618	502.45	1.61	1.60	153,199	44,131	17,890	4,748	5,000	293	57,567	37.6%	98	95,632
9	Nov-30 Dec-30	0.7		17,970,073 18,221,528	502.65 502.85	1.61		153,199 153,199	42,494	2,334 41,324	5,689 6,891	5,000	255 215	39,394 75,889	25.7% 49.5%	99 100	113,805 77,309
		-				1,700	-140				,			22,302			

TABLE 6.8: TSF CELL C WATER BALANCE - WET CONDITIONS

Year	Month	Rainfall	Evaporation	Cumulative	Tailings Level	Tailings (density	Water in Slurry	Supernatant Runoff	Rainfall Runoff	Evaporation	Pond Volume	Consolidation Volume	Available	TSF	Discharge	Make Up
real	Month		501	Tonnage	Level	Layers	Mass	Siurry	Kullon	Nullibit	Losses	votume	volume	Recycle	8	220	Requirement
		mm	mm	t	RLm	t/m3	t/m3	mª .	m²	m ^a	m*	mª	m ^a	mª	96	m²	m ^a
	Jan-31	0	404.1	18,472,984	503.05	1.51	1.60	153,199	40,815	60,504	7,238	5,000	189	94,270	51.5%	101	58,929
	Feb-31	19.2	320.8	18,724,439	503.25	1.61	1.60	153,199	42,180	39,384	5,473	5,000	171	76,261	49.8%	102	76,937
	Mar-31	17.5	288.7	18,975,894	503.45	1.61	1.60	153,199	44,151	103,916	7,303	15,698	152	130,219	85.0%	103	22,980
	Apr-31	57.7	192.9	19,227,350	503.65	1.60	1.60	153,199	46,608	1,439	4,717	5,000	143	54,172	35.4%	104	99,027
	May-31	29.2	128.3	19,478,805	503.86	1.57	1.60	153,199	48,509	27,114	1,925	5,000	132	73,829	48.2%	105	79,369
	Jun-31	20.7	90.5	19,730,260	504.08	1.47	1.60	153,199	49,479	38,229	1,322	5,044	124	86,466	56.4%	106	66,732
	Jul-31	13.9	98.4	19,981,716	504.29	1.49	1.60	153,199	49,343	4,895	1,442	5,000	120	52,959	34.6%	107	100,239
	Aug-31	35.1	136	19,981,716	504.29		1.60	J 55 15		4,835	2,039	5,000	106	2,901	0.0%	108	
	Sep-31	0.6	198.9	19,981,716	504.29	- 1	1.60			-	3,113	5,000	99		0.0%	109	-
	Oct-31	27.8	285.2	19,981,716	504.29	141	1.60		2 1	17,847	4,725	5,000	35	13,155	0.0%	110	
	Nov-31	0.7	332.8	19,981,716	504.29	*	1.60	39	- 4	2,328	5,659	5,000	14		0.0%	111	59
10	Dec-31	2	388.8	19,981,716	504.29	+ 7	1,60	et 1	F 1	41,245	6,853	5,000	7	34,399	0.0%	112	
	Jan-32	0	404.1	19,981,716	504.29	- 1	1.60	2 S	0.0	60,405	7,197	5,000	4	53,212	0.0%	113	370
	Feb-32	19.2	320.8	20,232,235	504.49	1.61	1.60	152,628	42,023	39,317	5,473	5,000	2	75,869	49.7%	114	76,759
	Mar-32	17.5	288,7	20,482,754	504.69	1.61	1.60	152,628	43,987	103,742	7,302	15,694	2	129,734	85.0%	115	22,894
	Apr-32	57.7	192.9	20,733,273	504.88	1.60	1.60	152,628	46,435	1,437	4,716	5,000	896	54,746	35,9%	116	97,882
11	May-32	29.2	128,3	20,983,791	505.09	1.57	1.60	152,628	48,328	27,069	1,925	5,000	1,417	74,889	49.1%	117	77,740

March Marc	Year	Month	Rainfall	Evaporation	Cumulative Tonnage	Tailings Level	Tailings		Water in Slurry	Supernatant Runoff	Rainfall Runoff	Evaporation Losses	Pond Volume	Consolidation Volume	Available		Discharge	Make Up Requirement
19 19 10 10 10 10 10 10			mm	mm	t	RLm	Layers t/m3	Mass t/m3	mª.	m²	m²	m*	m*	m ⁴			m²	m*
Column C				-		100000000000000000000000000000000000000												51,941
May 10 10 10 10 10 10 10 1	9																	55,507 51,237
1	9	-								100000000000000000000000000000000000000			_	-				59,139
10.20 10.20 10.0	1			The state of the s	The state of the s	A CONTRACTOR OF THE PARTY OF TH		-		- Annual Contract Con		- Andreite de Andr	-			-		42,352
Map 175 289 784, 289 784, 289 784, 289 784, 289 785 78																-		30,956
Map 100	- 1			-		The second second second		2012000	THE RESERVE OF THE PERSON NAMED IN	The second secon			The second name of the second	- Anti-Colonies	-	-		38,218 11,364
Mod 207 Sec 1,865,000 1075 1511 1515 1575				100000000		-	The second second					# 10 m 1 m 1 m 1 m 1 m 1 m 1 m 1 m 1 m 1	-		The second second	-		45,144
March 1985 1882																		36,196
Map	1																	26,890 49,255
Col.			-	100000	-	-						-	-					50,316
		Sep-23			1,740,971		1.61			22,955	- 2		5,000		and the second second	26.2%		55,888
Dec. 2	3			and the designation of the latest section in	Annual State of the State of th	and the second second second	The second secon			The second second second	The second limited by the second	The state of the s	The second second second			-		44,517 58,289
	2	- Charles Parketing St.		- Contraction and	THE RESERVE OF THE PERSON NAMED IN COLUMN 1	The second second second second	and the latest designation of the latest des	Turnish berkin	and the second section is a	- Contractor Contractor	- CONTRACTOR OF THE PARTY OF TH	200000000000000000000000000000000000000	The Part of the Pa	and the second	The second second	-	- Contraction	27,220
March 172 2867	9	200		- Contract		A CONTRACTOR OF THE PARTY OF TH	-	- Interestina		-	The second second second	-				-		11,354
March 1971 1972	9																	24,392 11,364
			1 1 1 1 1 1 1 1 1							-			_	-		-		17,830
March 175 17			29.2	1,100,000,000	2,735,811	-	1.61			23,994			-	30		-		28,692
				-	THE RESERVE AND PERSONS NAMED IN	The second second second		95456	THE RESERVE AND ADDRESS OF THE PARTY OF THE	The second secon	THE RESERVE AND PERSONS ASSESSED.	- American State Company	-			The second second second		17,272 48,187
Sep 24 Col. 200	3			- PARTICIPATION OF THE PARTICI	A STATE OF THE REAL PROPERTY.	A STATE OF THE PERSON NAMED IN		- Annual Contraction			The second secon	THE PERSON NAMED IN COLUMN 1	The state of the s			The second second		40,107
No.24 27 3812 2845213 5913 2 - 1.52 		Sep-24	0.6	198.9	2,984,521	493.58		1.59			-	3,113	5,000	32	-	0.0%	25	-
Berg-14 2 286.1 286.0 286.0																		14
No. 25	3		0.7						-	-					Victorial Control of Control			
March 17.5	-		0	100000000000000000000000000000000000000		-			-		100000000000000000000000000000000000000	-	-		************	-		
Mag-25 197 1936 3,506,002 241 1,105 1,66 11,663 3,615 3,544 5,500 21 5,500 34 6,004 3,103 3,104 6,004 3,104 3,				-		-			-						· · · · · · · · · · · · · · · · · · ·	- Contraction		53,347
May 25			-		The second second second	-		The state of the s	the second second second	The second second			The second second	100				17,525 63,002
Mary 10 10 10 10 10 10 10 1				-		-	-		-	-			_	are the same	The second second second		-	56,001
Aug 25 35.5 1.18 \$4,859.07 94.43 1.48 1.59 116.85 36.834 4.68 2.644 5.060 3.4 30.474 33.47 33.49 34.74 33.49 34.74 33.49 34.74 33.49 34.74 33.49 34.74 33.49 34.74 33.49 34.74 33.49 34.74 33.49 34.74 33.49 34.74 33.49 34.74 33.49 34.74 33.49 34.74 34.		Jun-25	20.7	90.5	3,943,368	494.47	1.53	1.59	116,835	37,737	36,492	1,322	5,039	22	72,890	52.4%	34	43,945
Sep-25 COL 1985 6,558,677 495.00 L61 159 118,685 9,595 - 1,112 5,000 17 12,240 277.00 377.	9															_		75,886 77,362
Nov. 20 0.7 33.26 6.976.276 6.973.10 1.681 1.60 116,635 22.473 3.962 5.000 13 2.6976 5.506 5.64 1.61 1.60 116,635 31.13 59.151 7.227 5.000 30 31 6.6976 5.506 4.61 1.60 116,635 31.13 59.151 7.227 5.000 30 80,137 72.75 41 41 42 116,635 41.15 41 42 116,635 41.15 41 42 116,635 41.15 41 42 116,635 41.15 41 42 116,635 41.15 41 42 116,635 41.15 41 42 42 42 42 42 42 42			-	- Antonio de la compansa del la compansa de la comp		-		-									-	84,586
Berox 9 388.8 597,978.00 495.00 1.61 1.60 116,815 31,477 40,306 5,908 5,000 11 64,971 52.05 40			-	1,100,000			-	-	THE RESERVE THE PERSON NAMED IN	-	-		-	-	- Annahista	-		70,535
Part	4				The state of the s					The second secon	The State of Street State of S							87,821 51,944
Mar-206 275 288.7 5.008,299 696.01 1.61 1.60 116,815 33,79 302,498 5.927 3),075 70 99.910 65.04 41				- Internation	The second second second	A STATE OF THE PARTY OF THE PAR		- Contractor Line	The second second	The second secon	The second second	The state of the s	-	- rysales	and the second second	- Company of the local division in the local	-	33,698
May 20 977 33.0 598.068 398.18 1.59 1.60 116,835 35.059 7.672 5,000 39 37,384 42.38 44.5						the second second second					111111111111111111111111111111111111111			The state of the s		1133 1133		51,457
Negro 722 128.3 507.802 496.34 1.61 1.60 116.855 37.795 78.755 1,324 5,000 48 61.874 53.076 46.0 1.60	8																	17,525 59,301
Dec 207 SO SO SO SO SO SO SO S	-			-					-					-		_	-	54,961
Section Sect				90.5		496.52		1.60		37,735	37,808	1,322			74,236	63.5%		42,599
Sep-266 0.06 159.0 56.19.910 567.01 1.61 1.60 116,835 35.55 1.756 43.102 50.00 38 32,088 27.06 49	- 1	- Contract C	the state of the later of the l	-	THE RESERVE AND ADDRESS OF THE PARTY OF THE	THE RESERVE AND PERSONS ASSESSED.	The second second	- Volentrette	A STREET, SQUARE, SQUARE,		THE RESERVE THE PERSON NAMED IN		The second second second	15.000	The state of the s	and the second second		75,708
Nov-76			The second second	THE RESERVE AND ADDRESS OF THE PARTY OF THE	The state of the s	and the second second second	The state of the s	The Contract of the Contract o		The state of the s		- samutovioles	THE RESERVE OF THE PERSON NAMED IN	- Indiana	the second secon	THE RESERVE NAMED IN	1000000	77,201 84,567
S			-		The second second second second	The second second second	-	-	activation for the state of the state of the state of		The second secon					_	-	70,122
190-27 0			0.7				-					100					-	87,765 51,275
Feb-77 192 330.6 7,586,988 697.66 - 1.00 38,955 5,444 5,000 21 33,542 0.0% 54	,		0															32,986
May-77 23-2 7.588,988 497.65 - 1.60 - - 1.622 3,008 3,000 6 - 0.0% 597.					The state of the s				3 8 8	H. 1								-
May-27 20.2 128.3 7,586,988 897,66 - 1.60 7,6885 1,930 5,000 2 24,670 0.0% 55	- 1	-				-		-			-				97,794	_		
Mul. 27 20.7 50.5 7.886,988 877.66	9					-		-	-	-				-	24,967	and the second second		-
Name				2000000	7,586,988	The second second second	17	-		-	The second second	10.000.000	-		The second secon	- Indiana di Contra	- tempire	
Sep-27 0.6 198.9 7.994,252 498.00 1.61 1.60 123,738 37,447 3,123 5,000 0 34,324 27,7% 61	- 1				Committee of the Part of Street, Stree	The second secon					The state of the s	The second secon			* The second sec	-		81,994
Nov-27	9															_		89,433
6 Dec 27 2 388.8 5.802,669 598.51 1.61 1.61 122,758 33,318 40,701 5.884 5,000 17 67,151 54.39 64 18.078 67	ĵ.		-														-	75,180
India Indi	6	-						-		-	The second second	- toleron and	-		- CARTESTAN	-		92,793 56,605
Minr 28	75	and an experience	0	The second second second	The second second second second second	The second second second second	The second secon		Section Set, Section of the Park of	THE RESERVE OF THE PARTY OF THE		- Taken and the second	The second second second second			The second second second		38,461
Apr-28 57.7 192.9 3,415,177 499.19 1.59 1.61 123,788 37,715 1.418 5,265 5,000 10 55,437 45.6% 68													_					56,429
May-28 29.2 128.3 9.618.310 499.35 1.61 1.61 1227.78 39.188 26.835 1.924 5,000 9 63.908 51.694 670		-	1000000	100000000000000000000000000000000000000		A STATE OF THE PARTY OF THE PAR		-	- Contract C	11.55 M 11.05 M	-				- I was a second and the second		-	18,564 67,321
Mar-29 17.5 18.6 19.024,574 499.70 1.55 1.60 123,758 39,801 4,811 1,441 5,000 10 43,281 35.0% 71.	1								123,758				_			_		59,850
Aug 28 35.1 1.16 10.024,574 499.70 - 1.60 4,753 2,039 5,000 10 2,723 0.0% 72 5ep-28 0.6 198.9 10.024,574 499.70 - 1.60 3,113 5,000 10 - 0.0% 73 6.28 10.024,574 499.70 - 1.60 17,544 4,777 5,000 5 12,822 0.0% 74 809.70 - 1.60 17,544 4,777 5,000 5 12,822 0.0% 74 809.70 - 1.60 2,288 5,699 5,000 2 - 0.0% 75 800.70 10 10 10 10 10 10 10 10 10 10 10 10 10				-				-		720000						-		47,574
Sep-28 0.6 198.9 10,024,574 499.70 - 1.60 - - 3,113 5,000 10 - 0.0% 73																		80,477
Cct-28 27.8 285.2 10,024,574 499.70 - 1.60 - - 17,544 4,727 5,000 5 12,822 0.0% 74		700.67107.00		A transfer of	THE RESERVE AND PARTY AND PERSONS ASSESSED.	The second section is a second section in		- Contractorium	-			- Indiana de la compansa del la compansa de la comp	5,000	10	-	- A lamental and the	73	- 2
7															12,822			
Jan-29	7			100000000000000000000000000000000000000		The second second second						100000000	The state of the s		33,692	100000		
Mar-29 17.5 288.7 10,504,212 500.08 1.61 1.60 146,109 42,130 102,309 7,647 17,599 0 124,193 85.0% 79	- 15	-		-		The second second					The same and the same				The state of the s			
Apr-29 57.7 192.9 10,744,032 500,28 1.60 1.61 146,109 44,466 1,418 4,967 5,000 5 53,520 36.6% 80			200000			-				-				100		A300 F 10 111		72,612
May-29 29.2 128.3 10,983,851 500.48 1.58 1.60 146,109 46,265 26,697 1,925 5,000 96 71,133 48.7% 81																-		21,916 92,589
Hun-29 20.7 90.5 11,223,670 500.69 1.48 1.60 146,109 47,190 37,654 1,322 5,043 198 83,677 57.3% 82 Hul-29 13.9 98.4 11,463,489 500.90 1.49 1.60 146,109 47,060 4,823 1,442 5,000 153 50,637 34.7% 83 Aug 29 35.1 136 11,703,308 501.09 1.60 1.60 146,109 46,060 4,764 2,045 5,000 124 48,904 33.5% 84 Sep-29 0.6 198.9 11,943,128 501.28 1.61 1.60 146,109 44,211 - 3,124 5,000 100 41,187 28.2% 85 Oct-29 27.8 285.2 12,182,947 501.47 1.61 1.60 146,109 42,092 17,589 4,747 5,000 946 55,880 38.2% 86 Nov-29 0.7 332.8 12,422,766 501.66 1.61 1.60 146,109 40,531 2,295 5,688 5,000 1,180 38,319 26.2% 87 Bec-29 2 388.8 12,662,585 501.85 1.61 1.60 146,109 39,337 40,653 6,890 5,000 684 73,784 50.5% 88 Bec-29 2 388.8 12,662,585 501.85 1.61 1.60 146,109 38,330 39,342 7,237 5,000 461 91,697 62.8% 89 Feb-30 19.2 320.8 12,902,405 502.05 - 1.60 - 38,300 39,342 7,237 5,000 355 33,625 0.0% 90 Mar-30 17.5 288.7 12,902,405 502.05 - 1.60 - 101,963 4,850 5,000 17 - 0.0% 91 Apr-30 57.7 192.9 12,902,405 502.05 - 1.60 - 26,712 1,920 5,000 5 24,796 0.0% 93 Jun-30 20.7 90.5 12,902,405 502.05 - 1.60 - 26,712 1,920 5,000 17 - 0.0% 94 Jul-30 13.9 98.4 12,902,405 502.05 - 1.60 - 26,712 1,920 5,000 1 3,403 0.0% 95 Aug-30 35.1 316 31,50,680 502.05 - 1.60 - 4,825 1,436 5,000 1 3,403 0.0% 95 Aug-30 35.1 316 31,50,680 502.25 1.59 1.60 151,261 47,684 4,768 2,045 5,000 1 50,409 33.3% 96 Sep-30 0.6 198.9 13,398,956 502.65 1.61 1.61 151,261 47,684 4,768 2,045 5,000 273 56,706 37.5% 98							-	-	-	-	- Contractions		-			-		74,976
Aug-29 35.1 136 11,703,308 501.09 1.60 1.60 146,109 46,660 4,764 2,045 5,000 124 48,904 33.5% 84 Sep-29 0.6 198.9 11,943,128 501.28 1.61 1.60 146,109 44,211 - 3,124 5,000 100 41,187 28.2% 85 Oct-29 27.8 285.2 12,182,947 501.47 1.61 1.60 146,109 42,092 17,589 4,747 5,000 946 55,880 38.2% 86 Nov-29 0.7 332.8 12,422,766 501.66 1.61 1.60 146,109 40,531 2,295 5,688 5,000 1,180 38,319 26.2% 87 Bec-29 2 388.8 12,662,585 501.85 1.61 1.60 146,109 39,337 40,653 6,890 5,000 684 73,784 50.5% 88 Ian-30 0 404.1 12,902,405 502.05 1.61 1.60 146,109 38,930 59,542 7,237 5,000 461 91,697 62.8% 89 Feb-30 19.2 320.8 12,902,405 502.05 - 1.60 - 38,714 5,444 5,000 355 33,625 0.0% 90 Mar-30 17.5 288.7 12,902,405 502.05 - 1.60 - 10,1963 4,850 5,000 289 97,402 0.0% 91 Apr-30 57.7 192.9 12,902,405 502.05 - 1.60 - 1,412 3,008 5,000 17 - 0.0% 92 May-30 29.2 128.3 12,902,405 502.05 - 1.60 - 26,712 1,920 5,000 5 24,796 0.0% 93 Jun-30 20.7 90.5 12,902,405 502.05 - 1.60 - 24,825 1,436 5,000 1 3.9 38,33 0.0% 95 Aug-30 35.1 136 13,515,680 502.25 1.59 1.60 151,261 47,684 4,768 2,045 5,000 1 1 50,409 33.3% 96 Sep-30 0.6 198.9 13,398,956 502.45 1.61 1.60 151,261 47,684 4,768 2,045 5,000 273 56,706 37.5% 98		Jun-29													The second second			62,432
Sep-29 0.6 198.9 11,943,128 501.28 1.61 1.60 146,109 44,211 - 3,124 5,000 100 41,187 28.2% 85 Oct 29 27.8 285.2 12,182,947 501.47 1.61 1.60 146,109 42,092 17,589 4,747 5,000 946 55,880 38.2% 86 Nov-29 0.7 332.8 12,422,766 501.66 1.61 1.60 146,109 40,531 2,295 5,688 5,000 1,180 38,319 26.2% 87 Dec-29 2 388.8 12,662,585 501.85 1.61 1.60 146,109 39,337 40,653 6,890 5,000 684 73,784 50.5% 88 Ian-30 0 404.1 12,902,405 502.05 1.61 1.60 146,109 38,330 59,542 7,237 5,000 461 91,697 62.8% 89 Feb-30 19.2 320.8 12,902,405 502.05 - 1.60 - 38,314 5,444 5,000 355 33,625 0.0% 90 Mar-30 17.5 288.7 12,902,405 502.05 - 1.60 - 38,314 5,444 5,000 355 33,625 0.0% 90 Mar-30 57.7 192.9 12,902,405 502.05 - 1.60 - 101,963 4,850 5,000 289 97,402 0.0% 91 Apr-30 57.7 192.9 12,902,405 502.05 - 1.60 - 1,412 3,008 5,000 17 - 0.0% 92 Mary-30 29.2 128.3 12,902,405 502.05 - 1.60 - 26,712 1,920 5,000 5 24,796 0.0% 93 Jun-30 20.7 90.5 12,902,405 502.05 - 1.60 - 37,655 1,320 5,013 2 36,325 0.0% 94 Jul-30 13.9 98.4 12,902,405 502.05 - 1.60 - 4,825 1,436 5,000 1 3,403 0.0% 95 Aug-30 35.1 136 13,150,680 502.25 1.59 1.60 151,261 47,684 4,768 2,045 5,000 1 32,73 56,706 37.5% 98 50 0 5 27.8 285.2 13,647,232 502.65 1.61 1.61 151,261 43,576 17,603 4,748 5,000 273 56,706 37.5% 98					The second second second			200	the state of the s				The second second		The second secon	The second second	- Linkson	95,472 97,205
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Nov-301 0.71 332.8133.895.508 502.85 1 1.61 1 1.61 1.151.261 1 41.960 1 2.297 1 5.689 1.50001 228 1 38.797 1 25.641 94 1		Oct-30	27.8	285.2	13,647,232	502.65	1.61	1.61	151,261	43,576	The second secon	4,748	5,000	273	56,706	37.5%	98	94,556
9 Dec-30 2 388.8 14,143,784 503.05 1.61 1.61 151,261 40,724 40,688 6,891 5,000 186 74,707 49.4% 100		Nov-30					1.61			41,960	2,297 40,588	5,689	5,000	228	38,797			112,464 76,554

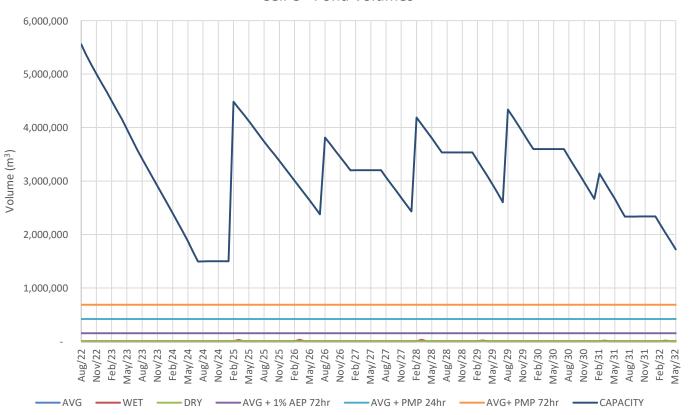
TABLE 6.9: TSF CELL D WATER BALANCE - WET CONDITIONS

Year	Month	Rainfall	Evaporation	Cumulative	Tailings Level	Tailings	density	Water in Slurry	Supernatant Runoff	Rainfall Runoff	Evaporation Losses	Pond Volume	Consolidation Volume	Available	TSF	Discharge	Make Up Requirement
Tear	Month		501	Tonnage	Level	Layers	Mass	Siurry	Rullon	Nullibit	LUSSES	votume	volume	Recycle	1	222	Requirement
		mm	mm	t	RLm	t/m3	t/m3	mª	m²	mª	m*	mª	m ⁴	m*	%	m²	m ^a
	Jan-31	0	404.1	14,392,060	503.25	1.51	1.61	151,261	40,303	59,593	7,238	5,000	158	92,816	51.4%	101	58,445
	Feb-31	19.2	320.8	14,640,335	503.45	1.61	1.61	151,261	41,650	38,804	5,473	5,000	140	75,121	49.7%	102	76,141
	Mar-31	17.5	288.7	14,888,611	503.65	1.61	1.61	151,261	43,594	102,415	7,235	15,329	127	128,572	85.0%	103	22,689
	Apr-31	57.7	192.9	15,136,887	503.86	1.60	1.61	151,261	46,019	1,419	4,567	5,000	118	53,218	35,2%	104	98,043
	May-31	29.2	128.3	15,385,163	504.06	1.57	1.61	151,261	47,896	26,742	1,925	5,000	107	72,820	48.1%	105	78,441
	Jun-31	20.7	90.5	15,633,439	504.28	1.47	1.60	151,261	48,854	37,718	1,322	5,044	111	85,317	56.4%	106	65,945
	Jul-31	13.9	98.4	15,881,714	504.50	1.49	1.60	151,261	48,720	4,831	1,442	5,000	105	52,259	34,5%	107	99,003
	Aug-31	35.1	136	15,881,714	504.50		1.60	S 55 B	-	4,772	2,039	5,000	98	2,830	0.0%	108	
	Sep-31	0.6	198.9	15,881,714	504.50		1.60			-	3,113	5,000	91		0.0%	109	-
	Oct-31	27.8	285.2	15,881,714	504.50	i÷: /,	1.60	[E	i= 1	17,615	4,725	5,000	30	12,918	0.0%	110	. 19
	Nov-31	0.7	332.8	15,881,714	504.50	× 1	1.60	39	- 4	2,298	5,659	5,000	11		0.0%	111	9
10	Dec-31	2	388.8	15,881,714	504.50	· · · · · · · · · · · · · · · · · · ·	1,60	65	F 6	40,708	6,853	5,000	4	33,860	0.0%	112	191
	Jan-32	0	404.1	15,881,714	504.50	- 1	1.60	2 S	0.0	59,619	7,197	5,000	2	52,424	0.0%	113	5 570
	Feb-32	19.2	320.8	16,129,989	504.70	1.61	1.60	151,260	41,649	38,821	5,473	5,000	1	74,998	49.6%	114	76,262
	Mar-32	17.5	288,7	16,378,263	504.90	1.61	1.60	151,260	43,593	102,457	7,222	15,258	1	128,571	85.0%	115	22,689
	Apr-32	57.7	192.9	16,626,537	505.09	1.60	1.60	151,260	46,018	1,420	4,657	5,000	886	53,925	35,7%	116	97,336
11	May-32	29.2	128.3	16,874,811	505.30	1.57	1.60	151,260	47,896	26,754	1,925	5,000	1,397	74,122	49.0%	117	77,139

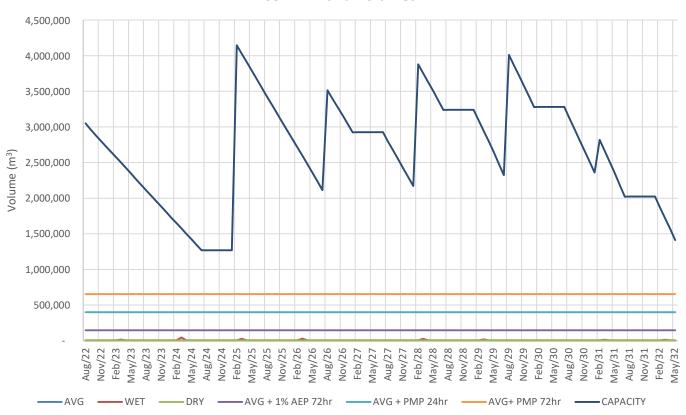
FIGURES



Cell C - Pond Volumes

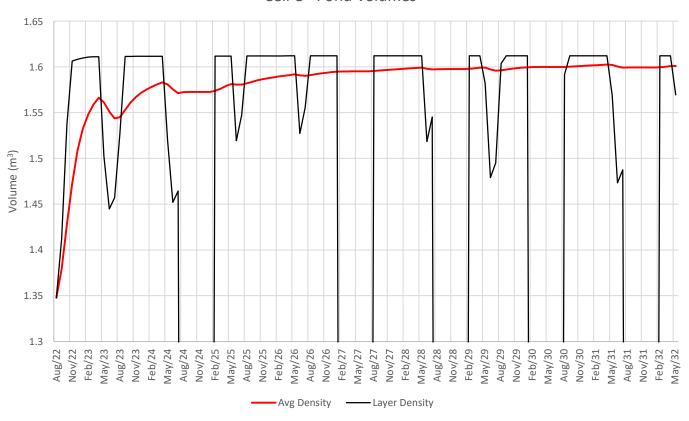




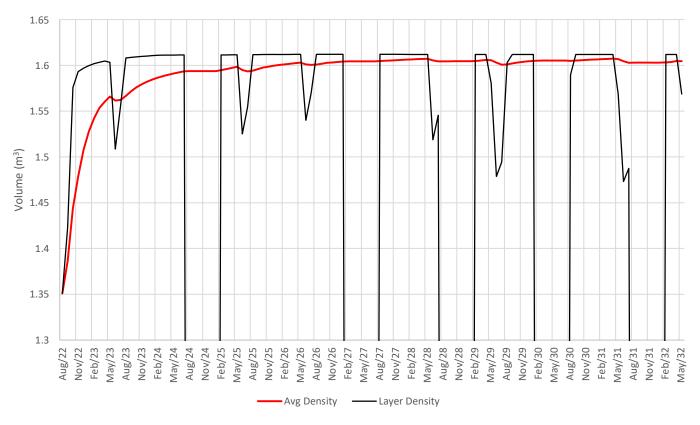




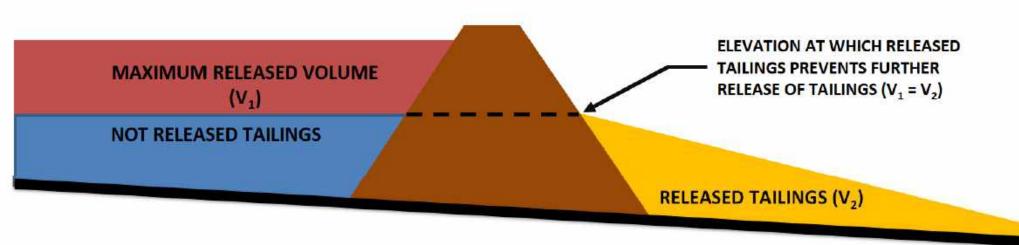
Cell C - Pond Volumes

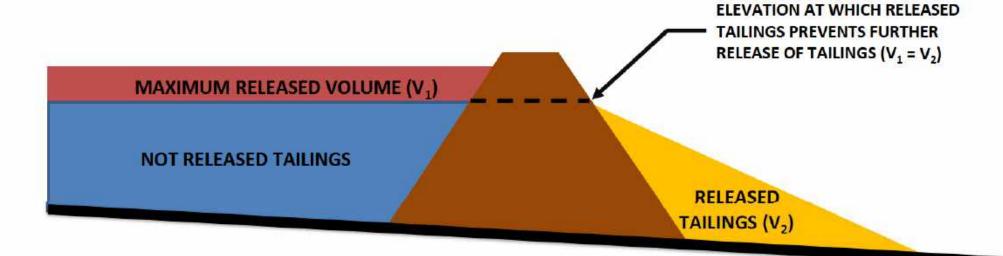




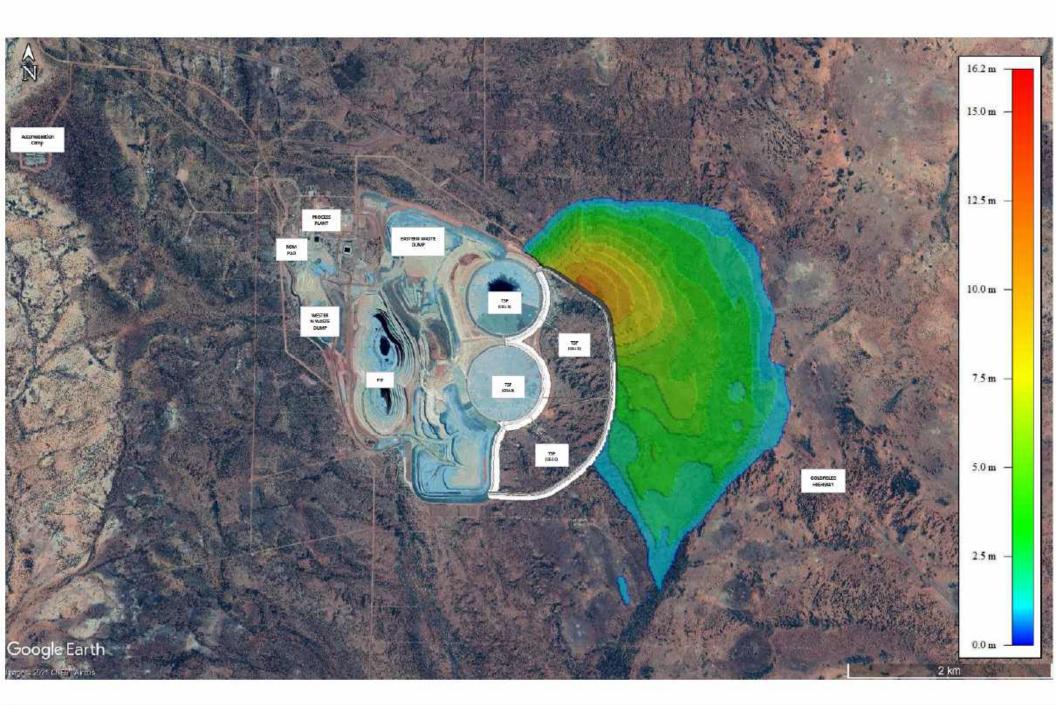




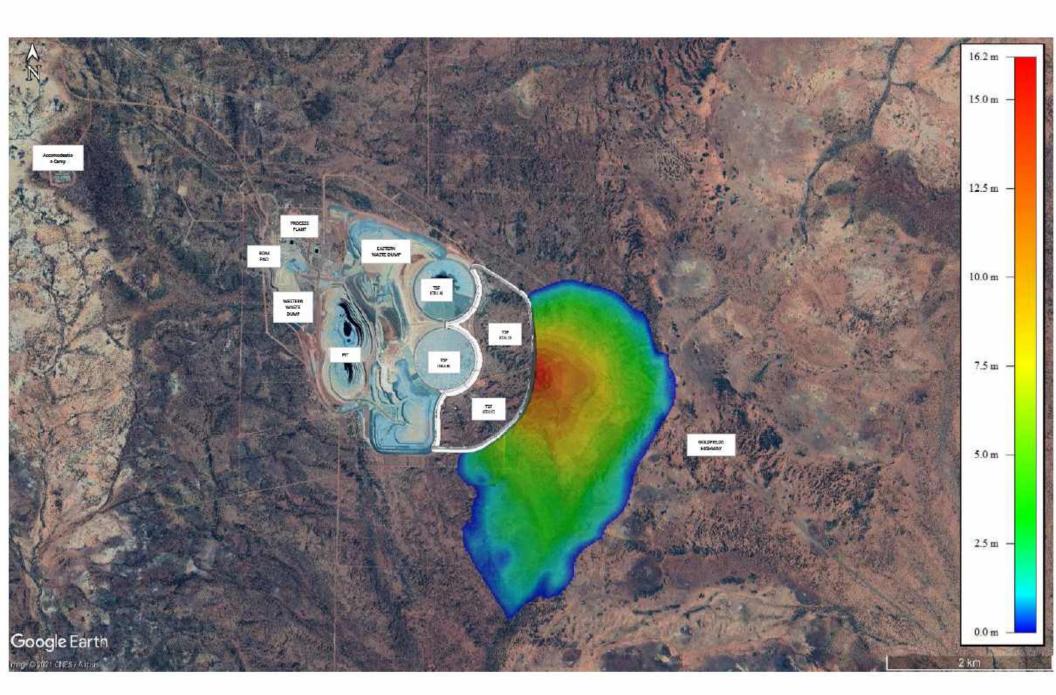








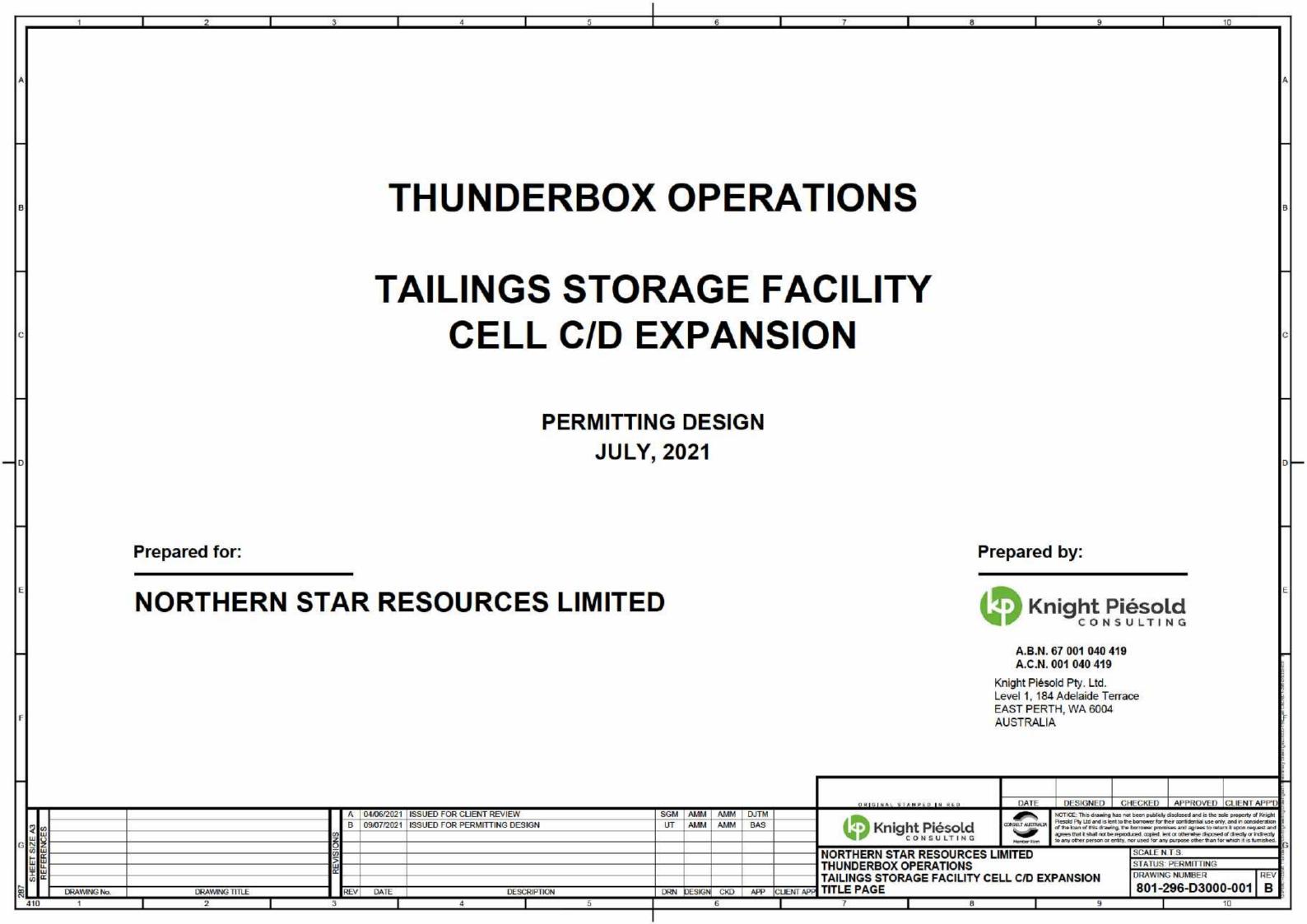


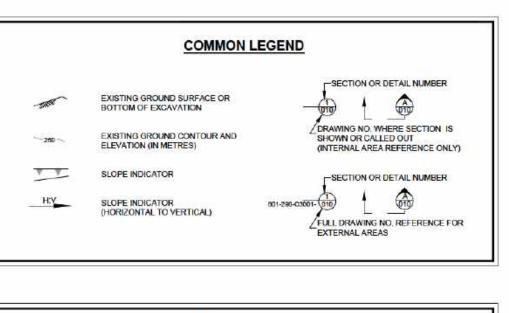




DRAWINGS





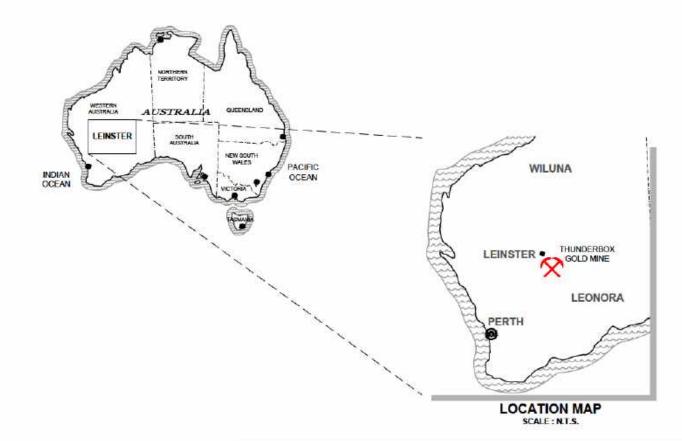


ABBREVIATIONS					
C.M.	CONSTRUCTION MANAGER	MMDD	MAXIMUM MODIFIED DRY DENSITY		
N.T.S.	NOT TO SCALE	MOM	NOMINAL		
APPROX. ~	APPROXIMATELY	OD	OUTSIDE DIAMETER		
BOP	BOTTOM OF PIPE	OMC	OPTIMUM MOISTURE CONTENT		
٤	CENTRELINE	PC	POINT OF CURVATURE		
СРТ	CORRUGATED POLYETHYLENE	PI	POINT OF INTERSECTION		
144440	TUBING	PT	POINT OF TANGENT		
DIA Ø	DIAMETER	PVC	POLYVINYL CHLORIDE		
ELEV	ELEVATION	RC	REINFORCED CONCRETE		
HLF	HEAP LEACH FACILITY	RCP	REINFORCED CONCRETE PIPE		
HDPE	HIGH DENSITY POLYETHYLENE	REQ'D	REQUIRED		
ID	INSIDE DIAMETER	R.L.	REDUCED LEVEL		
IL .	INVERT LEVEL	SDR	STANDARD DIMENSIONAL RATIO		
LCRS	LEACHATE COLLECTION & REMOVAL SYSTEM	SCH	SCHEDULE		
LLDPE	LINEAR LOW DENSITY	SMDD	STANDARD MAXIMUM DRY DENSITY		
	POLYETHYLENE	SOL	SETTING OUT LINE		
MH	MANHOLE	SOP	SETTING OUT POINT		
MAX	MAXIMUM	IWL.	INTEGRATED WASTE LANDFORM		
MIN	MINIMUM	TYP	TYPICAL		

GENERAL NOTES

- 1. ALL DIMENSIONS IN MILLIMETRES UNLESS OTHERWISE NOTED.
- ALL COORDINATES STATED IN GRID PROJECTION: MGA (GDA94) ZONE 51.
- 3. ALL ELEVATIONS IN METRES, RELATIVE TO MEAN SEA LEVEL.
- 4. DRAWINGS ARE NOT TO BE SCALED.
- 5. DRAWINGS TO BE READ IN CONJUNCTION WITH THE TECHNICAL SPECIFICATIONS.

INDEX OF DRAWINGS					
DRAWING TITLE	REV.	DRAWING No.			
TITLE PAGE	В	801-296-03000-001			
INDEX, COMMON LEGEND AND ABBREVIATIONS	В	801-296-D3000-002			
SITE GENERAL ARRANGEMENT PLAN	В	801-296-D3000-100			
GENERAL ARRANGEMENT PLAN - STAGE 1	В	801-296-03000-110			
GENERAL ARRANGEMENT PLAN - STAGE FINAL	В	801-296-03000-115			
TYPICAL EMBANKMENT SECTIONS AND DETAILS - SHEET 1	В	801-296-D3000-301			
TYPICAL EMBANKMENT SECTIONS AND DETAILS - SHEET 2	В	801-296-03000-302			
TYPICAL EMBANKMENT SECTIONS AND DETAILS - SHEET 3	В	801-296-03000-303			
UNDERDRAINAGE LAYOUT	В	801 296 D3000 400			
UNDERDRAINAGE SECTIONS AND DETAILS - SHEET 1	В	801-296-03000-451			
UNDERDRAINAGE SECTIONS AND DETAILS - SHEET 2	В	801-296-D3000-452			
DECANT SYSTEM LAYOUT	В	801-296-03000-500			
DECANT SYSTEM SECTIONS AND DETAILS - SHEET 1	В	801 -296 D3000 -551			
DECANT SYSTEM SECTIONS AND DETAILS - SHEET 2	В	801-296-03000-552			
SPIGOT PIPEWORK DETAIL	В	801-296-03000-810			
SPIGOT CLAMP DETAILS	В	801-296-D3000-812			
MONITORING INSTRUMENTATION LAYOUT	В	801-296-D3000-900			
MONITORING INSTRUMENTATION DETAILS	В	801-296-03000-905			



Knight Piésold

DATE

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SCALE N.T.S.
STATUS: PERMITTING
DRAWING NUMBER

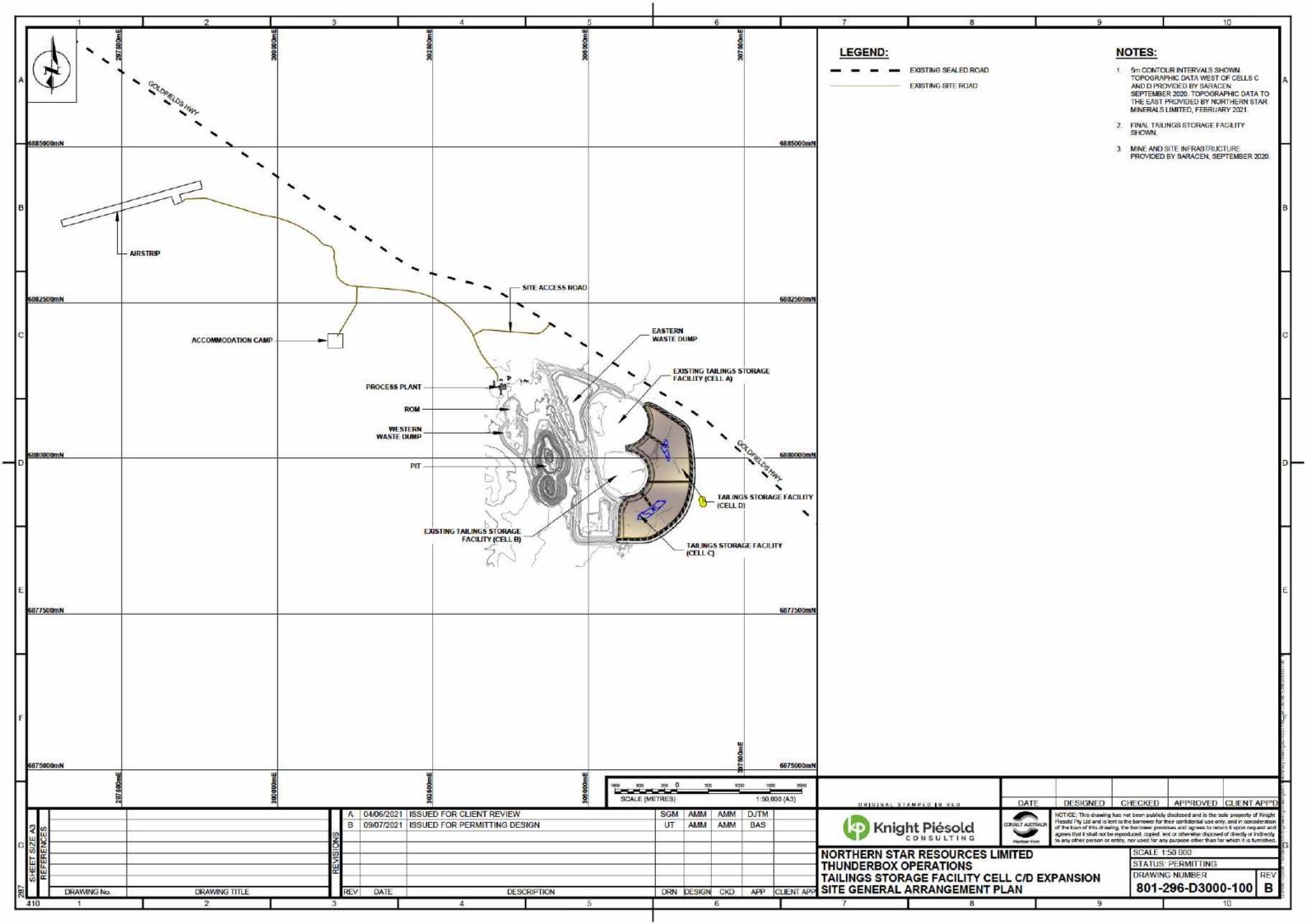
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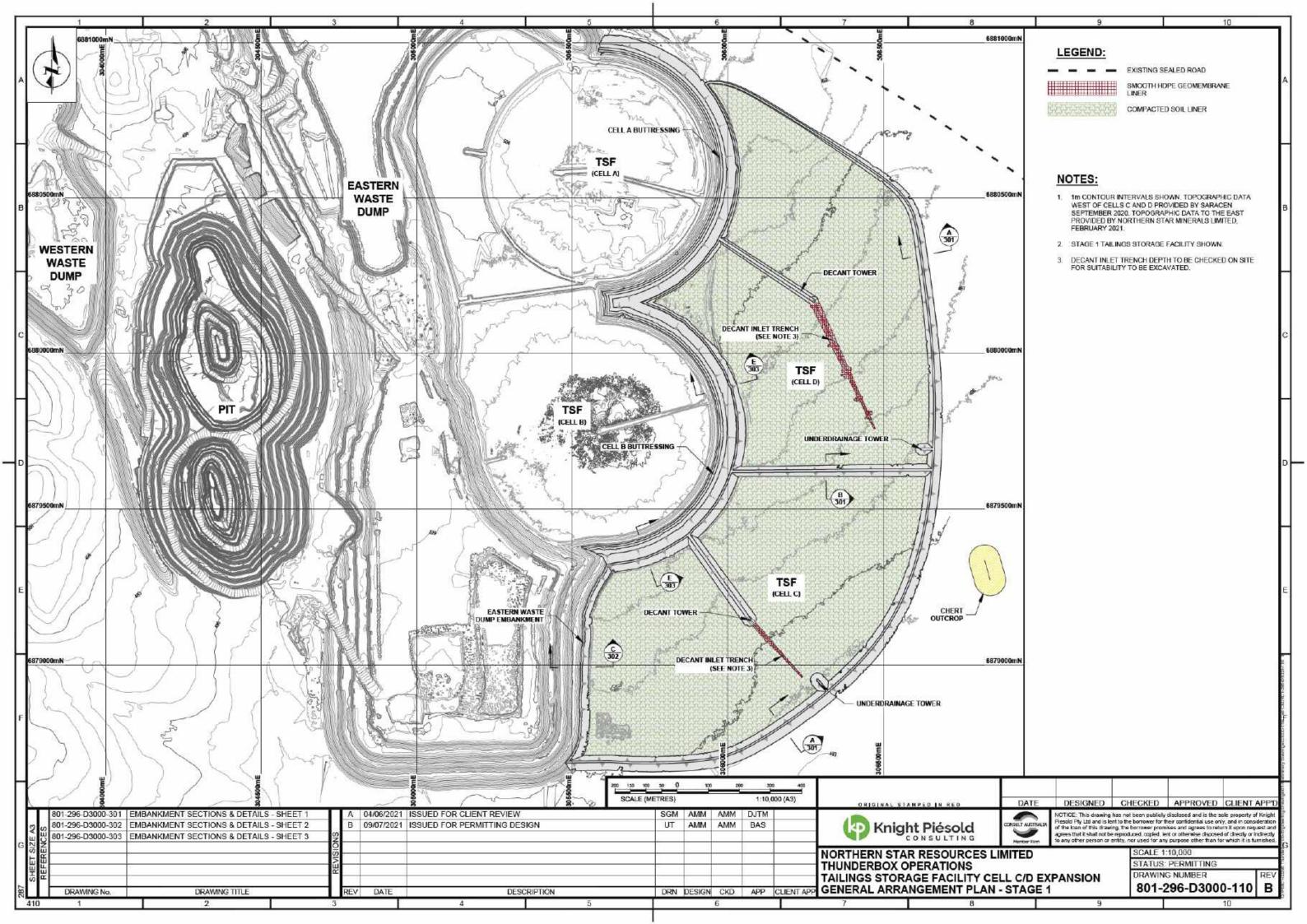
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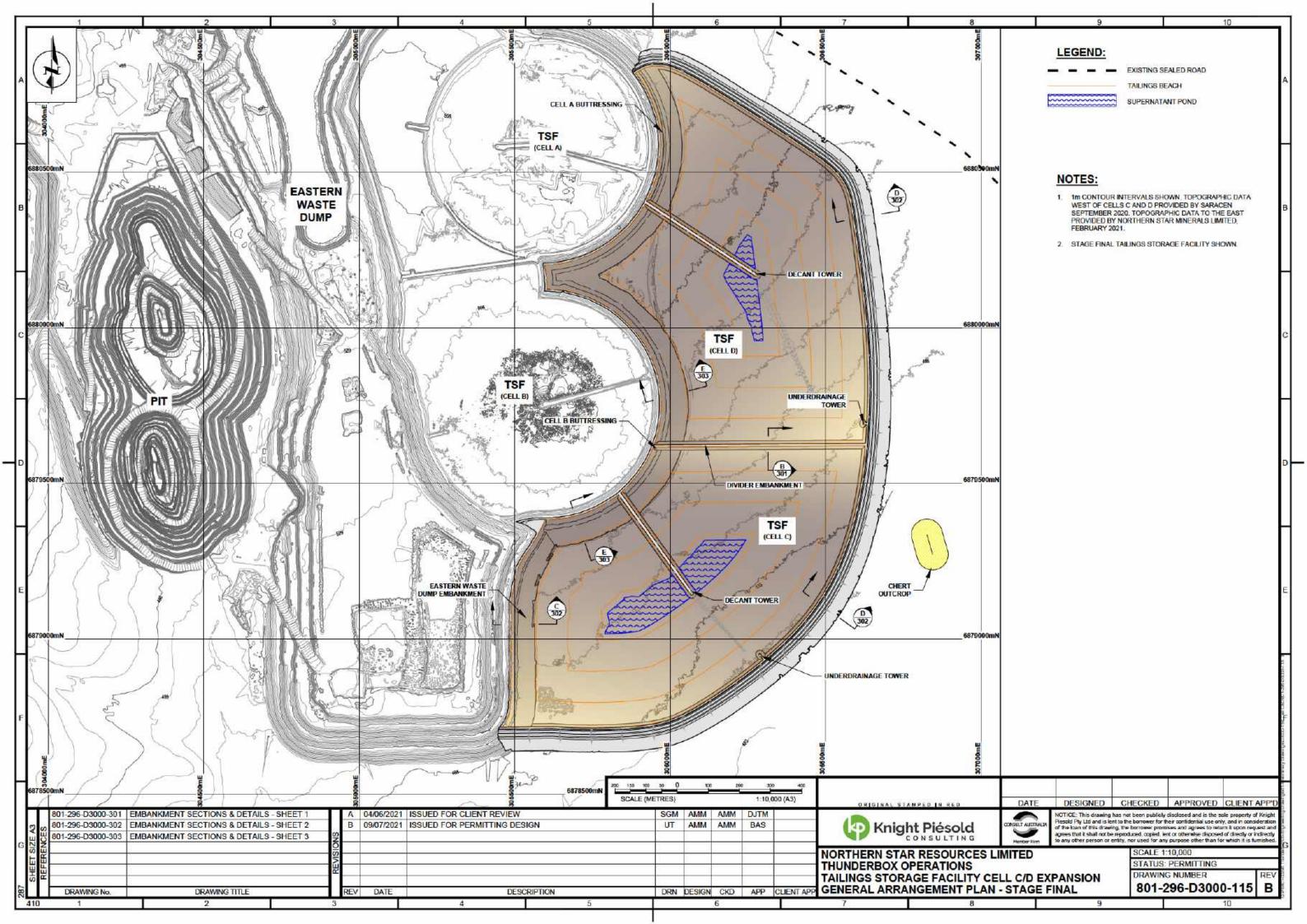
TAILINGS STORAGE FACILITY CELL C/D EXPANSION INDEX, COMMON LEGEND AND ABBREVIATIONS

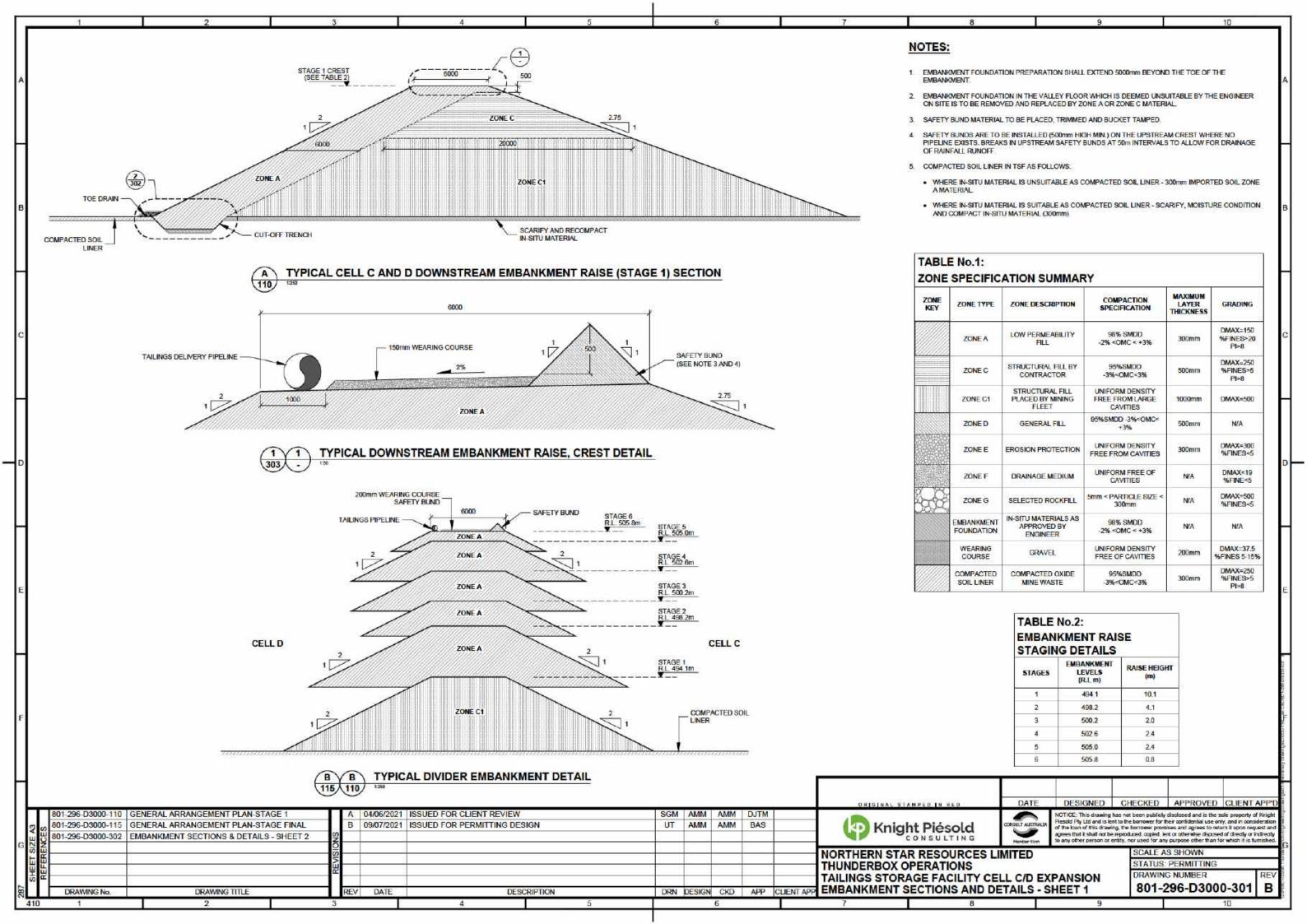
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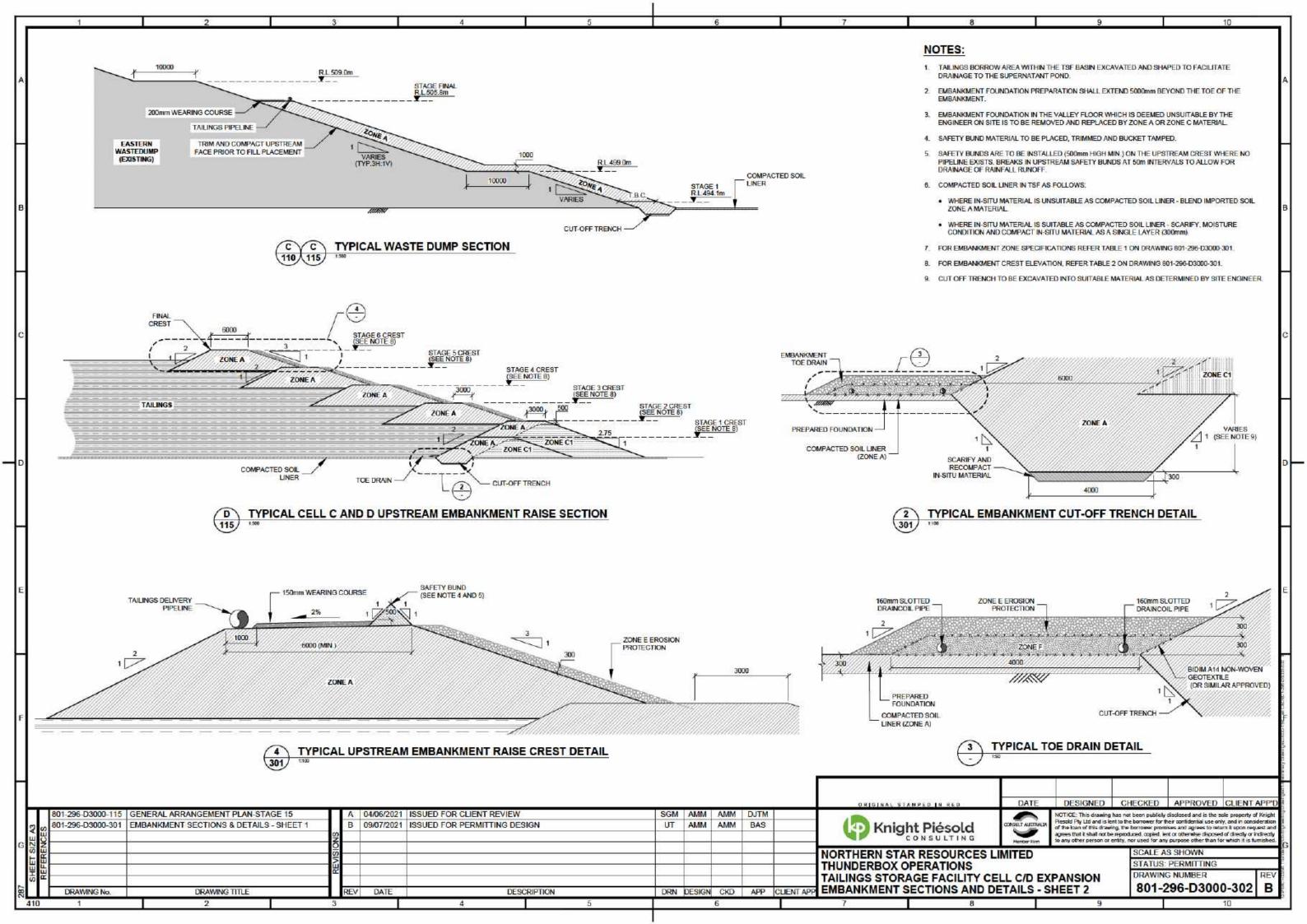
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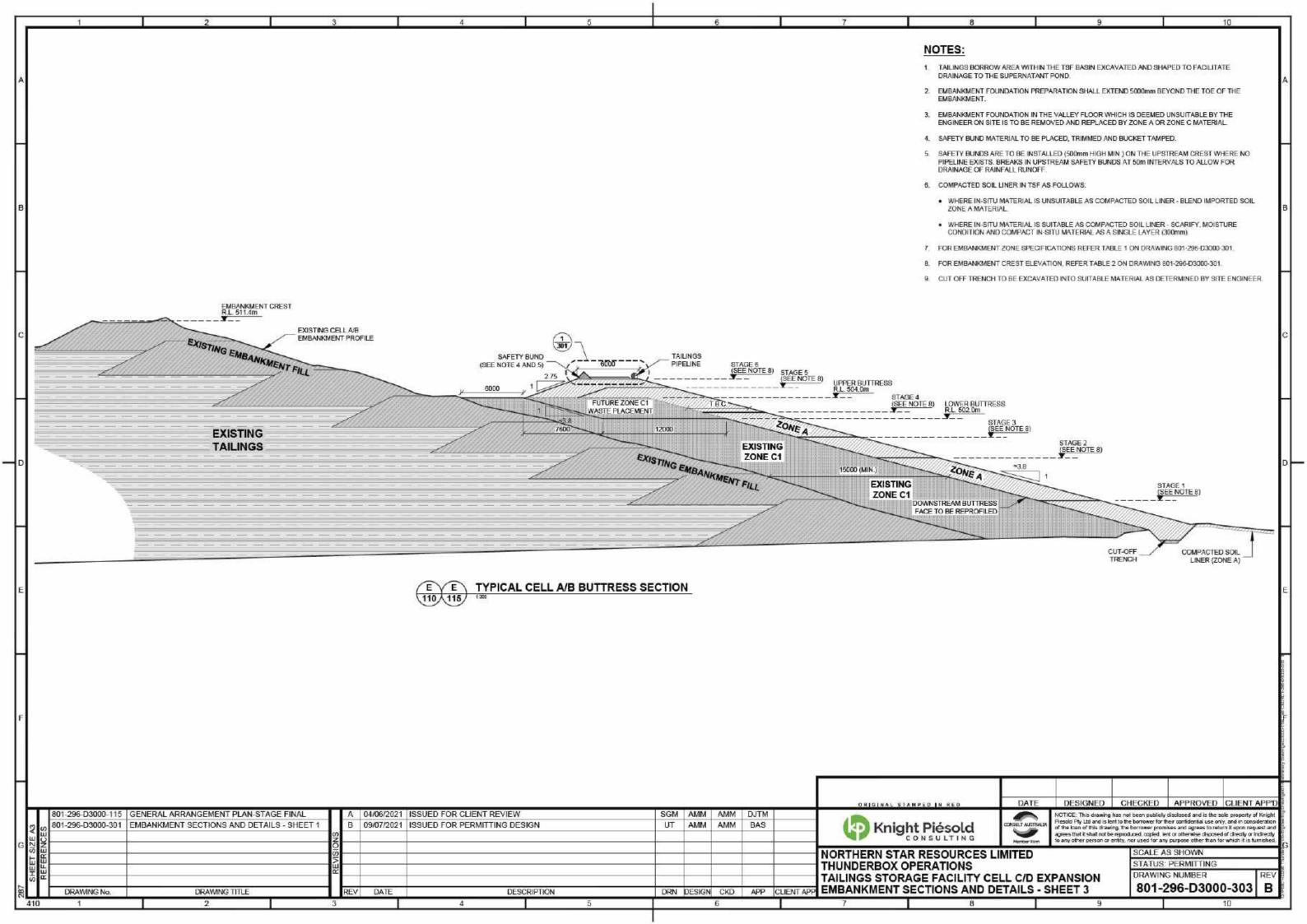


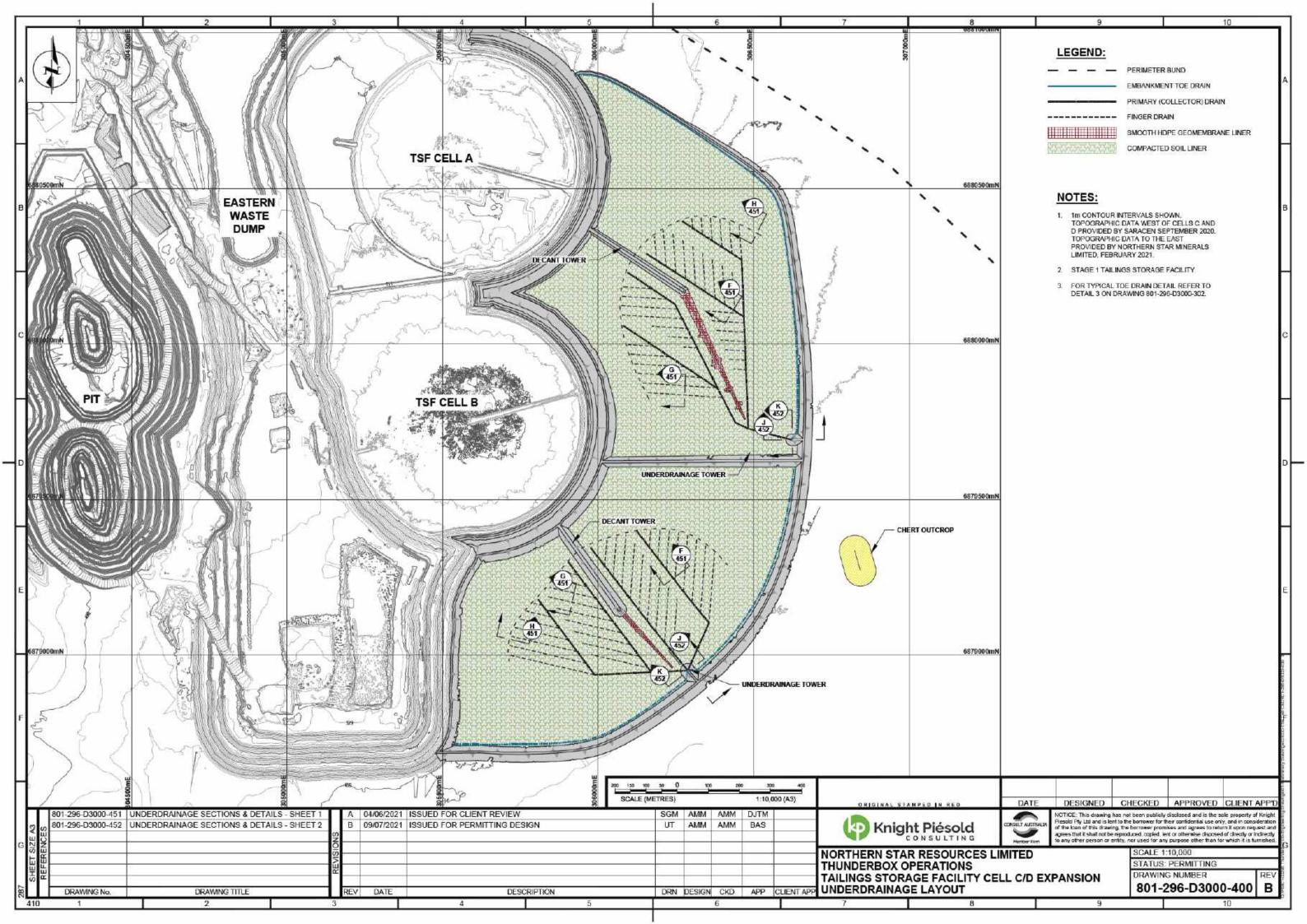


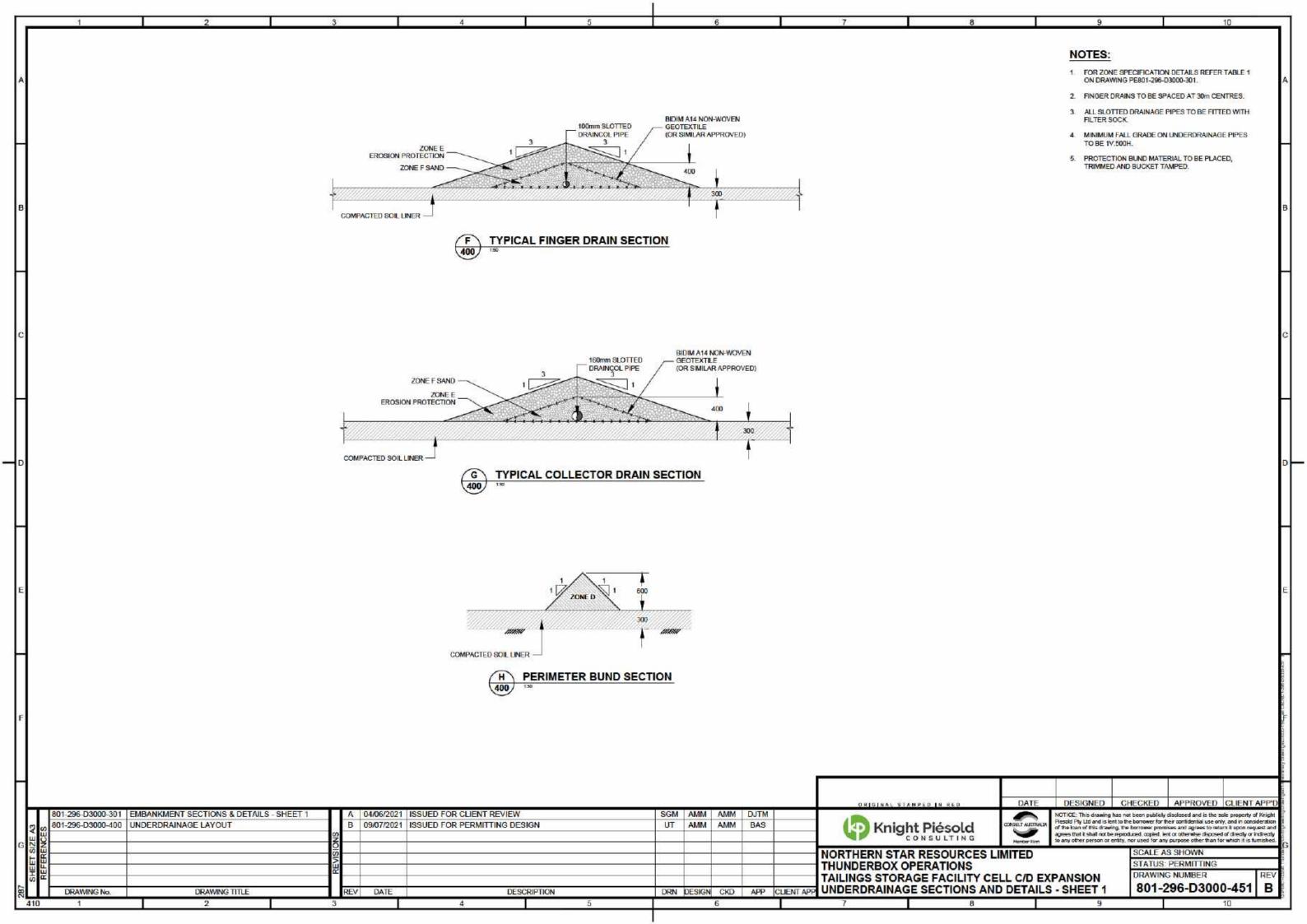


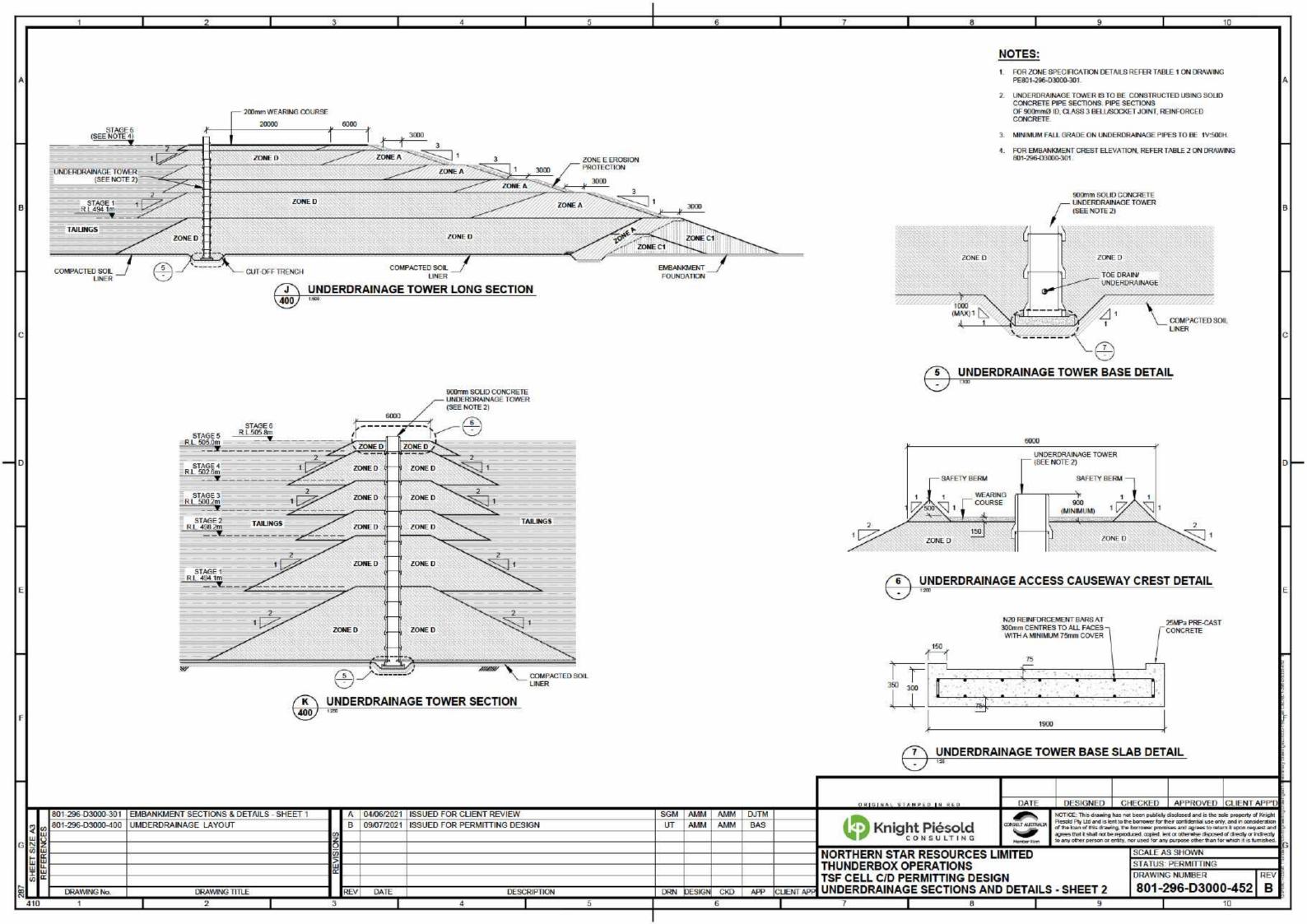


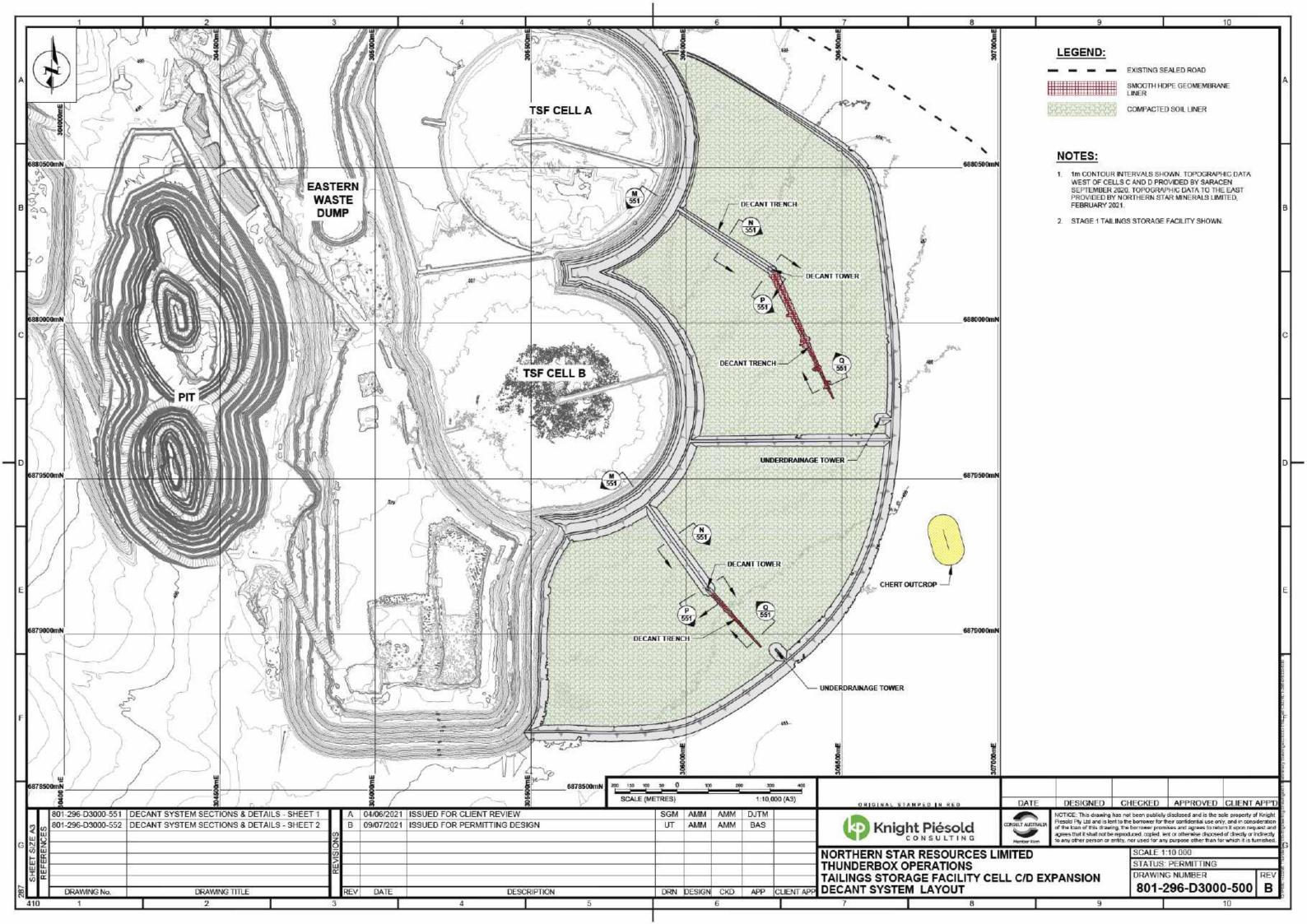


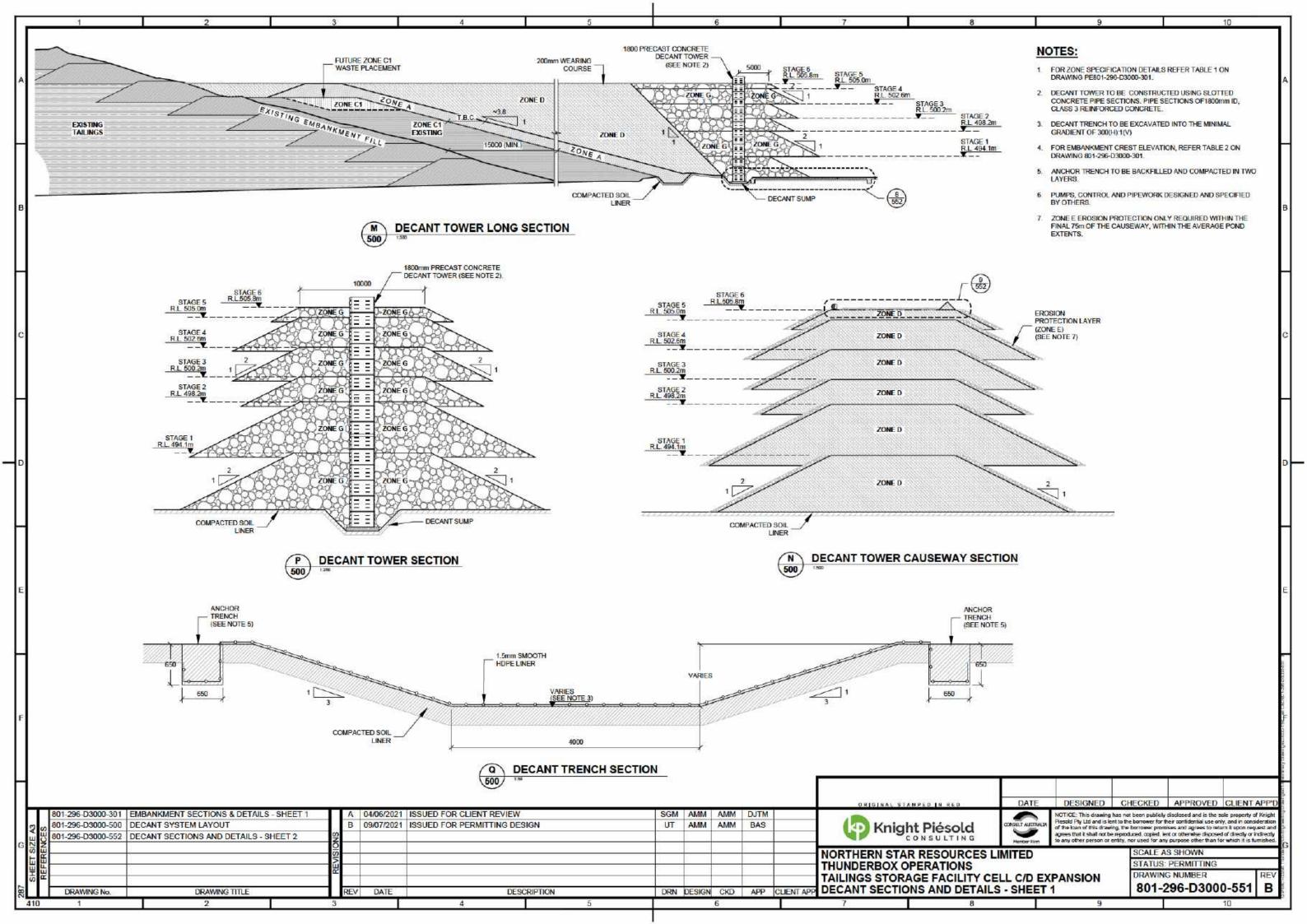


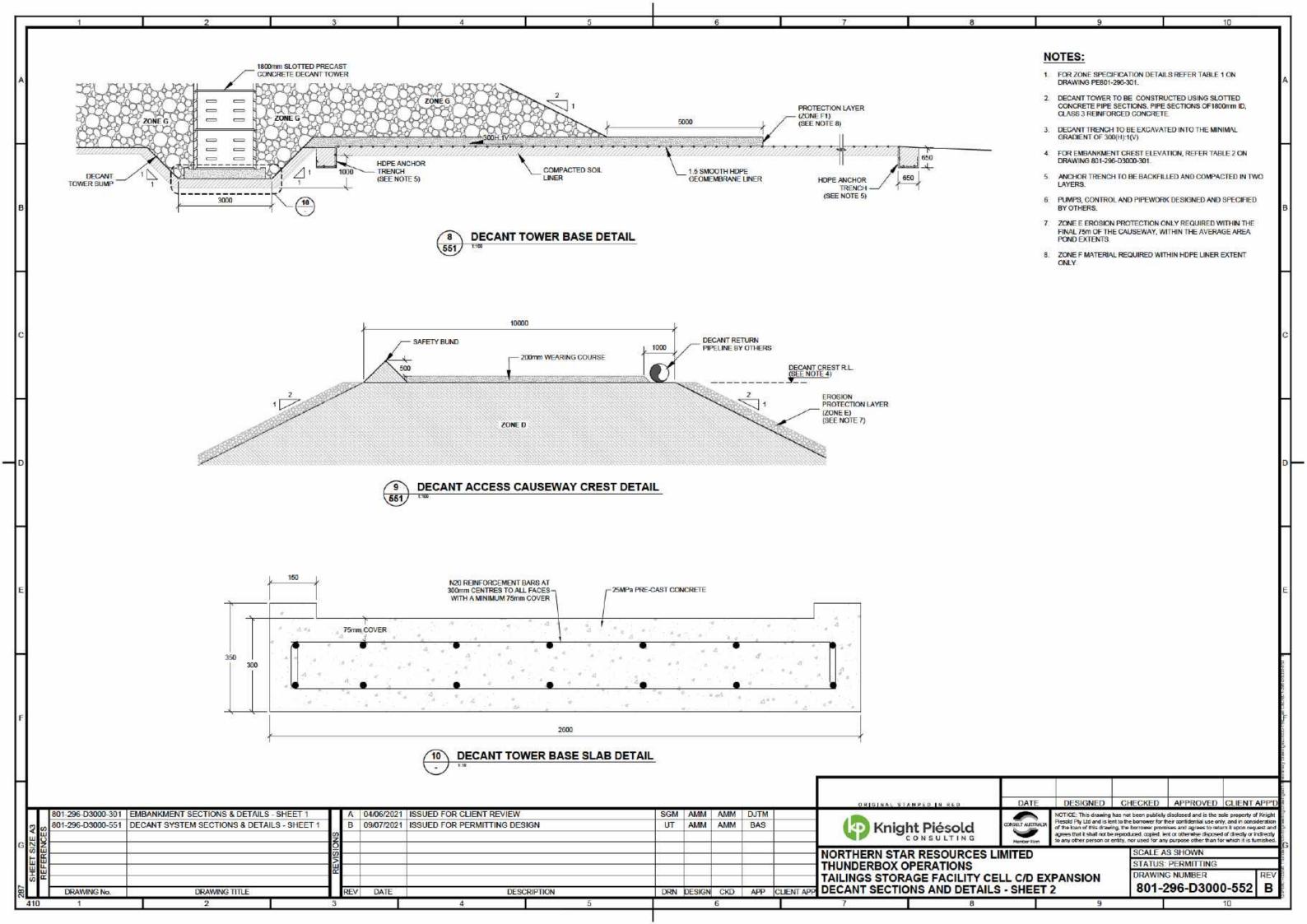


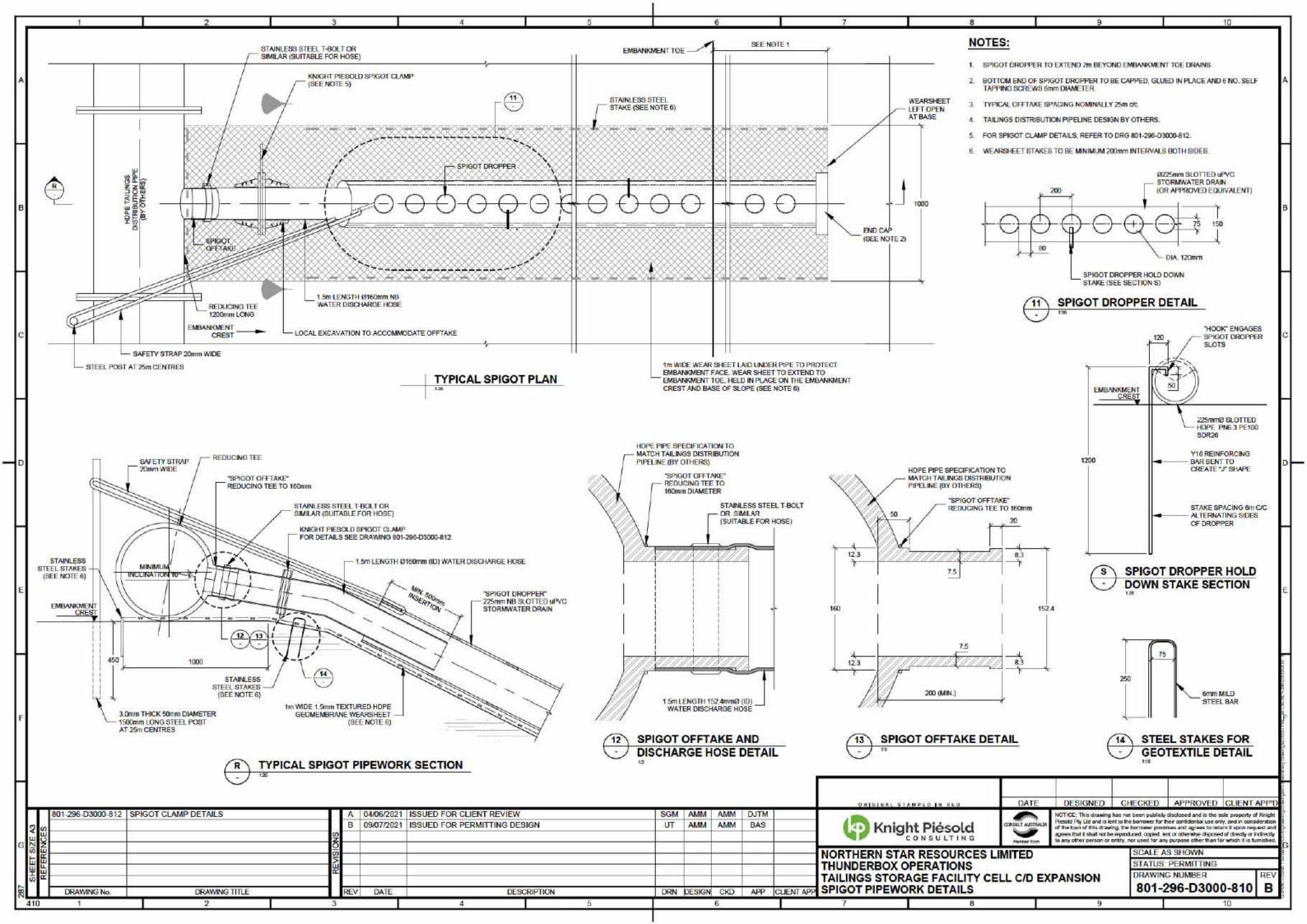


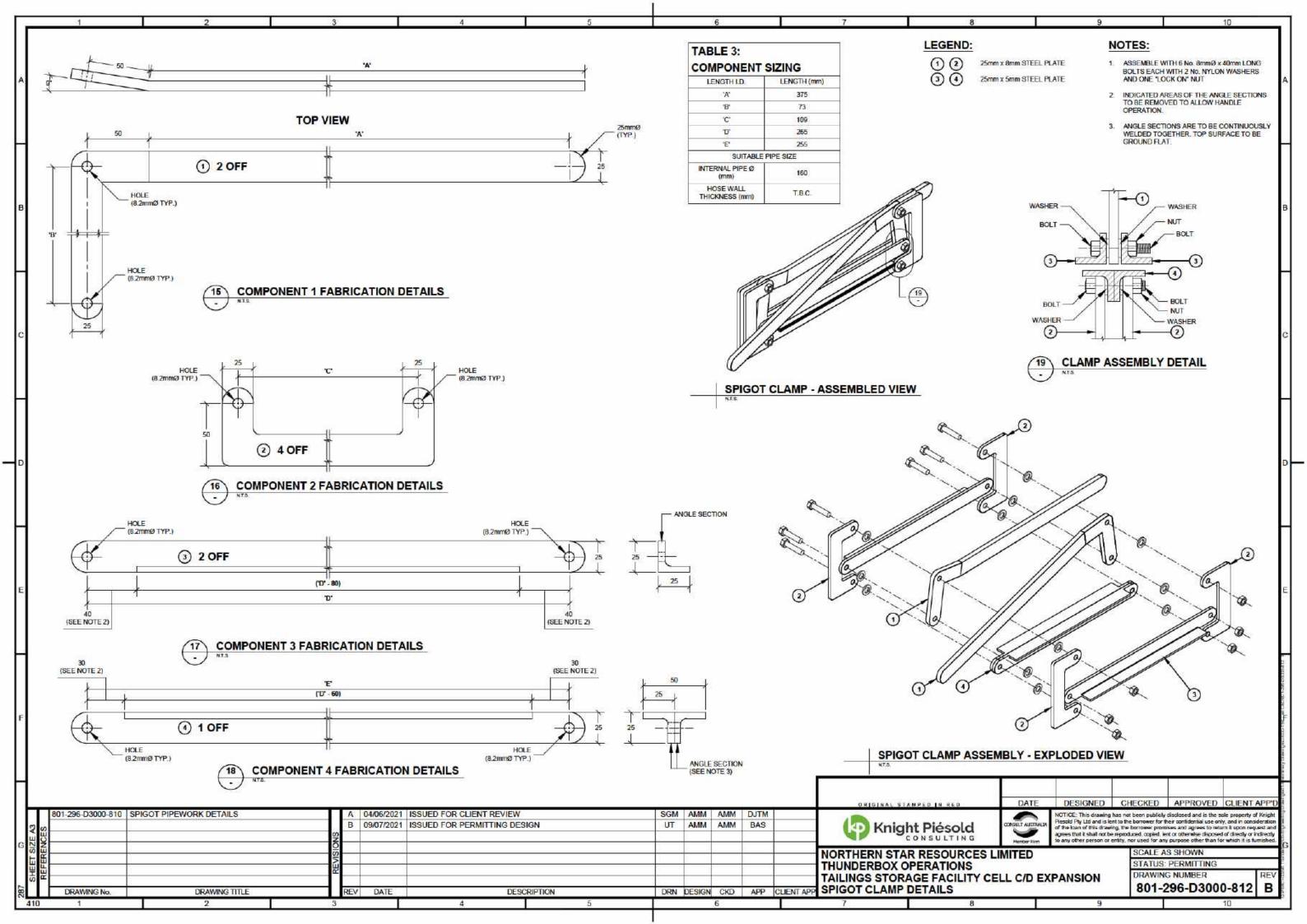


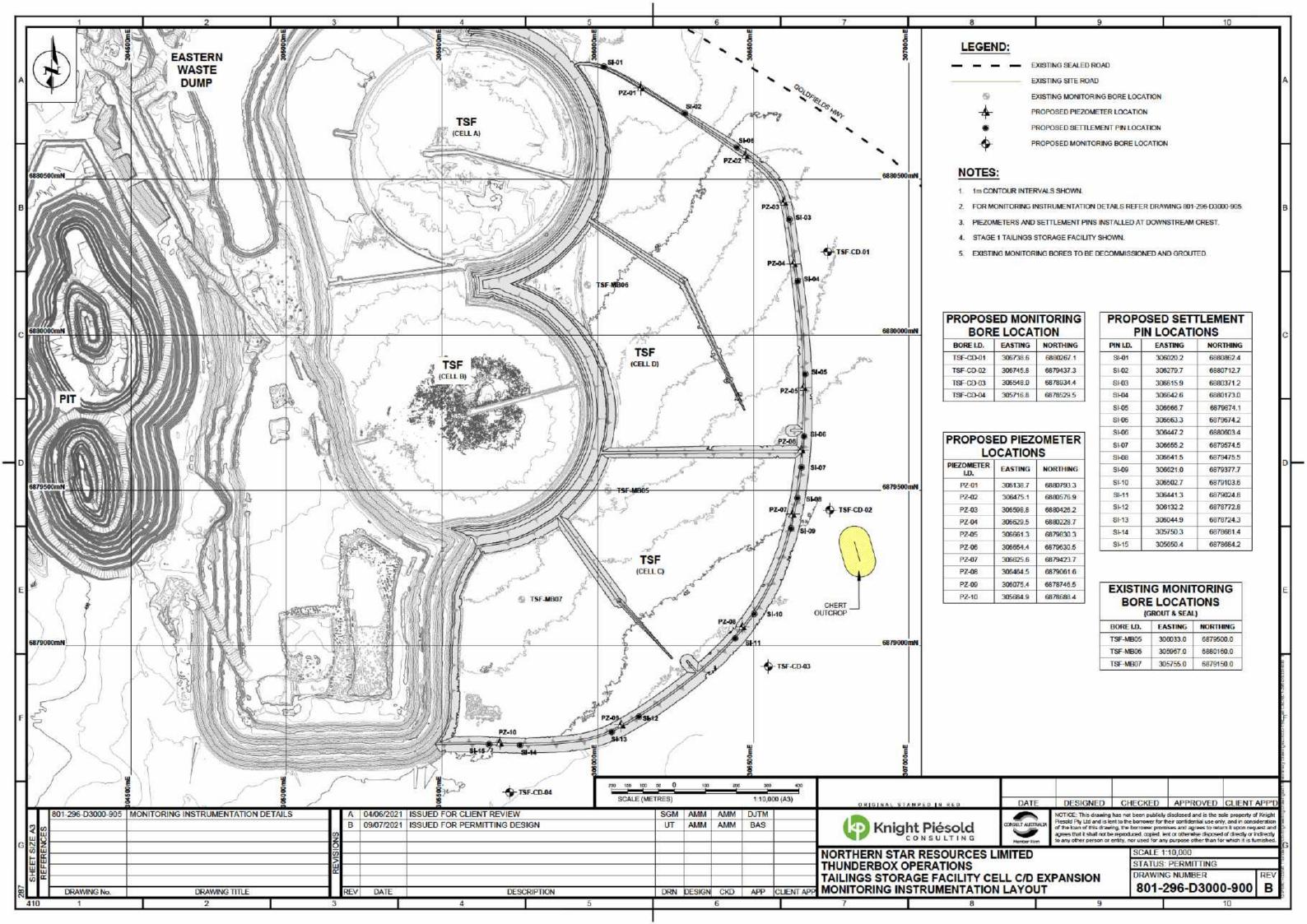


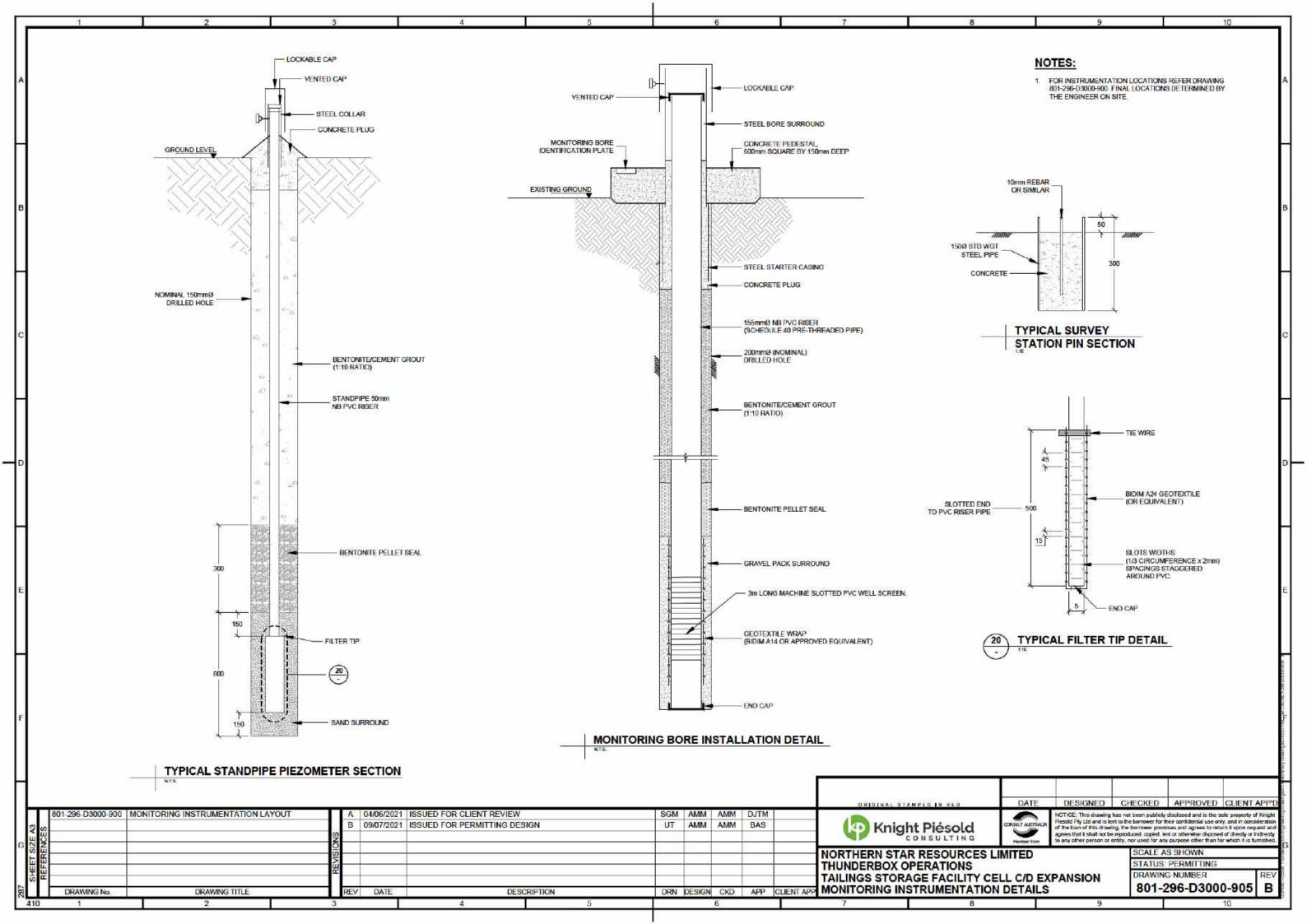












APPENDIX A
Basic Climatology





MEMORANDUM

To: Saracen Metals Pty Ltd Date: 10 October 2017

Our Ref: PE17-00920

KP File Ref.: PE801-00296/10-A djtm M17011

From:

RE: THUNDERBOX PROJECT – BASIC CLIMATOLOGY

Knight Piésold (KP) is pleased to provide the following memorandum regarding basic climatology that has been developed for the Tailings Storage(TSF) Expansion Permitting for the Thunderbox Project site, 290 km North of Kalgoorlie in Western Australia.

1. DATA SOURCE / SUMMARY

Daily historic climate data was obtained from the Scientific Information for Land Owners (SILO) database for use in deriving the basic climatology.

Historic daily climate data from SILO, extracted as a data drill down, spanning the period January 1889 – September 2017 were analysed to obtain desired design inputs for water balance modelling, namely:

- Typical variability of annual precipitation and evaporation;
- Typical variability of monthly precipitation and evaporation; and
- Synthetic monthly precipitation sequences for three climate scenarios:
 - 100 year Annual Recurrence Interval (ARI) Wet precipitation, 1 year duration;
 - Average precipitation, 1 year duration;
 - 100 year ARI Dry precipitation, 1 year duration;

Design storms were developed using the BOM's 2016 IFD generating tool. Probable Maximum Precipitation (PMP) was estimated using the methods developed by the BOM.

2. WATER BALANCE PRECIPITATION ANALYSES

Precipitation data from the SILO dataset were reduced into a standardised format and then analysed on annual and monthly time scales to develop inputs for water balance modelling. The following sections detail the various analyses conducted and the results achieved. It is noted that as the wetter period of the year runs from January through to June, the annual data were analysed as Water Years, spanning October through to September.

2.1 ANNUAL ANALYSES

Daily precipitation records from the SILO dataset were summed to produce annual totals for the 129 years of record available. KP notes that water years were excluded from the analysis if greater than 15% of the daily data were missing. This was done to prevent missing records from introducing bias into computed climate statistics. Of the 129 water years of data, 1 year was excluded from the annual analysis. Sampling statistics were



computed from the annual sums to provide a broad overview of the variability of annual climate at the project site, as given in Table 2.1.

Table 2.1: Annual precipitation statistics

Selected	Precipitation Value
Statistic	(mm)
Average	224
Median	211
Std. Deviation	106
Minimum	44
Maximum	617
25 th Percentile	142
75 th Percentile	288

2.2 MONTHLY ANALYSES

Daily precipitation data from SILO dataset were summed in a manner similar to that described in Section 2.1 to produce monthly totals. Monthly totals were excluded from the analyses if greater than 25% of the daily data from that month were missing (approximately 1 week); for the same reason as given in Section 2.1. Sampling statistics were computed on the monthly sums to provide a broad overview of the variability of monthly climate at the project site, as given in Table 2.2.

Table 2.2: Monthly precipitation statistics (mm)

Table 2.2. Monthly precipitation statistics (IIIII)							1
Month	Average	Median	Std. Dev.	Min.	Max.	25 th Pct.	75 th Pct.
Oct	8	3	12	0	58	0	9
Nov	11	5	16	0	88	1	14
Dec	16	10	18	0	81	2	24
Jan	25	11	35	0	177	2	34
Feb	27	10	41	0	290	3	32
Mar	33	14	46	0	235	3	49
Apr	21	12	27	0	178	2	32
May	23	16	24	0	165	6	32
Jun	23	17	21	0	110	6	32
Jul	17	12	16	0	72	5	22
Aug	13	9	15	0	68	3	18
Sep	7	2	12	0	94	0	8

The sample statistics for monthly precipitation were also depicted as "box and whisker" plots to illustrate the variability of monthly values at the project site, as shown on Figure 2.1. For each month shown on these figures, the "box and whisker" plots are read as follows:

- Top of each "box" indicates the 75th percentile of the monthly values;
- Central line within each "box" indicates the median (or 50th percentile) of the monthly values;
- Bottom of each "box" indicates the 25th percentile of the monthly values;
- Red diamond inside each "box" indicates the average of the monthly values;



- The "whiskers", each of length 1.5 times the inter-quartile range (which is the 75th minus the 25th percentile values) indicate the range of expected readings above and below each "box". Values above and below the "whiskers" are considered to be outliers; and
- Individual monthly outlier values are indicated as blue crosses with monthly readings adjacent.

As is clearly shown on Figure 2.1, March is typically the wettest month of the year at the Thunderbox Project, with measureable precipitation occurring throughout the year. It is noted that the monthly precipitation is quite variable.

2.3 SYNTHETIC WATER BALANCE SCENARIOS

KP performed frequency analysis on annual duration values (from SILO climate dataset) to estimate the statistical likelihood of experiencing extremely "Wet" or "Dry" periods of weather at the project site. Exceedance and non-exceedance probabilities were assigned to various duration totals of daily precipitation values, by sorting the values in descending (for the "Wet" series) and ascending (for the "Dry" series) order.

A number (64) of different probability distributions, e.g.: Log-Pearson 3, Generalised Extreme Value, Wakeby, Inverse Gaussian, etc. were fitted to the various sums of daily precipitation data using EasyFit 5.4 Professional software. Three of the best fits were selected for comparison in each case, as shown on the following figures:

- Wet precipitation series, 1 year duration Figure 2.2; and
- Dry precipitation series, 1 year duration Figure 2.3.

Ideally, the distributions having the best weighted Kolmogorov-Smirnov and Anderson-Darling goodness-of-fit test ranking results will match across all scenarios, thereby providing consistent results. In this case, the goodness-of-fit test results indicated the following on a case-by-case basis:

- 1 year duration Wet scenarios = Log-Pearson 3 (KP Fit) distribution; and
- 1 year duration Dry scenarios = Johnson SB distribution.

The Log-Pearson 3 distribution was fit to the data for the Wet precipitation series for the data points with ARIs ≥10 years using least-squares regression analysis. This was done as this fit provides a more reasonable fit to the larger wet events than the other fits. This fit was then selected for estimating the extreme wet precipitation series depths. For the Dry precipitation series the Johnson SB distribution, as fit by EasyFit 5.4, was selected as it was the best statistical fit. The resulting estimated extreme wet and dry annual precipitation depths are shown in Table 2.3.

 Table 2.3: Design precipitation, various duration Wet and Dry totals

ARI	1 Year Wet / Dry Precipitation Totals (mm)		
(yr)	Wet	Dry	
100	571	52	
50	506	62	
20	422	82	
10	359	102	



In order to apportion the statistically-computed climate series precipitation totals to monthly time series for use in water balance modelling, the following observed rainfall patterns were used:

- 1 year duration Wet scenarios = (October 1941 September 1942), the wettest observed water year of record;
- 1 year duration Average scenario = (October 1940 September 1941), the median observed water year of record;
- 1 year duration Dry scenarios = (October 1968 September 1969), the driest observed water year of record;

Ratios of observed monthly to observed total precipitation were computed for each scenario. These ratios were then multiplied by the computed statistical totals (at the selected design frequency, 100 years ARI) to form the desired synthetic climate scenarios for water balance modelling. The resulting climate scenarios are summarized (in monthly format) in Table 2.4.

Table 2.4: Annual synthetic climate scenarios

Table 2.4: Annual synthetic climate scenarios								
Month	Annual C	limate Scena	rios (mm)					
	100 year	Average	100 year					
	Dry		Wet					
Oct	0.0	27.8	29.9					
Nov	2.4	0.7	3.9					
Dec	0.0	2.0	69.1					
Jan	4.0	0.0	101.2					
Feb	7.3	19.2	65.8					
Mar	0.0	17.5	173.3					
Apr	2.4	57.7	2.4					
May	46.2	29.2	45.4					
Jun	0.0	20.7	64.0					
Jul	0.0	13.9	8.2					
Aug	0.0	35.1	8.1					
Sep	0.0	0.6	0.0					

It is noted that the 100 year Wet year may have months where the precipitation is less than the average or the 100 year dry year. Similarly the 100 year Dry may have months where the precipitation is greater than the average or the 100 year wet year. This not an error and is due to the rainfall pattern within the specific year selected to develop the rainfall patterns.

3. DESIGN STORMS

Aside from climate scenarios for water balance modelling, design storms were derived for site water conveyance / handling design.

3.1 PROBABLE MAXIMUM PRECIPITATION

The Thunderbox Project site lies in the GSAM-GTSMR Transition Zone. As such the Generalised Tropical Storm Method, Revised (GTSMR) procedures (Ref. 1) and the Generalised Southeast Australia Method (GSAM) (Ref. 2) were evaluated to determine



the governing method for deriving PMP for storm durations ≥24 hours. The results for the Thunderbox Project site are summarised in Table 3.1.

Table 3.1: PMP Depths

Duration (h)	GTSMR IAZ Depth (mm)	GSAM IAZ Depth (mm)
24	560	540
30	620	560
36	660	580
48	760	610
60	840	630
72	920	660
96	1,040	670

As the GTSMR IAZ produced the more conservative results, it was selected for use at the Thunderbox Project.

3.2 INTENSITY / FREQUENCY / DURATION

Intensity / Frequency / Duration (IFD) and corresponding Depth / Frequency / Duration (DFD) relationships for design storms (10 years \leq ARI \leq 2000 years) were sourced from the BOM 2016 IFD tool for the grid cell (28.19%, 121.02%). It is noted that for storms greater than the 100 year ARI, only the 24 hour to 168 hour durations are available. The design storm information sourced is summarised as storm depths in Table 3.2.

Table 3.2: Design storm depths

Duration		Pre	cipitation	Depth (m	nm) for a	given AR	l (yr)	
(h)	10	20	50	100	200	500	1000	2000
0.25	17.6	21.7	27.8	33.0	-	-	-	-
0.50	23.4	29.1	37.4	44.6	1	-	-	-
1	29.6	36.9	47.8	57.3	ı	-	-	-
2	36.7	45.8	59.4	71.4	-	-	-	-
3	41.6	51.8	67.2	80.4	-	-	-	-
6	52.0	64.5	82.8	98.4	1	-	-	-
12	65.4	80.8	103.0	121.0	ı	-	-	-
24	81.3	100.0	127.0	149.0	169.0	201.0	227.0	255.0
48	97.0	121.0	154.0	182.0	212.0	257.0	294.0	335.0
72	104.0	131.0	169.0	201.0	238.0	293.0	340.0	392.0
96	108.0	136.0	177.0	212.0	253.0	315.0	368.0	429.0
120	110.0	139.0	181.0	218.0	262.0	327.0	384.0	450.0
144	111.0	140.0	183.0	221.0	266.0	332.0	391.0	460.0
168	111.0	140.0	184.0	222.0	266.0	333.0	391.0	461.0

4. EVAPORATION

Daily evaporation data from SILO dataset were summed in a manner similar to that described in Section 2.1 to produce monthly totals. Monthly totals were excluded from the analyses if greater than 25% of the daily data from that month were missing (approximately 1 week); for the same reason as given in Section 2.1. Average monthly pan evaporation values for the site were calculated from the SILO dataset for 1970 onwards and are summarised in Table 4.1 below.

Table 4.1: Average monthly pan evaporation.

Month	Pan Evap. (mm)
Oct	285
Nov	333
Dec	389
Jan	404
Feb	321
Mar	289
Apr	193
May	128
Jun	91
Jul	98
Aug	136
Sep	199

As per "Evaporation Data for Western Australia" (Ref. 3), the pan factor for the Thunderbox Project is estimated as 0.66.

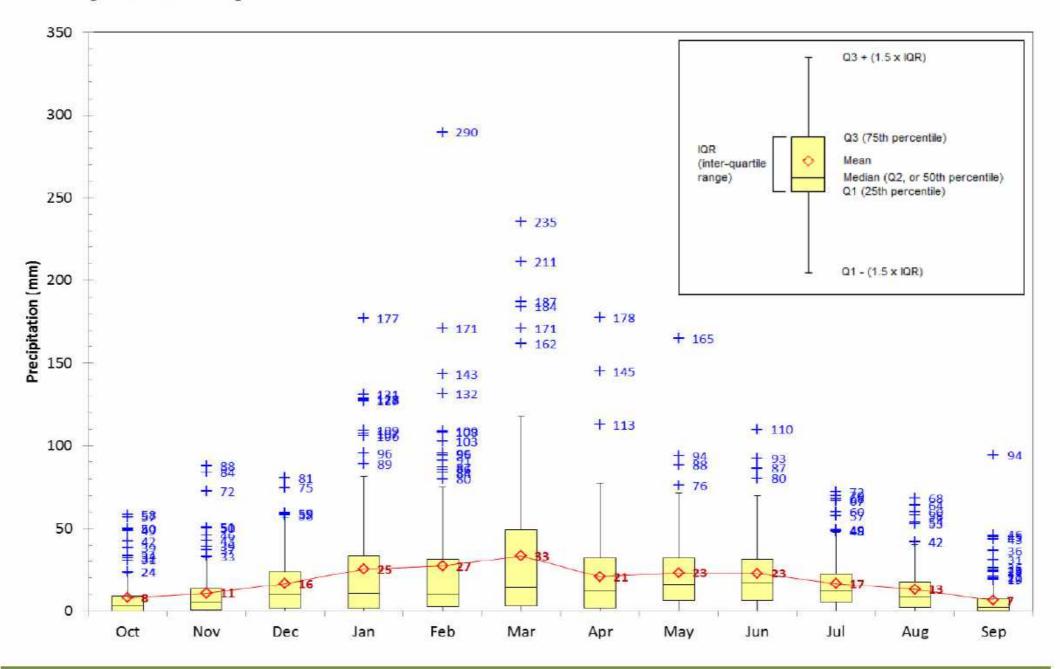


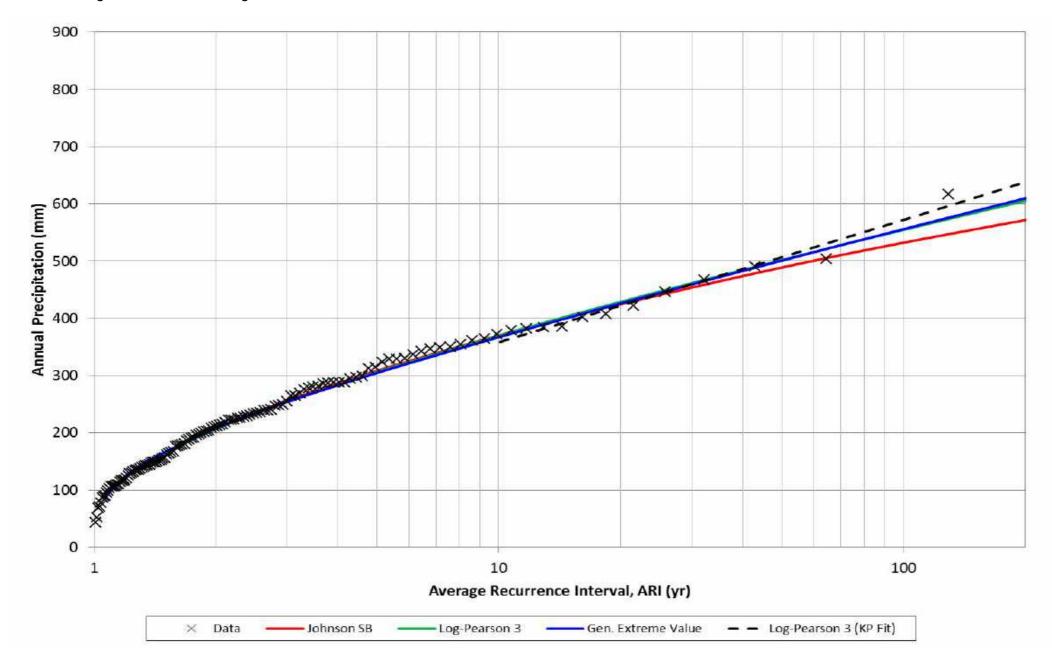


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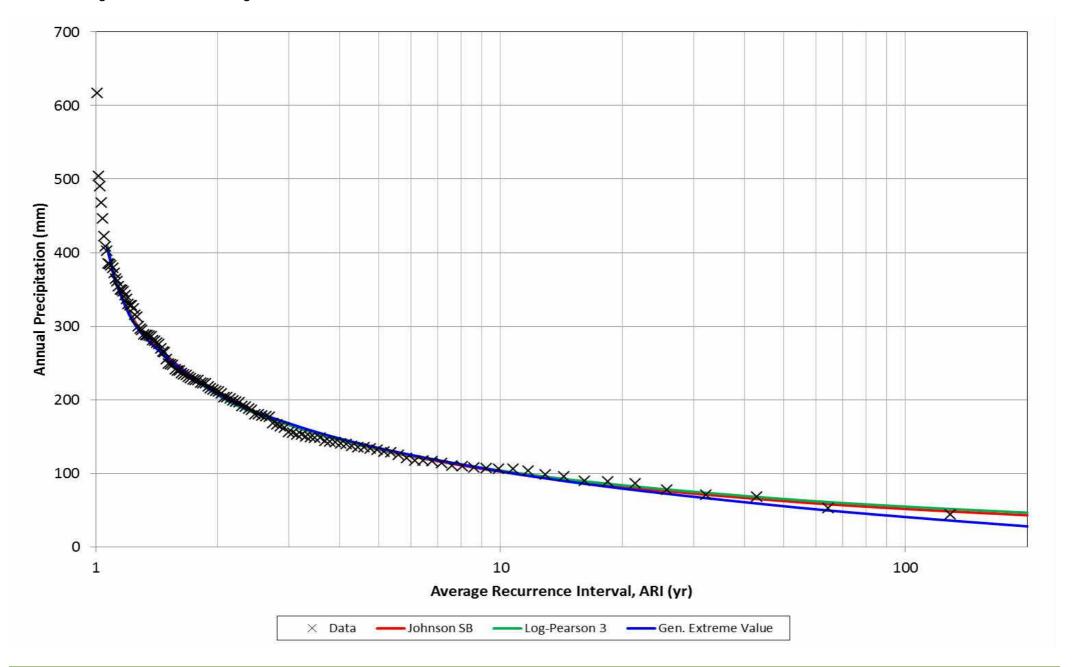
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- 2. Australian Bureau of Meteorology. "Development of the *Generalised Southeast Australia Method for Estimating Probable Maximum Precipitation, HRS Report No. 4"*, 1996. Melbourne, VIC. Australia
- 3. Luke G.J., Burke K.L., O'Brien T.M. (1987), "Evaporation Data for Western Australia". Department of Agriculture and Food, Western Australia. Report 65.

FIGURES











APPENDIX B TSF Expansion CPT Investigation - 2020





MEMORANDUM

To: Saracen Minerals Holdings Ltd

Our Ref: PE20-01517

KP File Ref.: PE801-00296/25-A LL M20005

From:

RE: THUNDERBOX GOLD PROJECT – 2020 TSF CELLS A AND B EXPANSION GEOTECHNICAL INVESTIGATION

Knight Piésold Pty Limited (KP) was requested by Saracen Minerals Holdings Ltd (Saracen) to undertake a geotechnical site investigation of proposed TSF Cells A and B at the Thunderbox Gold Mine, Western Australia.

The site investigation was undertaken from the 2nd to the 8th of September 2020. This memorandum presents the factual data obtained from the geotechnical investigation and the associated interpretation of the ground conditions at the TSF Cells A and B.

1. SITE CHARACTERISTICS

1.1 INTRODUCTION

The Thunderbox Gold Mine is located 45 km south of the town of Leinster in Western Australia and immediately adjacent to the sealed Goldfields Highway.

1.2 TOPOGRAPHY AND SETTING

The existing TSF cells are bounded to the east by the Goldfields Highway and to the immediate west by the Eastern Waste Dump.

The surface of the tailings at TSF Cells A and B are generally flat, with the ground surface tapering up to the embankment level around the edges of the TSFs. The tailings surface is generally quite dry and no sludge/ tailings slurry is present. It was noted that the more recent deposits of the tailings slurry were deposited at the west side of TSF Cell A.

1.3 CLIMATE

The region is in a semi-arid environment and has a dry climate with hot summers and cool winters. The average annual rainfall is about 224 mm. February and March tend to be the wettest months, September and October are the driest months (Ref. 1).

1.4 GEOLOGY

The 1:250,000 geological map of Leonora (Ref. 2) indicates that the site is founded on igneous rock which is overlain by colluvial (Cza) and possibly alluvial (Qa) deposits:

- Qa: alluvial clay, silt, sand and gravel. The site is located at the uppermost reach of the drainage line and alluvial material is expected to be limited or absent.
- Cza: colluvial clay, silt and sand (pebbly in places).







Figures 1.1 and 1.2 present the geological map and map key.

1.5 SITE COORDINATE SYSTEM

The coordinates used in this memorandum are to UTM WGS84 Zone 51J (north) grid whereas Thunderbox uses a site grid. As the horizontal difference between the two grid systems is less than 1 m, the CPT locations were set out using a hand held GPS to approximately +/- 5 m accuracy, no grid conversion has been used.

2. SITE INVESTIGATION

2.1 INTRODUCTION

A geotechnical investigation was undertaken to assess strength of the existing tailings to provide design parameters for foundation and earthwork design. The scope of work focussed on conducting cone penetrometer tests (CPTs) at the existing TSF Cells A and B to assess the tailings strengths and behaviour.

The scope of work was carried out during September 2020 and comprised:

- Eight cone penetrometer tests with dissipation tests to assess excess pore pressures within the existing TSF cells.
- Collection of undisturbed samples of soils and tailings material for subsequent laboratory testing.

Fieldwork was supervised by a geotechnical engineer from KP Perth. The test locations were pre-determined and set out on site by a Saracen surveyor and verified by the geotechnical engineer, prior to fieldwork.

The site investigation was undertaken in accordance with Australian Standard AS1726–2017 (Ref. 3), where practicable.

Figure 2.1 shows the site investigation plan.

2.2 CONE PENETROMETER TESTING

Eight cone penetrometer tests (CPTu) were conducted to depths between 13.7 m and 18.5 m using a Morooka mounted CPT rig supplied and operated by Hagstrom Drilling.

The CPT tip was fitted with a pore pressure sensor (piezocone) to measure the water pressure with depth, in addition to tip resistance and friction.

The CPTu were conducted at the surveyed locations with the exception of KP CPT05 and KP CPT07 which were not located on the access track on the top of the embankment at TSF Cell B. Both locations were subsequently moved further away from the embankment wall and their new positions were closer to the centre of the TSF than KP CPT06 and KP CPT08 respectively.

A summary of the CPTu investigation is presented in Table 2.3.

CPT traces and photographs are presented in Appendices C and D respectively. The CPTu locations are shown in Figure 2.1.



Table 2.3: Summary of cone penetrometer tests

CPT ID	Location	Elevation (m RL)	Final Depth (m)
KP-CPT01	TSF Cell A	509.2	16.9
KP-CPT02	TSF Cell A	508.7	16.3
KP-CPT03	TSF Cell A	509.2	18.4
KP-CPT04	TSF Cell A	508.7	18.0
KP-CPT05A	TSF Cell B	503.3	14.1
KP-CPT06	TSF Cell B	504.2	13.7
KP-CPT07A	TSF Cell B	503.3	15.7
KP-CPT08	TSF Cell B	504.2	17.5

Twelve dissipation tests were carried out to assess the rate of dissipation of the excess pore pressures generated from the CPTu. The dissipation tests are summarised in Table 2.4.

Table 2.4: Summary of dissipation tests

CPT ID	Test Depth(s) (m)
KP-CPT01	12.0
KP-CPT02	9.1, 12.0, 15.0
KP-CPT03	None
KP-CPT04	10.5, 13.5, 16.5
KP-CPT05A	7.0, 12.0
KP-CPT06	None
KP-CPT07A	6.2, 11.5
KP-CPT08	None

Seven undisturbed push tube/piston tube samples and five disturbed Mostap samples were taken for laboratory testing to estimate in situ strength and classification for the tailings materials encountered during the CPTu investigation. The piston and Mostap sample test details are summarised in Table 2.5.



	Table 2.5:	Summary	of cone	penetrometer	samples
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CPT ID	Sample type	Final Depth (m)
KP-CPT02	Piston	3.5 - 4.0
KP-CPT02	Piston	10.5 - 11.0
KP-CPT02	Piston	14.5 - 15.0
KP-CPT04	Piston	6.0 - 6.5
KP-CPT04	Piston	13.0 - 13.5
KP-CPT05A	Piston	10.0 - 10.5
KP-CPT05A	Mostap	10.0 - 10.5
KP-CPT06	Mostap	8.5 - 9.0
KP-CPT07A	Piston	12.0 - 12.5
KP-CPT07A	Mostap	5.5 - 6.0
KP-CPT08	Mostap	4.0 - 4.5
KP-CPT08	Mostap	13.0 - 13.5

3. INTERPRETATION OF CPT TEST RESULTS

Three parameters were measured during cone penetration testing:

- Cone tip resistance.
- Cone sleeve friction resistance.
- Pore-water pressure.

In addition, pore-water pressure dissipation tests were undertaken at various locations.

Interpretation of test results was performed using the GeoLogismiki CPT bundle (CPeT-IT and Cliq, Ref. 4), which was primarily based on Robertson's CPT interpretation method (Ref. 5). The methodology and interpretation details are discussed below and include:

- Phreatic surface level and pore water pressure.
- Material classification.
- Soil parameters including friction angles (for dilative/drained material), undrained shear strengths (for contractive material), permeability, soil modulus and state parameter/ flow liquefaction etc.
- Liquefaction potential.

Cut sections based on the contractive and dilative behaviours (Table 3.1) of each of the cells are presented in figures 3.1 to 3.4.



Table 3.1: Interpreted Dilative and Contractive Behavi	<i>i</i> our Horizon
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Cell	CPT ID	Dilative/Contractive Behaviour Horizon Depth* (m)
	KP-CPT01	6.5
А	KP-CPT02	6.0
	KP-CPT03	4.6
	KP-CPT04	3.5
	KP-CPT05A	4.0
В	KP-CPT06	3.0
	KP-CPT07A	2.7
	KP-CPT08	2.0

^{*}Tailings above the depth is generally dilative and below it contractive.

3.1 PHREATIC LEVEL

Dynamic pore-water pressure measurements are recorded during site CPT in-situ probing. Positive pore-water pressure readings throughout the tailings indicate layers at, or near, saturation. Negative pore-water pressure readings indicate interbedded unsaturated layers (possibly capillary head).

Pore-water pressure dissipation tests (PPDTs) were performed at various depths during the probing programme to provide a better understanding of phreatic surface conditions. Under hydrostatic conditions the pore pressure during each PPDT will tend to stabilise, thereby providing a measurement of hydrostatic pressure at the test depth, and hence an approximation of the phreatic surface. The phreatic surface depths derived from the PPDTs are presented in Table 3.2.

 Table 3.2: Interpreted Phreatic Surface Depth

Cell	CPT ID	Pore Pressure Dissipation Test Depth (m)	Interpreted Phreatic Depth (m)
	KP-CPT01	12.0	10.2
В	KP-CPT02	9.1, 12.0, 15.0	10.2
	KP-CPT03	None	10.2
	KP-CPT04	10.5, 13.5, 16.5	8.8
	KP-CPT05A	7.0, 12.0	7.6
	KP-CPT06	None	5.8
	KP-CPT07A	6.2, 11.5	7.9
	KP-CPT08	None	9.3

The pore-water pressure dissipation tests indicate a variable phreatic level of between 5.8 m and 10.5 m below tailings surface. The phreatic surface depths are generally consistent across each TSF cell.



3.2 MATERIAL CLASSIFICATION

The material classification was based on the soil behaviour types (SBT - non-normalised or SBTn - normalised) recommended by Robertson (1986, Ref. 6; 1990, Ref. 7; 2009, Ref. 5) and Schneider et.al.(2008, Ref. 8), as presented in Table 3.3. The classification plots (Normalised SBTn and Schneider method) are included in Appendix B.

In the normalised SBTn assessment, the approximate boundary of soil behaviour types is given in terms of the SBTn index, I_c . The approximate boundary between sand-like and clay-like behaviour is about I_c = 2.6. This value was adopted to define sand-like and clay-like behaviour.

Table 3.3: Material Classification (Robertson 1986, 1990 and 2009)

Soil Type	Description	Ic (SBTn index)
1	Sensitive, fine grained	N/A
2	Organic soils, peats	>3.6
3	Clay - Clay to silty clay	2.95 – 3.6
4	Silt mixtures - Clayey silt to silty clay	2.6 – 2.96
5	Sand mixture - Silty sand to sandy silt	2.05 – 2.6
6	Sand – Clean sand to silty sand	1.31 – 2.05
7	Gravelly sand to dense sand	<1.31
8	Very stiff sand to clayey sand	N/A
9	Very stiff fine grained	N/A

Table 3.4 lists the interpreted depths of sand-like tailings and clay-like tailings throughout each probing. From the investigation it is apparent that the tailings in TSF Cell A is more sand-like, in comparison to TSF Cell B which is more clay-like in behaviour.

From the normalised friction ratio (R_f) and the normalised cone resistance (Q_t) chart by Robertson (2009) the tailings are interpreted to mainly comprise:

- "Sand Clean sand to silty sand" (6).
- "Sand mixture Silty sand to sandy silt" (5).
- "Silt mixtures clay silt to silty clay" (4).

Horizons that were identified as "Clay - Clay to Silty Clay" (3) were also encountered at depth.



Table 3.4:	Subsurface	Conditions
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TSF Cell	CPT ID	Depth Range (m)	Material Behaviour		
	KP-CPT01	0.0 - 9.0	Sand like		
	KP-CP101	9.0 – 16.9	Clay/Silt like		
	KP-CPT02	0.0 - 8.7	Sand like		
Α	KP-CP102	8.7 – 16.3	Clay/Silt like		
A	KP-CPT03	0.0 - 5.6	Sand like		
	KF-CF103	5.6 – 18.4	Clay/Silt like		
	KP-CPT04	0.0 - 8.0	Sand like		
	KP-CP104	8.0 – 18.0	Clay/Silt like		
	KP-CPT05A	0.0 - 3.8	Sand like		
	RF-CF 105A	3.8 – 14.1	Clay/Silt like		
	KP-CPT06	0.0 - 3.8	Sand like		
В	KP-CP100	3.8 – 13.7	Clay/Silt like		
Б	KP-CPT07A	0.0 – 3.1	Sand like		
	KP-CP10/A	3.1 – 15.7	Clay/Silt like		
	KP-CPT08	0.0 - 4.0	Sand like		
	KF-0F100	4.0 – 17.5	Clay/Silt like		

3.3 MATERIAL PARAMETERS

3.3.1 State Parameters

Fine grained soils that develop high pore-water pressure at low strain, and loose sand-like materials that contract during shear, are most prone to strain softening and flow liquefaction. Flow liquefaction occurs when materials exhibiting brittle behaviour are subjected to rapid large loads i.e. there is insufficient time for pore pressure to dissipate. This may occur during upstream embankment construction if construction rates are excessive, or during deposition if rates of rise are too high.

Flow liquefaction is, broadly speaking, the cause of failure for both the Samarco (2015) and Brumadinho (2019) tailings dam failures in Brazil. Following these two failures ANCOLD published revised guidelines (July, 2019) addressing, amongst other things, flow liquefaction assessments for upstream constructed tailings storages.

The state parameter, which is defined as the difference between the in situ void ratio and the void ratio at critical state, at the same mean effective stress, is used to infer the susceptibility of materials to strain softening behaviour and flow liquefaction. Material with a state parameter of less than -0.05 (Ψ < -0.05) are dilative and will strain harden during undrained shear (i.e. Dilative material is not prone to flow liquefaction, which is good). Conversely, materials with a state parameter greater than -0.05 (Ψ > -0.05) are contractive and will strain soften during undrained shear, and may be susceptible to flow liquefaction (i.e. Contractive material is prone to flow liquefaction, which is bad).

The interpretations of the state parameters are based on the following method:

Robertson and Cabal (2015) (Ref. 9).



The results indicate that the profiles typically comprise:

- A top tailings layer with typical dilative behaviour (Ψ < -0.05), in the unsaturated zone above the phreatic surface.
- A contractive tailings layer with $\Psi > -0.05$.

The profiles of dilative and contractive layers are shown in figures 3.1 to 3.4.

The peak friction angles (ϕ') were estimated for the sand-like primary tailings, according to (Ref. 9):

$$\varphi' = \varphi'_{cv} + 15.84 [\log Q_{tn,cs}] - 26.88 \tag{3.1}$$

Where φ'_{cv} is the constant volume (or critical state) friction angle:

 $Q_{tn,cs}$ is the clean sand equivalent normalized cone resistance.

The assumed value of φ'_{cv} = 33° (for the tailings) was used in Equation 3.1.

Coffey Mining conducted a geotechnical investigation in 2014 (Coffey Report Ref. No. MINEWPER01227AA, Ref 10) and found that the friction angles for the deposited tailings ranged between 30 to 45°. The interpreted peak friction angle ranges for the sand-like tailings layer for each of the CPT are listed in Table 3.5. The friction angle is higher for the CPTs located in TSF Cell A, which is more sand-like than the CPTs located in TSF Cell B.

 Table 3.5:
 Interpreted Peak Friction Angles for Sand-like Primary Tailings

TSF Cell	CPT ID	Depth Range (m)	Friction Angle φ' (°)		
	KP-CPT01	0.0 - 9.0	32 – 45		
A	KP-CPT02	0.0 - 8.7	32 – 40		
A	KP-CPT03	0.0 - 5.6	32 – 44		
	KP-CPT04	0.0 - 8.0	32 – 40		
	KP-CPT05A	0.0 - 3.8	32 – 40		
В	KP-CPT06	0.0 - 3.8	32 – 42		
Б	KP-CPT07A	0.0 – 3.1	32 – 38		
	KP-CPT08	0.0 - 4.0	32 – 39		

3.3.2 State Parameters and Tailings Undrained Shear Strength

Layering for each material was determined based on the 80^{th} percentile values of state parameters. Material with a state parameter less than -0.05 was deemed to be dilative and material greater than -0.05 was deemed to be contractive. The undrained shear strength ratio was estimated using the total cone resistance (q_t) and a cone factor (N_{kt}). The cone factor, N_{kt} , typically varies between 10 and 18, with an average of 14. A cone factor of 14 was applied in this assessment.

In deriving the residual shear strength ratio, two methods were used:

- Method 1 Based on CPTe-IT interpretation
- Method 2 Robertson 2020 updated method.



Method 1 evaluates the residual shear strengths using the embedded method in the CPTe-IT 3.0.1.17 programme. Method 2 is an evaluation method updated by Robertson 2020 (Ref. 11).

The mean and 80th percentile values of state parameters and undrained shear strength ratios are listed in Table 3.6.

Table 3.6: Interpreted State Parameters and Undrained Shear Strength Ratios for Clay-like Tailings

CPT Location	Depth Range (m)		State Parameter - I _C Cutoff		Strengtl	Undrained Shear Strength Ratio - Peak		nod 1 - idual, lded/Sen ivity	Method 2 - Residual, Robertson 2020	
	From	То	Mean	80th	Mean	80th	Mean	80th	Mean	80th
	0.0	5.8	-0.20	-0.06						
KP CPT01	5.8	7.7	-0.02	0.00	0.94	0.86	0.15	0.12	0.17	0.09
KF CF IUI	7.7	8.0	-0.09	-0.07			Dila	ative		
	8.0	16.9	0.00	0.02	0.52	0.32	0.14	0.09	0.18	0.07
	0.0	6.6	-0.08	-0.05			Dila	ative		
KP CPT02	6.6	10.4	-0.03	-0.02	0.99	0.83	0.23	0.21	0.24	0.13
	10.4	16.2	0.02	0.03	0.36	0.29	0.12	0.09	0.12	0.09
	0.0	6.0	-0.17	-0.06			Dila	ative		
KP CPT03	6.0	10.4	-0.02	-0.01	0.98	0.66	0.22	0.15	0.17	0.10
	10.4	18.4	0.03	0.04	0.39	0.30	0.09	0.07	0.09	0.06
	0.0	5.8	-0.08	-0.05			Dila	ative		
KP CPT04	5.8	6.7	-0.03	-0.02	1.18	0.84	0.29	0.24	0.33	0.18
KF CF104	6.7	8.1	-0.06	-0.05	Dilative					
	8.1	18.0	0.01	0.02	0.45	0.30	0.13	0.10	0.13	0.09
	0.0	3.9	-0.13	-0.04			Dila	ative		
KP CPT05	3.9	7.5	-0.03	-0.01	0.84	0.52	0.24	0.16	0.26	0.12
	7.5	14.2	0.03	0.04	0.43	0.33	0.09	0.06	0.08	0.04
	0.0	3.4	-0.13	-0.06			Dila	ative		
KP CPT06	3.4	6.2	-0.03	0.00	0.83	0.62	0.22	0.14	0.49	0.14
	6.2	13.6	0.01	0.03	0.52	0.40	0.13	0.08	0.12	0.06
	0.0	3.4	-0.11	-0.06			Dila	ative		
KP CPT07	3.4	6.4	-0.03	-0.01	0.87	0.46	0.29	0.15	0.51	0.14
AF OF 107	6.4	7.4	-0.09	-0.07			Dila	ative		
	7.4	15.7	0.00	0.02	0.59	0.42	0.13	0.09	0.12	0.07
KP CPT08	0.0	9.3	-0.08	-0.04			Dila	ative		
KF CF 106	9.3	17.5	0.00	0.02	0.50	0.36	0.14	0.10	0.14	0.09

The undrained shear strength varies with the state parameters significantly. A few peak undrained shear strength ratios appear to be too high, possibly influenced by thin sand lenses in the claylike layers, which will be capped at the drained shear strengths for the stability assessment.

The tailings appear to be brittle, the residual shear strengths are relatively low.



3.3.3 Permeability

Permeability was interpreted from the dissipation tests using coefficient of consolidation values calculated using Houlsby and Teh's (1988, Ref. 12) theory:

$$c_h = \frac{T \times r^2 \times I_r^{0.5}}{t_{50}}$$

$$k_h = \frac{c_h \times \gamma_w}{M}$$

Where:

- T: time factor given by Houlsby and Teh (1988).
- r: piezocone radius.
- Ir: stiffness index, shear modulus G divided by the undrained strength of clay.
- t₅₀: time corresponding to 50% consolidation.
- M: 1-D constrained modulus.
- y_w is the unit weight of water.

The dissipation results of the Coffey investigation (Ref. 10), indicated that the typical permeability coefficients for Cell A range from 2.5 x 10^{-8} to 8.1 x 10^{-8} m/s and for Cell B 9.5 x 10^{-9} to 2.1 x 10^{-8} m/s. Table 3.7 provides the estimated permeability with results indicating values ranging between 1.03×10^{-8} m/s and 1.66×10^{-7} m/s for TSF Cell A, and 3.93×10^{-9} m/s to 1.36×10^{-7} m/s for Cell B. Thus, the results are consistent with the Coffey investigation in 2014.

Table 3.7: Interpreted Permeability

TSF Cell	CPT ID	Dissipation Depth (m)	Permeability (m/s)		
	KP-CPT01	12.0	2.86 x 10 ⁻⁸		
		9.1	1.03 x 10 ⁻⁸		
	KP-CPT02	12.0	3.53 x 10 ⁻⁸		
Α		15.0	1.66 x 10 ⁻⁷		
		10.5	7.48 x 10 ⁻⁸		
	KP-CPT04	13.5	1.16 x 10 ⁻⁷		
		16.5	8.81 x 10 ⁻⁸		
	KP-CPT05A	7.0	3.93 x 10 ⁻⁹		
В	KP-CP105A	12.0	1.36 x 10 ⁻⁷		
В	KP-CPT07A	6.2	6.18 x 10 ⁻⁸		
	KF-CPTU/A	11.5	4.09 x 10 ⁻⁸		



1. LABORATORY TESTING

Representative samples of the in-situ soils in the tailings were taken from the boreholes for laboratory testing. The purpose of the laboratory testing was to classify and characterise the in situ materials so as to assess their behavioural characteristics under embankment loading.

The laboratory testing was undertaken by E-Precision soil lab and comprised:

- Particle size distribution.
- · Atterberg limits.
- · Linear shrinkage.
- Compaction tests.
- Falling head permeability tests.
- · Emerson tests.
- Consolidated Undrained (CU) Triaxial tests.
- Uniaxial compressive strength.

The laboratory test reports are presented in Appendix B and the test results are summarised in Table 4.1.



Table 4.1: Summary of laboratory testing results

	Location Depth (m)		epth (m) Description		A-line	Parti	cle size d	listributio	n (%)	Atterb	erg Lim	its (%)		ed Undrained est
Sample						Gravel	Sand	Silt	Clay	Н	PL	PI	c' (kPa)	φ' (deg)
	KP-CPT06	8.5 - 9.0	SILT with sand, trace clay	ML	CL	0	22	73	5	28	22	6	-	-
þe	KP-CPT05A	10.0 - 10.5	SILT with sand, trace clay	ML	-	0	25	72	3	-	-	-	3.63	36.73
Disturbed	KP-CPT08	4.0 - 4.5	SILT trace sand and clay	ML	ML	0	13	80	7	35	28	7	-	-
Dist	KP-CPT08	13.0 - 13.5	SILT with sand, trace clay	ML	-	0	21	72	7	-	-	-	-	-
	KP-CPT07A	5.5 - 6.0	SILT trace sand and clay	ML	ML	0	5	86	9	45	34	11	-	-
	KP-CPT02	3.5 - 4.0	SILT trace sand and clay	ML	ML	0	15	80	5	33	26	7	-	-
	KP-CPT02	10.5 - 11.0	-	-	-	-	-	-	-	-	-	-	1.66	38.32
pec	KP-CPT02	14.5 - 15.0	SILT trace sand and clay	ML	ML	0	14	77	9	38	31	7	-	-
Undisturbed	KP-CPT04	6.0 - 6.5	SILT with sand, trace clay	ML	ML	0	24	68	8	31	24	7	-	-
ndis	KP-CPT04	13.0 - 13.5	-	-	ML	-	-	-	-	31	23	8	8.85	26.81
	KP-CPT05A	10.0 - 10.5	-	-	CL	-	-	-	-	28	22	6	-	-
	KP-CPT07A	12.0 - 12.5	-	-	-	-	-	-	-	-	-	-	10.98	36.73

- Abbreviations:

 a. LL liquid limit

 b. PL plastic limit

 c. PI plasticity index

 d. c' effective cohesion
- ϕ^{\prime} effective stress friction angle



2. INVESTIGATION FINDINGS

The behaviour of the tailings materials in both TSF Cells A and B typically comprised:

- Sandy silt to depths of 2.5 to 6.0 m in Cell A and 1.0 to 3.0 m in Cell A.
- Clay like/Silty Clay likely to the maximum depth of investigation.

The general profile of the tailings in both Cells A and B are generally consistent in terms of the underlying stratigraphy. However, the CPTs for Cell A (CPT01 and CPT03), close to the embankment indicate a very dense layer (approximately 4 m) of sand like material, which may be the result of compaction during the construction of the embankment.

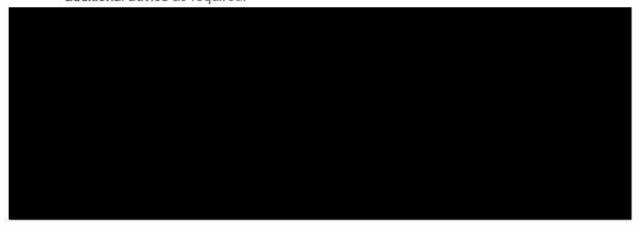
3. CONCLUSIONS AND RECOMMENDATIONS

Based on the site geotechnical investigation and laboratory test results, the following conclusions can be drawn:

- The site investigation provided the indicative depths and thickness of the tailings material layers at TSF Cell A and Cell B.
- The top 3 to 6 m dilative tailings overlies contractive tailings. Tailings are generally becoming weaker with depth. The lower part of the tailings is sensitive with relatively low residual strength.
- The tailings close to the embankments are generally dilative possibly due to traffic compaction during construction.
- The friction angle for the dilative materials ranges from 32° to 45°, with TSF Cell A having a slightly higher friction angle than TSF Cell B.
- The undrained shear strengths vary significantly with state parameters, a few
 peak undrained shear strength ratios appear to be too high, possibly
 influenced by thin sand lenses in the clay like layers. The undrained shear
 strengths need to be capped at the drained shear strengths for stability
 assessment.
- The tailings have low permeability and as such will tend to hold moisture in the tailings for a significant amount of time prior to drainage, assuming natural drainage without the aid of any underdrainage system.

The results of this investigation are being utilised in the embankment stability assessments currently being undertaken.

We trust the information provided is sufficient at this stage and will be pleased to provide any additional advice as required.

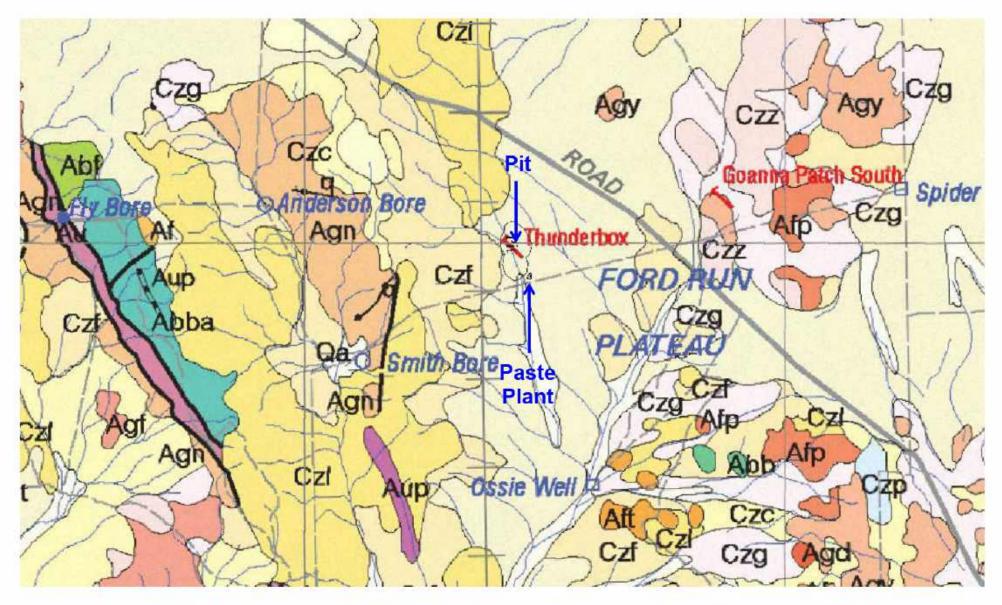




REFERENCES

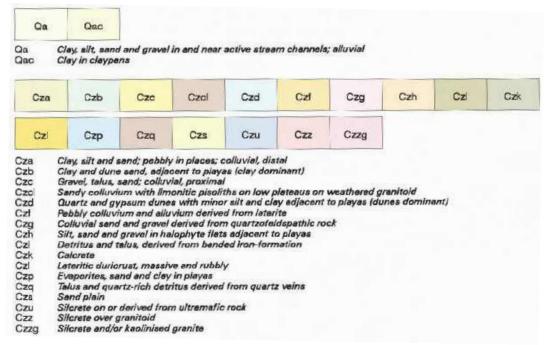
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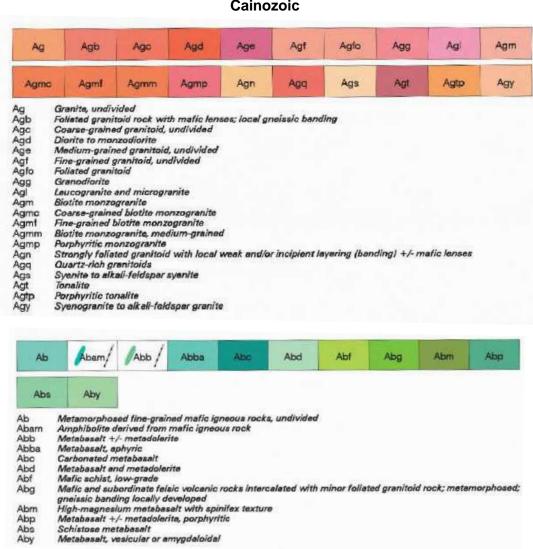


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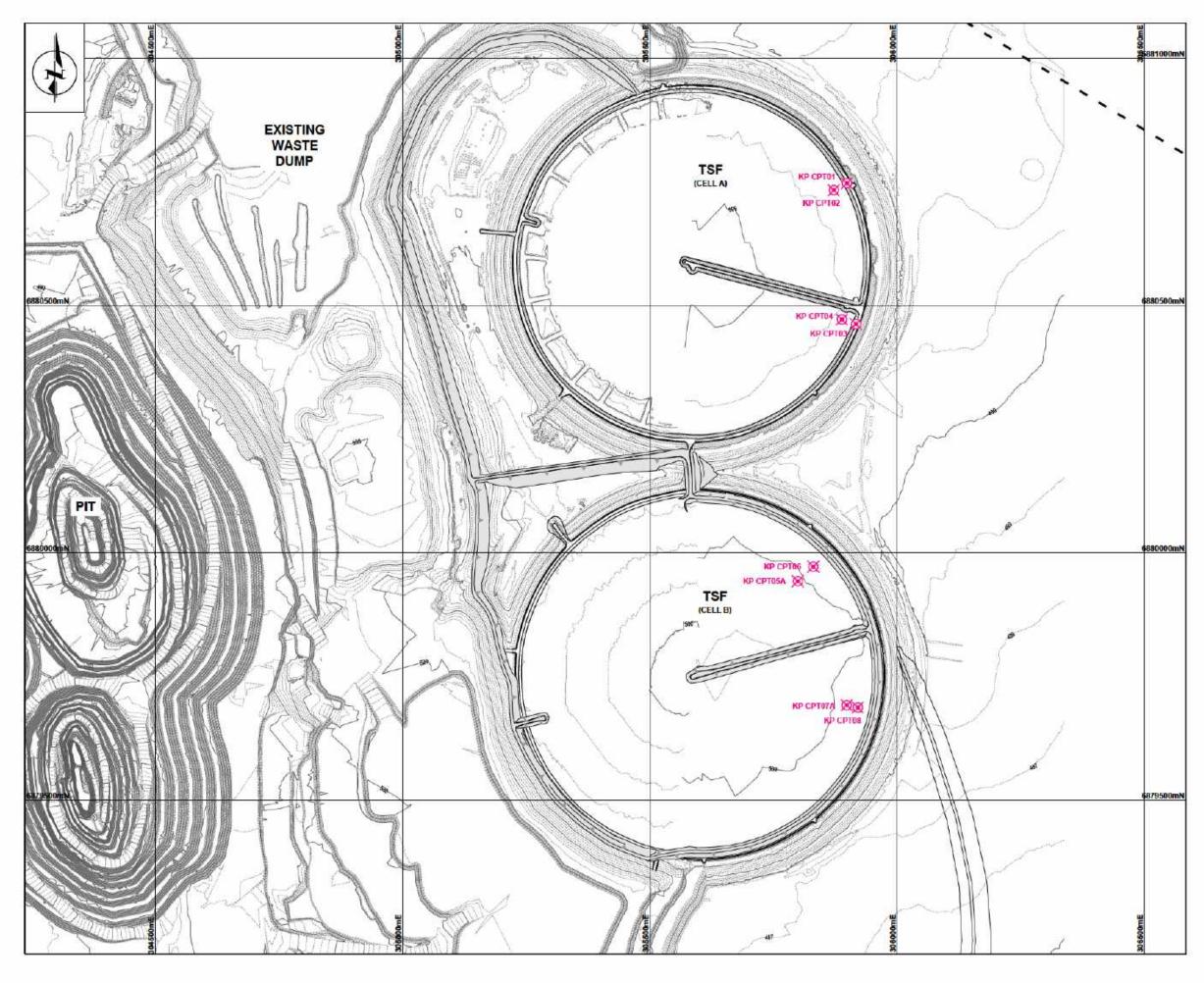


Cainozoic



Archaean





LEGEND:



CPT PROBE LOCATIONS

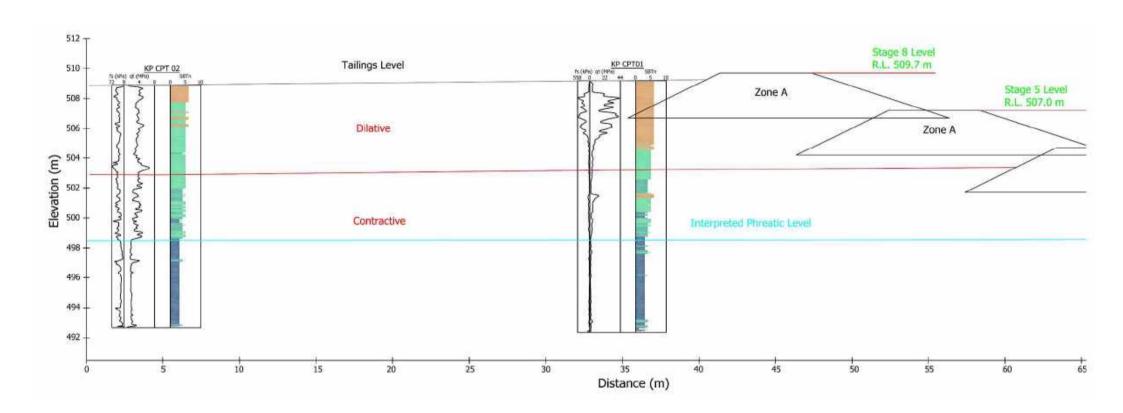
NOTES:

1. 1m CONTOUR INTERVALS SHOWN.

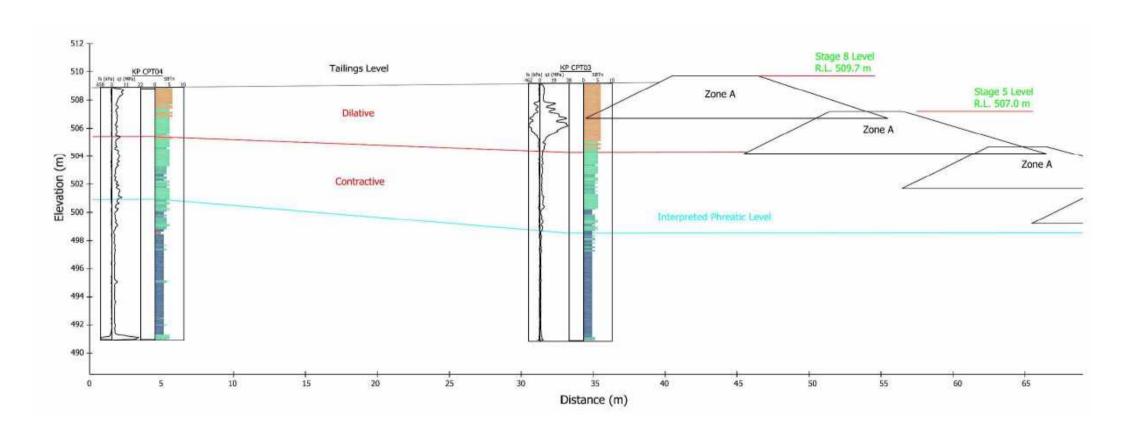
CPT LOCATIONS							
CPT LD No.	EASTING	NORTHING					
KP CPT01	305898.4	6880747.5					
KP CPT02	305871.8	6880733.7					
KP CPT03	305916.6	6880462.7					
KP CPT04	305888.1	6880472.1					
KP CPT05A	305799.0	6879943.0					
KP CPT06	305830.3	6879972.3					
KP CPT07A	305898.1	6879692.3					
KP CPT08	305920.4	6879687.6					



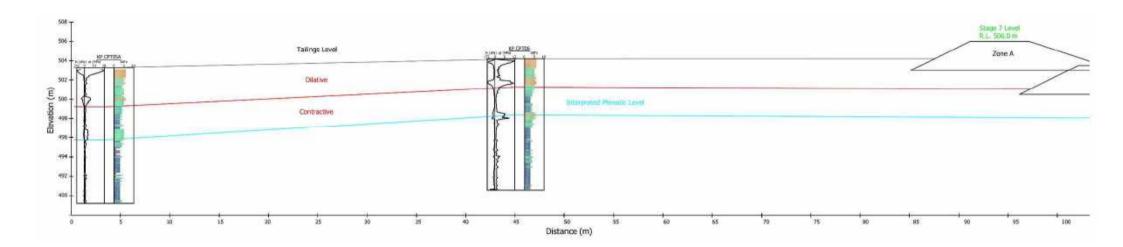




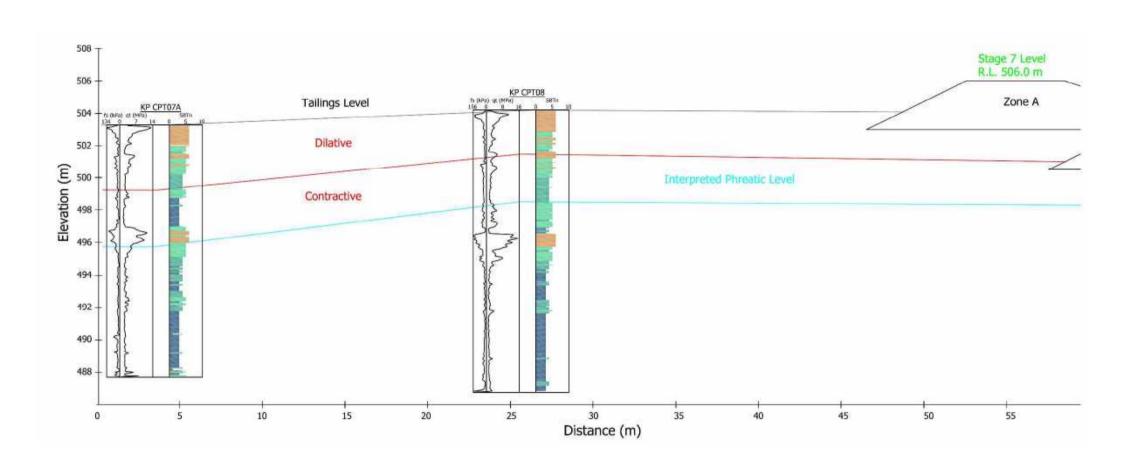














APPENDIX A CPT Results

Knight Piésold Consulting Knight Piésold Perth WA 6004, Australia http://www.knightpiesold.com/

Level 1, 184 Adelaide Terrace

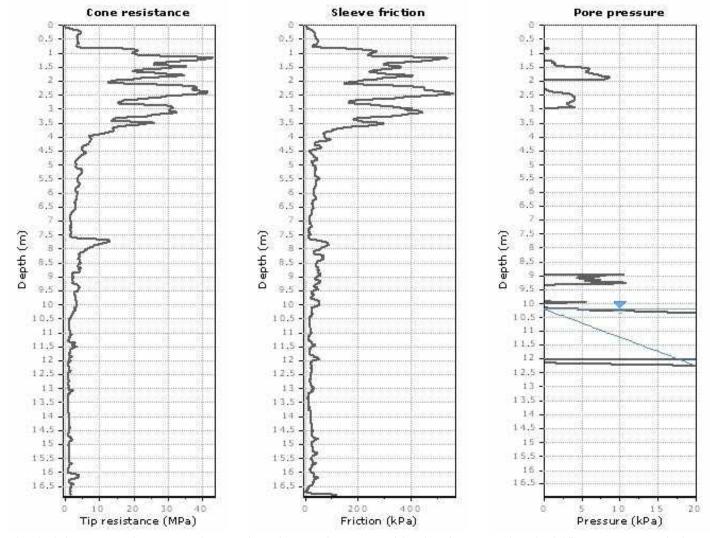
Thunderbox Gold Mine Project:

Location: Leinster, Western Australia, Australia

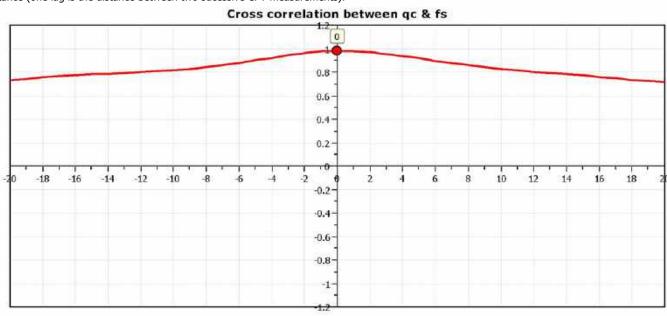
CPT: KP CPT01

Total depth: 16.88 m, Date: 3/09/2020 Coords: lat 305898.4° lon 6880747.5°

Cone Type: 180911



The plot below presents the cross correlation coeficient between the raw qc and fs values (as measured on the field). X axes presents the lag distance (one lag is the distance between two sucessive CPT measurements).



Knight Piésold

Knight Piésold Consulting

Level 1, 184 Adelaide Terrace Perth WA 6004, Australia http://www.knightpiesold.com/

Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

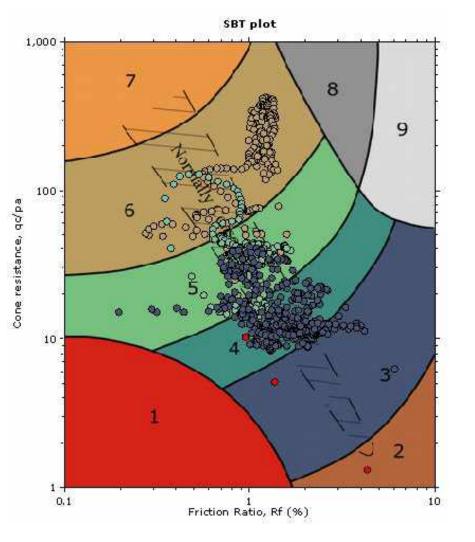
CPT: KP CPT01

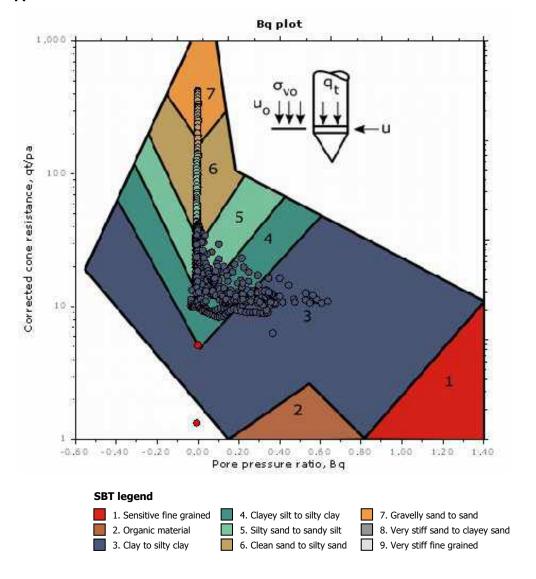
Total depth: 16.88 m, Date: 3/09/2020 Coords: lat 305898.4° lon 6880747.5°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

SBT - Bq plots





Knight Piésold

Knight Piésold Consulting

Level 1, 184 Adelaide Terrace Perth WA 6004, Australia http://www.knightpiesold.com/

Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

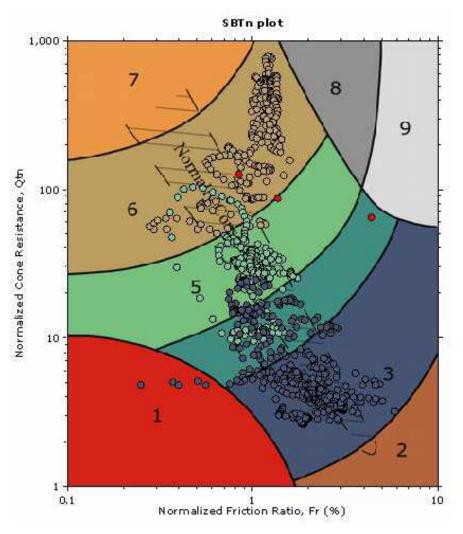
CPT: KP CPT01

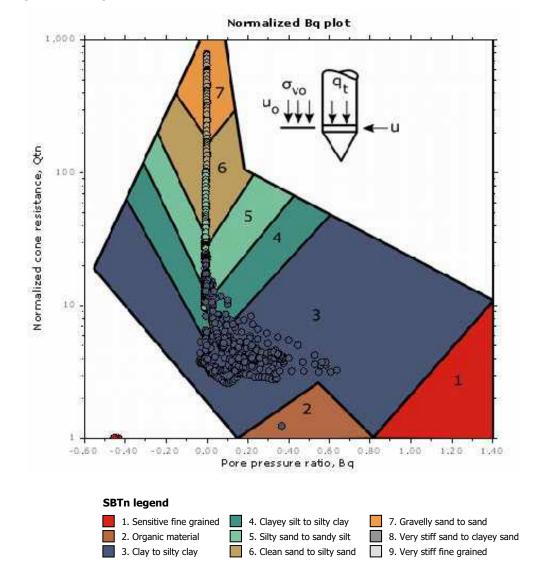
Total depth: 16.88 m, Date: 3/09/2020 Coords: lat 305898.4° lon 6880747.5°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

SBT - Bq plots (normalized)







Knight Piésold Consulting

Level 1, 184 Adelaide Terrace Perth WA 6004, Australia http://www.knightpiesold.com/

Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

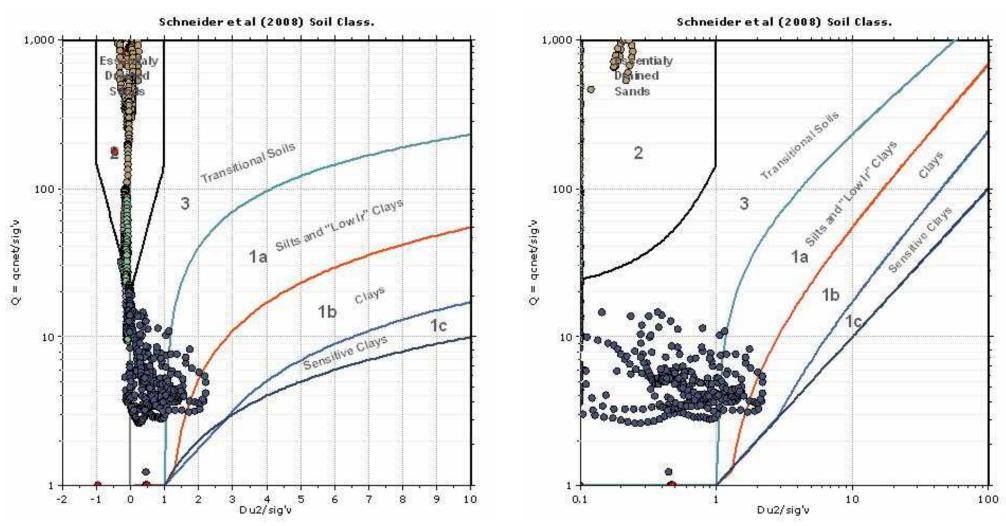
CPT: KP CPT01

Total depth: 16.88 m, Date: 3/09/2020 Coords: lat 305898.4° lon 6880747.5°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

Bq plots (Schneider)







http://www.knightpiesold.com/

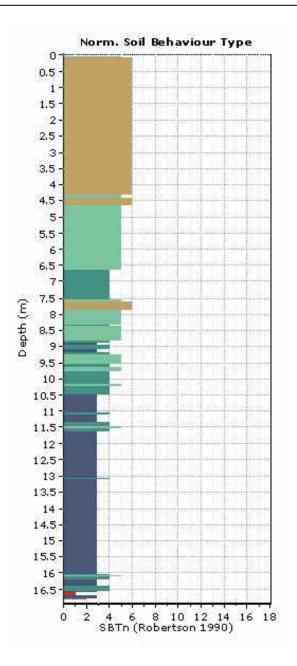
Thunderbox Gold Mine Project:

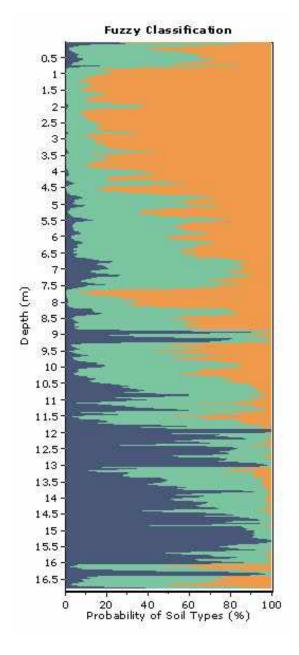
Location: Leinster, Western Australia, Australia

Total depth: 16.88 m, Date: 3/09/2020 Coords: lat 305898.4° lon 6880747.5°

Cone Type: 180911

CPT: KP CPT01





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Level 1, 184 Adelaide Terrace Perth WA 6004, Australia http://www.knightpiesold.com/

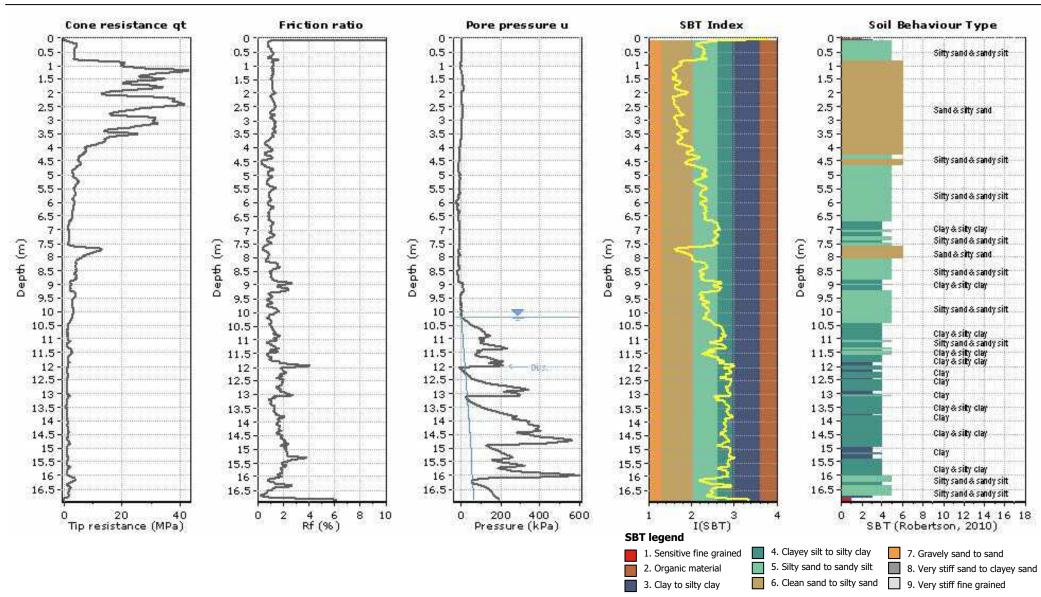
Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT01

Total depth: 16.88 m, Date: 3/09/2020 Coords: lat 305898.4° lon 6880747.5°

Cone Type: 180911



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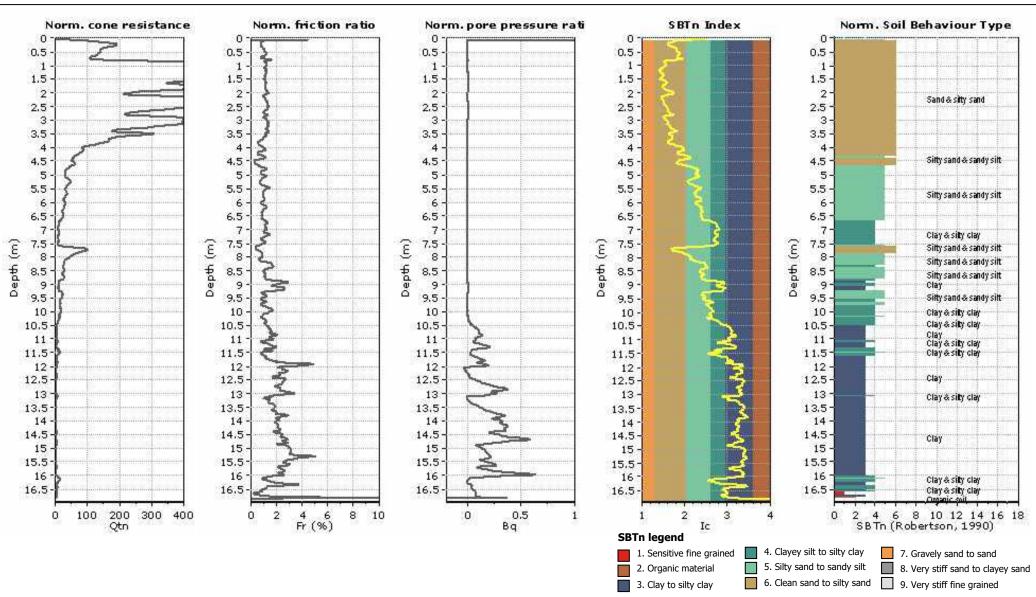
Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT01

Total depth: 16.88 m, Date: 3/09/2020 Coords: lat 305898.4° lon 6880747.5°

Cone Type: 180911



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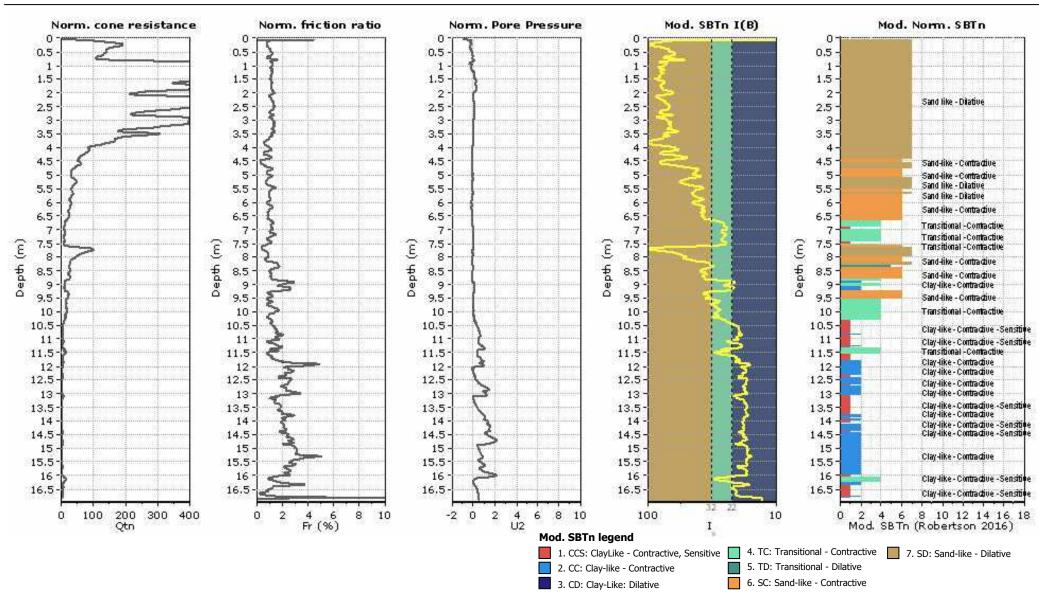
Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT01

Total depth: 16.88 m, Date: 3/09/2020 Coords: lat 305898.4° lon 6880747.5°

Cone Type: 180911



Knight Piésold Consulting

Level 1, 184 Adelaide Terrace Perth WA 6004, Australia http://www.knightpiesold.com/

Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

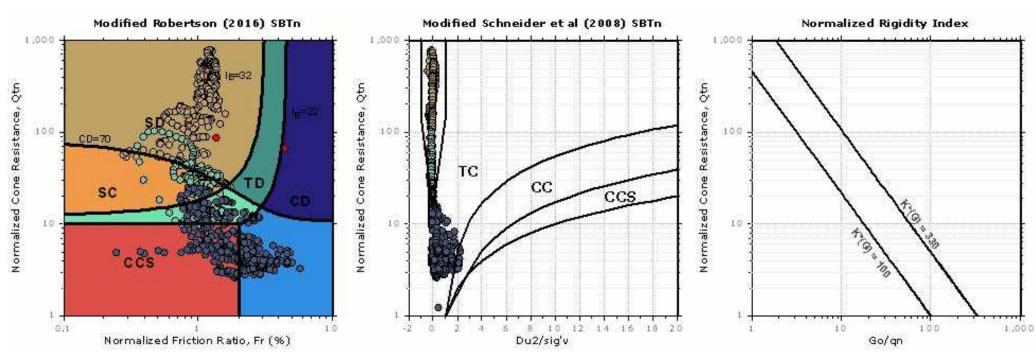
CPT: KP CPT01

Total depth: 16.88 m, Date: 3/09/2020 Coords: lat 305898.4° lon 6880747.5°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

Updated SBTn plots



CCS: Clay-like - Contractive - Sensitive

CC: Clay-like - Contractive
CD: Clay-like - Dilative
TC: Transitional - Contractive
TD: Transitional - Dilative
SC: Sand-like - Contractive

Sand-like - Dilative

K(G) > 330: Soils with significant microstructure (e.g. age/cementation)

Knight Piésold Consulting Knight Piésold

Level 1, 184 Adelaide Terrace Perth WA 6004, Australia http://www.knightpiesold.com/

Project: **Thunderbox Gold Mine**

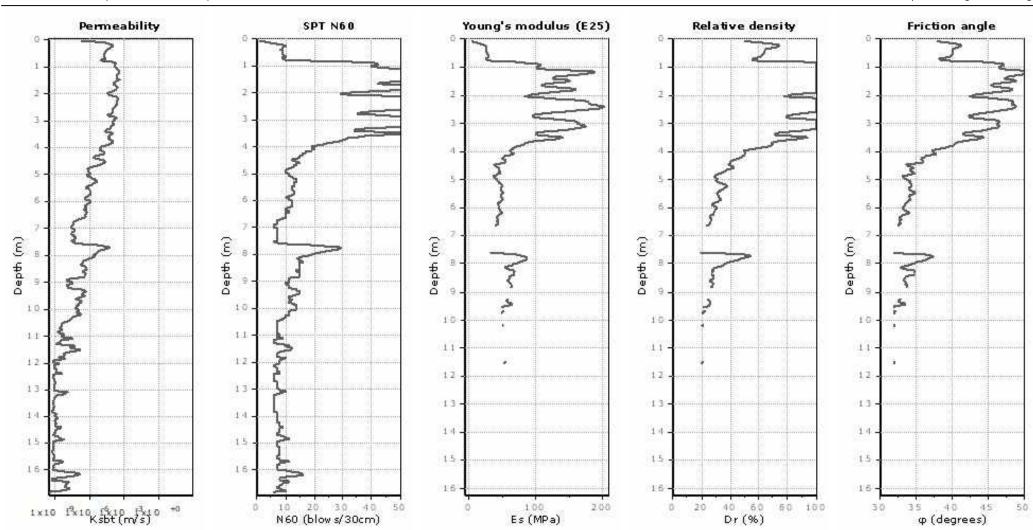
Location: Leinster, Western Australia, Australia

CPT: KP CPT01

Total depth: 16.88 m, Date: 3/09/2020 Coords: lat 305898.4° lon 6880747.5°

Cone Type: 180911

Cone Operator: Hagstrom Drilling



Calculation parameters

Permeability: Based on SBT_n SPT N₆₀: Based on I_c and q_t

Young's modulus: Based on variable alpha using I_c (Robertson, 2009)

Relative density constant, C_{Dr}: 350.0 Phi: Based on Kulhawy & Mayne (1990) ____ User defined estimation data

Knight Piésold Pert

Knight Piésold Consulting

Level 1, 184 Adelaide Terrace Perth WA 6004, Australia http://www.knightpiesold.com/

Project: Thunderbox Gold Mine

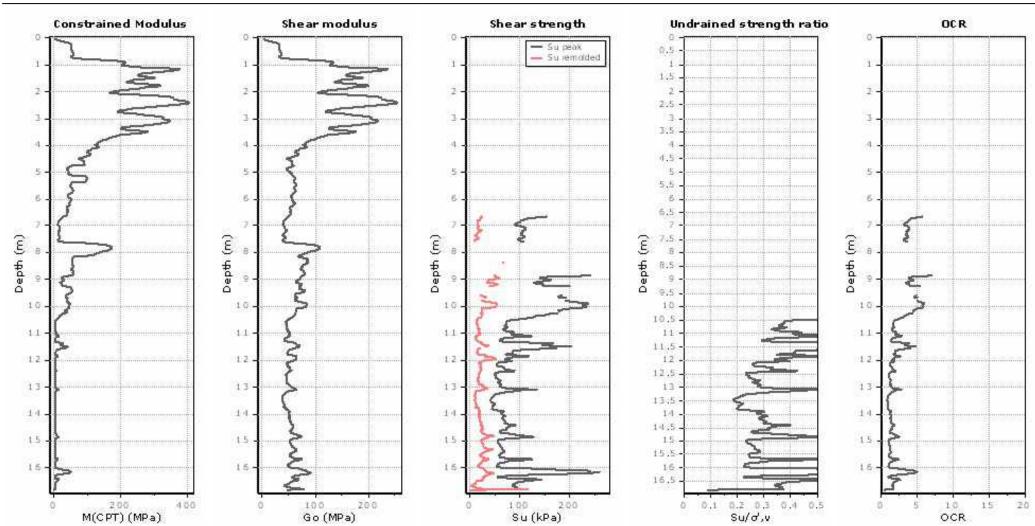
Location: Leinster, Western Australia, Australia

CPT: KP CPT01

Total depth: 16.88 m, Date: 3/09/2020 Coords: lat 305898.4° lon 6880747.5°

Cone Type: 180911

Cone Operator: Hagstrom Drilling



Calculation parameters

Constrained modulus: Based on variable alpha using $\,I_c$ and $\,Q_{tn}$ (Robertson, 2009) Go: Based on variable alpha using $\,I_c$ (Robertson, 2009)

Undrained shear strength cone factor for clays, N_{kt}: 14

OCR factor for clays, N_{kt}: 0.33

User defined estimation data

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Project: Thunderbox Gold Mine

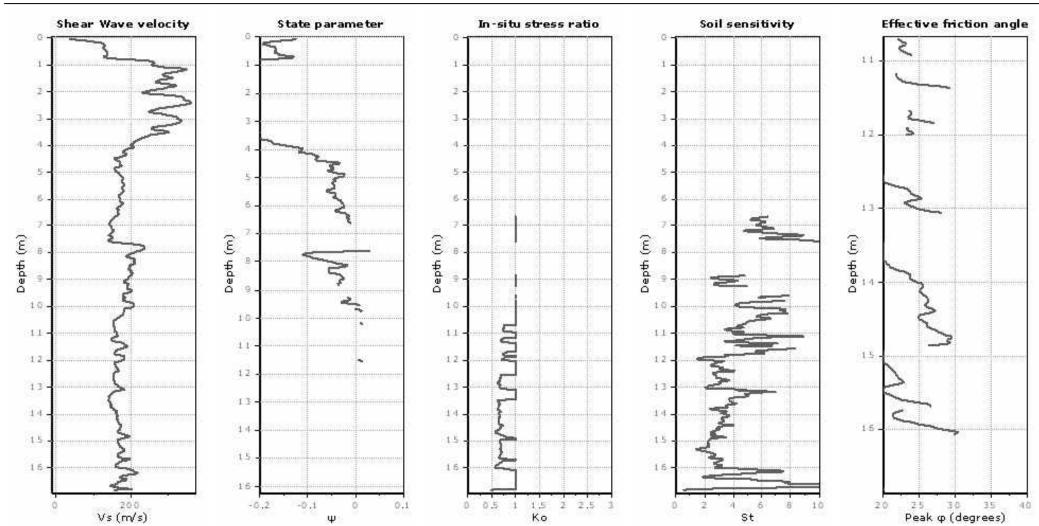
Location: Leinster, Western Australia, Australia

CPT: KP CPT01

Total depth: 16.88 m, Date: 3/09/2020 Coords: lat 305898.4° lon 6880747.5°

Cone Type: 180911

Cone Operator: Hagstrom Drilling



Calculation parameters

Soil Sensitivity factor, N_S: 7.00

User defined estimation data

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Total depth: 16.88 m, Date: 3/09/2020 Coords: lat 305898.4° lon 6880747.5°

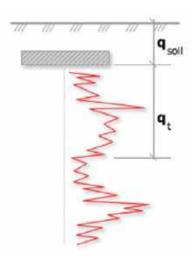
Cone Type: 180911

CPT: KP CPT01

Cone Operator: Hagstrom Drilling

Thunderbox Gold Mine Project:

Location: Leinster, Western Australia, Australia



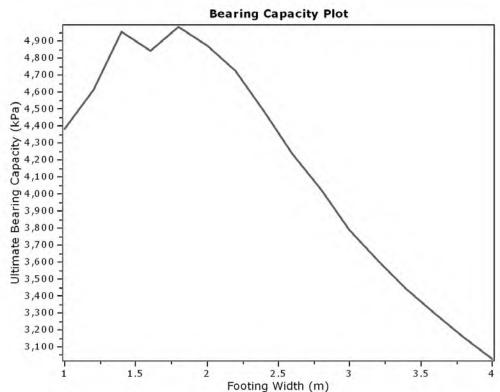
Bearing Capacity calculation is perfromed based on the formula:

$$Q_{ult} = R_k \times q_t + q_{soil}$$

where:

R_k: Bearing capacity factor qt: Average corrected cone resistance over calculation depth $q_{\text{soil}} \text{:} \text{ Pressure applied by soil}$

above footing



:: Tabu	lar results ::						
No	B (m)	Start Depth (m)	End Depth (m)	Ave. q _t (MPa)	R _k	Soil Press. (kPa)	Ult. bearing cap. (kPa)
1	1.00	0.50	2.00	21.88	0.20	9.50	4386.02
2	1.20	0.50	2.30	23.03	0.20	9.50	4616.32
3	1.40	0.50	2.60	24.73	0.20	9.50	4955.07
4	1.60	0.50	2.90	24.18	0.20	9.50	4844.71
5	1.80	0.50	3.20	24.88	0.20	9.50	4985.03
6	2.00	0.50	3.50	24.31	0.20	9.50	4871.27
7	2.20	0.50	3.80	23.59	0.20	9.50	4727.81
8	2.40	0.50	4.10	22.38	0.20	9.50	4485.75
9	2.60	0.50	4.40	21.15	0.20	9.50	4238.73
10	2.80	0.50	4.70	20.09	0.20	9.50	4026.81
11	3.00	0.50	5.00	18.91	0.20	9.50	3790.74
12	3.20	0.50	5.30	17.99	0.20	9.50	3608.11
13	3.40	0.50	5.60	17.17	0.20	9.50	3443.10
14	3.60	0.50	5.90	16.42	0.20	9.50	3294.20
15	3.80	0.50	6.20	15.75	0.20	9.50	3158.75
16	4.00	0.50	6.50	15.11	0.20	9.50	3032.04

Total depth: 16.88 m, Date: 3/09/2020 Coords: lat 305898.4° lon 6880747.5°

Cone Type: 180911

CPT: KP CPT01

Cone Operator: Hagstrom Drilling

Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

Dissipation Tests Results

Dissipation tests

Dissipation tests consists of stopping the piezocone penetration and observing porepressures (u) with elapsed time (t). The data are automatic recorded by the field computer and should take place until a minimum of 50% dissipation.

The porepressures are plotted as a function of square root of (t). The graphical technique suggested by Robertson and Campanella (1989), yields a value for t_{50} , which corresponds to the time for 50% consolidation.

The value of the coefficient of consolidation in the radial or horizontal direction c_h was then calculated by Houlsby and Teh's (1988) theory using the following equation:

$$c_h = \frac{T \times r^2 \times I_r^{0.5}}{t_{50}}$$

where:

T: time factor given by Houlsby and Teh's (1988) theory corresponding to the porepressure position

r: piezocone radius

I.: stiffness index, equal to shear modulus G divided by the undrained strength of clay (S_u).

t₅₀: time corresponding to 50% consolidation

Permeability estimates based on dissipation test

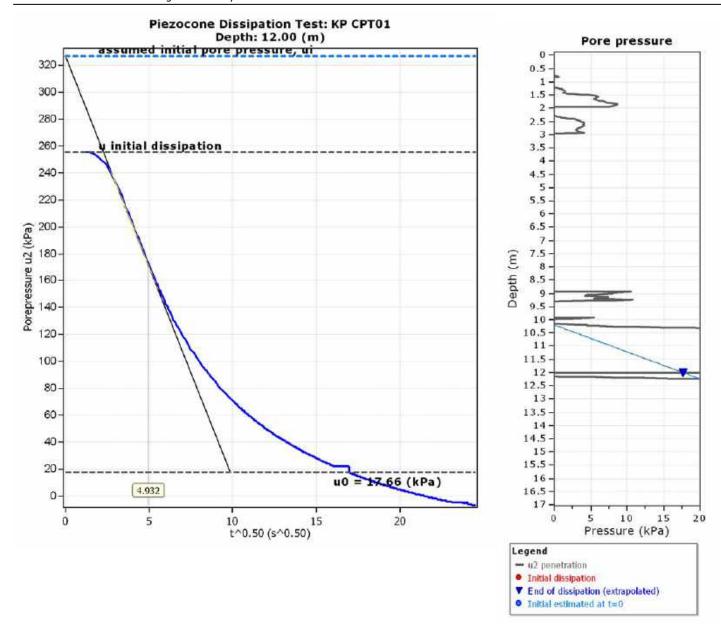
The dissipation of pore pressures during a CPTu dissipation test is controlled by the coefficient of consolidation in the horizontal direction (c_h) which is influenced by a combination of the soil permeability (k_h) and compressibility (M), as defined by the following:

$$k_h = c_h \times \gamma_w / M$$

where: M is the 1-D constrained modulus and γ_w is the unit weight of water, in compatible units.

Tabular results

	CPTU Borehole	Depth (m)	(t ₅₀) ^{0.50}	t ₅₀ (s)	t ₅₀ (years)	G/S _u	C _h (m²/s)	C _h (m²/year)	M (MPa)	k _h (m/s)
I	KP CPT01	12.00	4.9	24	7.71E-007	100.00	3.19E-005	1006	10.94	2.86E-008



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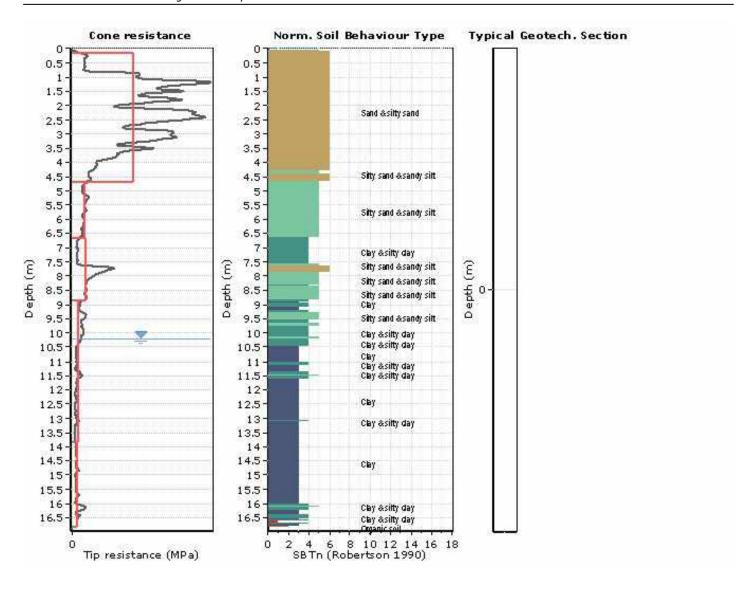
Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT01

Total depth: 16.88 m, Date: 3/09/2020 Coords: lat 305898.4° lon 6880747.5°

Cone Type: 180911



Tabular results

.:: Layer No: 1 ::.

Code: Layer_1 Start depth: 0.12 (m), End depth: 4.70 (m)

Description: Sand & silty sand

Basic results

Total cone resistance: $18.67 \pm 11.80 \text{ MPa}$ Sleeve friction: $215.29 \pm 155.62 \text{ kPa}$

Ic: 1.68 ±0.17 SBT_n: 6

SBTn description: Sand & silty sand

Estimation results

Permeability: 1.19E-04 ±1.04E-04 m/s Constrai

N₆₀: 38.75 ±22.04 blows Es: 101.67 ±51.87 MPa Dr (%): 79.89 ±21.42 φ (degrees): 43.03 ±4.57 °

Unit weight: $19.00 \pm 0.00 \text{ kN/m}^3$

Constrained Mod.: 203.33 ±103.75 MPa

Go: 127.42 ± 65.02 MPa Su: 0.00 ± 0.00 kPa Su ratio: 0.00 ± 0.00 O.C.R.: 0.00 ± 0.00 .:: Layer No: 2 ::.

Code: Layer_2 **Start depth:** 4.70 (m), **End depth:** 6.66 (m)

Description: Silty sand & sandy silt

Basic results

Estimation results

Total cone resistance: 3.56 ±0.60 MPa

Sleeve friction: 33.65 ±6.97 kPa

Ic: 2.31 ±0.10 SBT_n: 5

SBTn description: Silty sand & sandy silt

Permeability: 1.12E-06 ±9.28E-07 m/s

 N_{60} : 11.63 ±1.30 blows Es: 45.56 ±3.46 MPa Dr (%): 30.52 ±3.31 φ (degrees): 33.69 ±0.65 °

Unit weight: 19.00 ±0.00 kN/m³

Go: 57.03 ±4.38 MPa Su: $0.00 \pm 0.00 \text{ kPa}$

Constrained Mod.: 53.01 ±18.39 MPa

CPT name: KP CPT01

Su ratio: 0.00 ±0.00 O.C.R.: 0.00 ±0.00

.:: Layer No: 3 ::.

Code: Layer_3 Start depth: 6.66 (m), End depth: 8.84 (m)

Description: Silty sand & sandy silt

Basic results

Estimation results

Total cone resistance: 4.08 ±2.98 MPa Sleeve friction: 37.24 ±22.21 kPa

Ic: 2.46 ±0.30 SBT_n: 5

SBTn description: Silty sand & sandy silt

Permeability: 3.65E-06 ±1.07E-05 m/s N_{60} : 13.13 ±6.50 blows

Es: 65.55 ±11.13 MPa Dr (%): 33.50 ±8.88 φ (degrees): 34.33 ± 1.36 °

Unit weight: 19.00 ±0.00 kN/m³

Constrained Mod.: 59.48 ±51.08 MPa

Su: $0.00 \pm 0.00 \text{ kPa}$ Su ratio: 0.00 ±0.00 O.C.R.: 0.00 ±0.00

Go: 65.63 ±22.23 MPa

.:: Layer No: 4 ::.

Code: Layer_4 **Start depth:** 8.84 (m), **End depth:** 13.86 (m)

Description: Clay

Basic results

Estimation results

Total cone resistance: 1.81 ±0.91 MPa

Total cone resistance: 1.49 ±0.61 MPa

Sleeve friction: 25.45 ±11.46 kPa

Sleeve friction: 24.64 ±12.55 kPa Ic: 3.00 ± 0.30

SBT_n: 3 SBTn description: Clay Permeability: 4.36E-08 ±8.45E-08 m/s

 N_{60} : 8.72 ±2.60 blows Es: 0.00 ±0.00 MPa Dr (%): 0.00 ±0.00 φ (degrees): 0.00 ±0.00 °

Unit weight: 19.00 ±0.00 kN/m3

Constrained Mod.: 15.95 ±16.24 MPa

Go: 56.06 ±11.89 MPa Su: 100.60 ±54.26 kPa Su ratio: 0.52 ±0.32 O.C.R.: 2.38 ±1.48

.:: Layer No: 5 ::.

Code: Layer_5 **Start depth:** 13.86 (m), **End depth:** 16.82 (m)

Description: Clay

Basic results

Ic: 3.21 ±0.22

SBT_n: 3

Estimation results

Permeability: 7.26E-09 ±2.09E-08 m/s

 N_{60} : 8.47 ±2.01 blows Es: 0.00 ±0.00 MPa Dr (%): 0.00 ±0.00 ϕ (degrees): 0.00 ±0.00 ° Constrained Mod.: 7.39 ±9.73 MPa

Go: 58.81 ±9.91 MPa Su: 84.60 ±41.26 kPa Su ratio: 0.35 ± 0.17 O.C.R.: 1.62 ±0.77

SBTn description: Clay

Unit weight: $19.00 \pm 0.00 \text{ kN/m}^3$

CPeT-IT v.3.0.1.17 - CPTU data presentation & interpretation software - Report created on: 11/11/2020, 1:11:13 PM Project file: \pefs3\technical\PE801-00296 Thunderbox\Technical\Geotech\2020 TSF Investigation\CPT Data\CPT Data.cpt



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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT01

Total depth: 16.88 m, Date: 3/09/2020 Coords: lat 305898.4° lon 6880747.5°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

Summary table of mean values

From depth To depth (m)	Thickness (m)	Permeability (m/s)	SPT _{N60} (blows/30am)	E₅ (MPa)	D _r (%)	Friction angle	Constrained modulus, M (MPa)	Shear modulus, G _o (MPa)	Undrained strength, S _U (kPa)	Undrained strength ratio	OCR	Unit weight (kN/m³)
0.12	4.58	1.19E-04	38.8	101.7	79.9	43.0	203.3	127.4	0.0	0.0	0.0	19.0
4.70	1.50	(±1.04E-04)	(±22.0)	(±51.9)	(±21.4)	(±4.6)	(±103.7)	(±65.0)	(±0.0)	(±0.0)	(±0.0)	(±0.0)
4.70	1.96	1.12E-06	11.6	45.6	30.5	33.7	53.0	57.0	0.0	0.0	0.0	19.0
6.66	1.50	(±9.28E-07)	(±1.3)	(±3.5)	(±3.3)	(±0.7)	(±18.4)	(±4.4)	(±0.0)	(±0.0)	(±0.0)	(±0.0)
6.66	2.18	3.65E-06	13.1	65.6	33.5	34.3	59.5	65.6	0.0	0.0	0.0	19.0
8.84	2.10	(±1.07E-05)	(±6.5)	(±11.1)	(±8.9)	(±1.4)	(±51.1)	(±22.2)	(±0.0)	(±0.0)	(±0.0)	(±0.0)
8.84	5.02	4.36E-08	8.7	0.0	0.0	0.0	16.0	56.1	100.6	0.5	2.4	19.0
13.86	5.02	(±8.45E-08)	(±2.6)	(±0.0)	(±0.0)	(±0.0)	(±16.2)	(±11.9)	(±54.3)	(±0.3)	(±1.5)	(±0.0)
13.86	2.96	7.26E-09	8.5	0.0	0.0	0.0	7.4	58.8	84.6	0.4	1.6	19.0
16.82	2.90	(±2.09E-08)	(±2.0)	(±0.0)	(±0.0)	(±0.0)	(±9.7)	(±9.9)	(±41.3)	(±0.2)	(±0.8)	(±0.0)

Depth values presented in this table are measured from free ground surface

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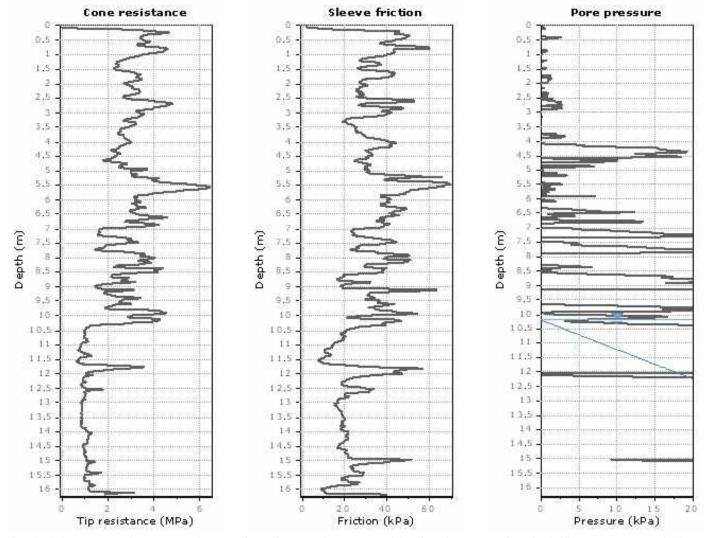
Total depth: 16.24 m, Date: 3/09/2020

Coords: lat 305871.8° lon 6880733.7° Cone Type: 180911

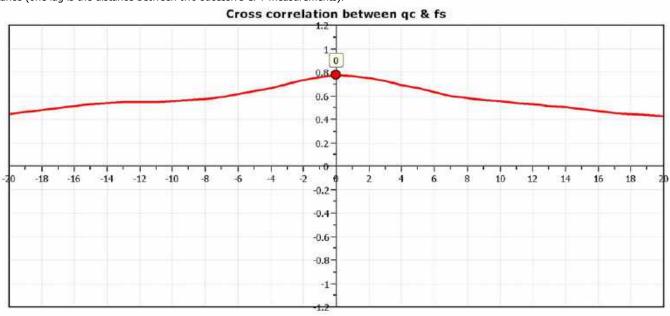
Cone Operator: Hagstrom Drilling

CPT: KP CPT 02





The plot below presents the cross correlation coeficient between the raw qc and fs values (as measured on the field). X axes presents the lag distance (one lag is the distance between two sucessive CPT measurements).



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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

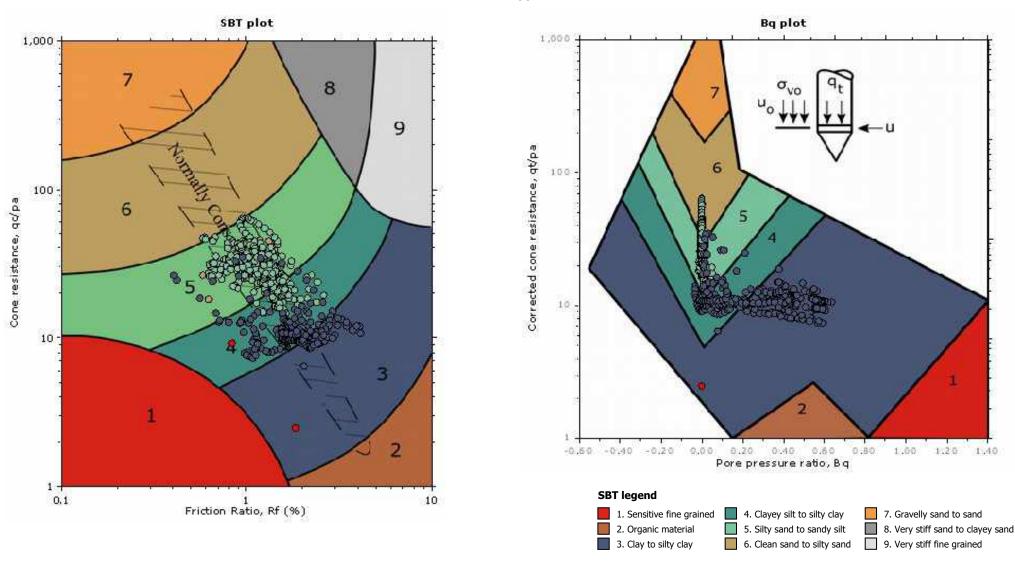
CPT: KP CPT 02

Total depth: 16.24 m, Date: 3/09/2020 Coords: lat 305871.8° lon 6880733.7°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

SBT - Bq plots



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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

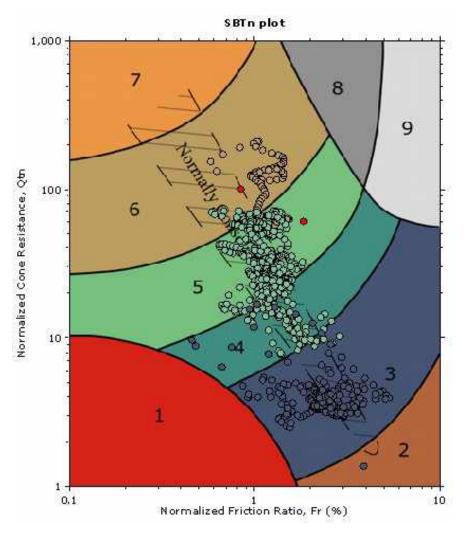
CPT: KP CPT 02

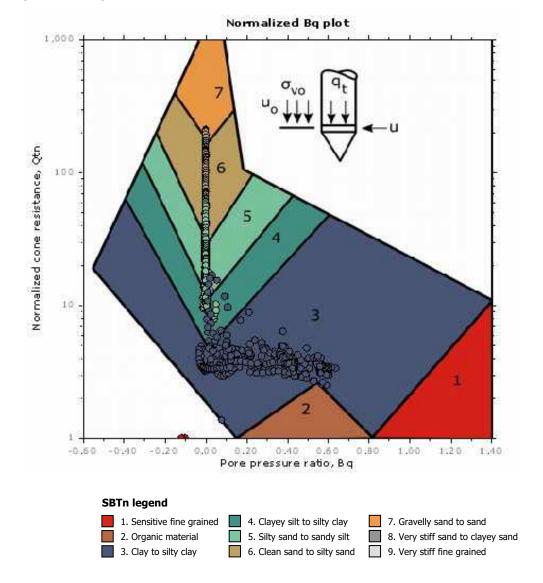
Total depth: 16.24 m, Date: 3/09/2020 Coords: lat 305871.8° lon 6880733.7°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

SBT - Bq plots (normalized)







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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

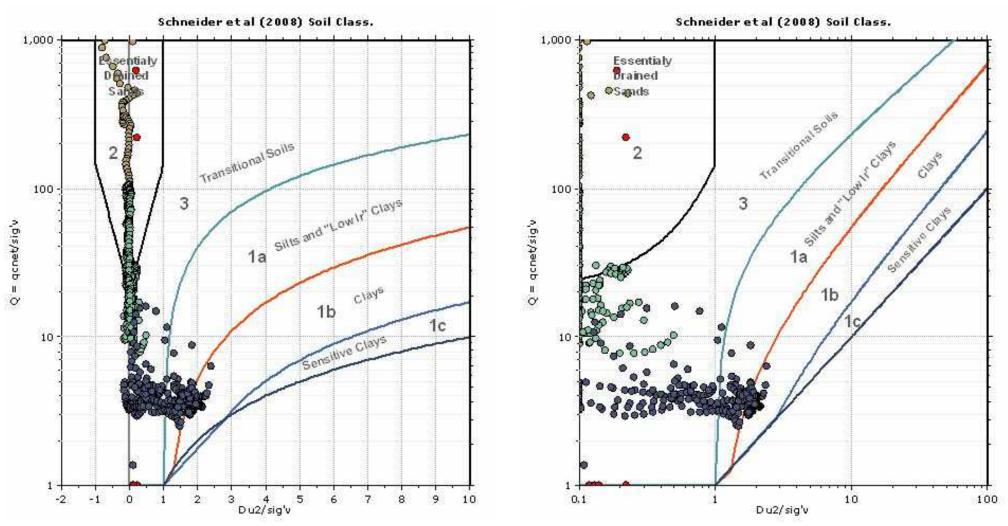
CPT: KP CPT 02

Total depth: 16.24 m, Date: 3/09/2020 Coords: lat 305871.8° lon 6880733.7°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

Bq plots (Schneider)







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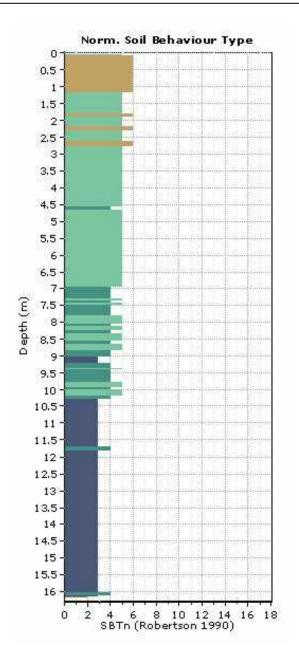
Thunderbox Gold Mine Project:

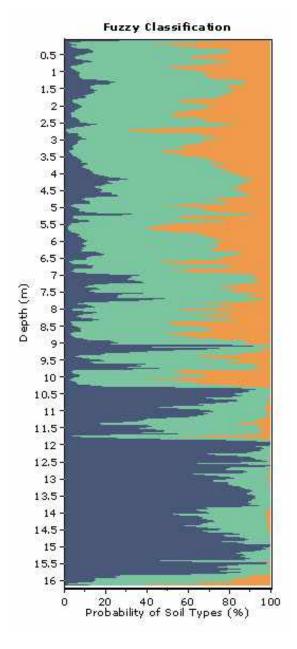
Location: Leinster, Western Australia, Australia

CPT: KP CPT 02

Total depth: 16.24 m, Date: 3/09/2020 Coords: lat 305871.8° lon 6880733.7°

Cone Type: 180911





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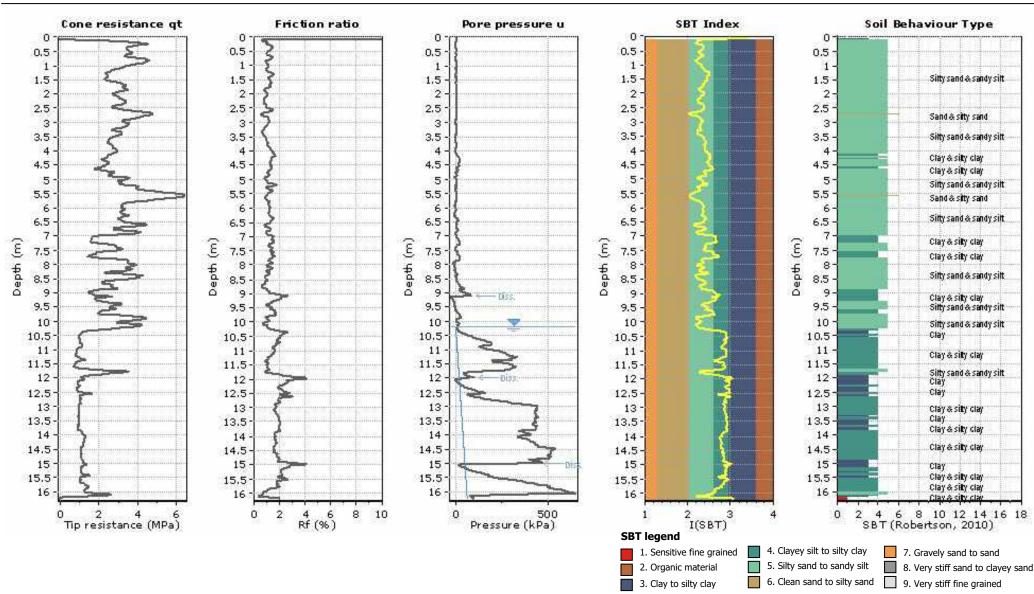
Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT 02

Total depth: 16.24 m, Date: 3/09/2020 Coords: lat 305871.8° lon 6880733.7°

Cone Type: 180911



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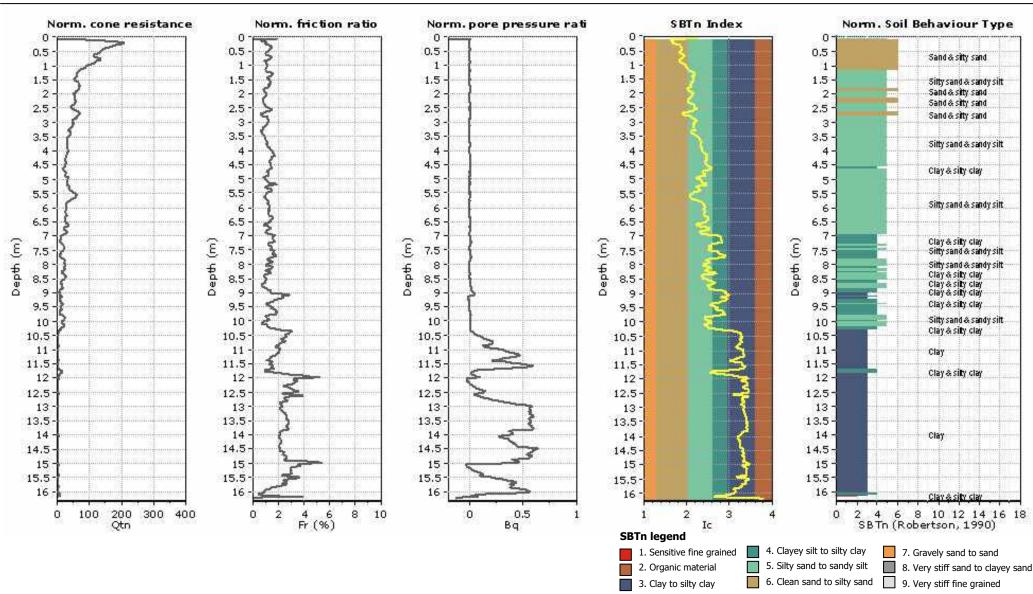
Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT 02

Total depth: 16.24 m, Date: 3/09/2020 Coords: lat 305871.8° lon 6880733.7°

Cone Type: 180911



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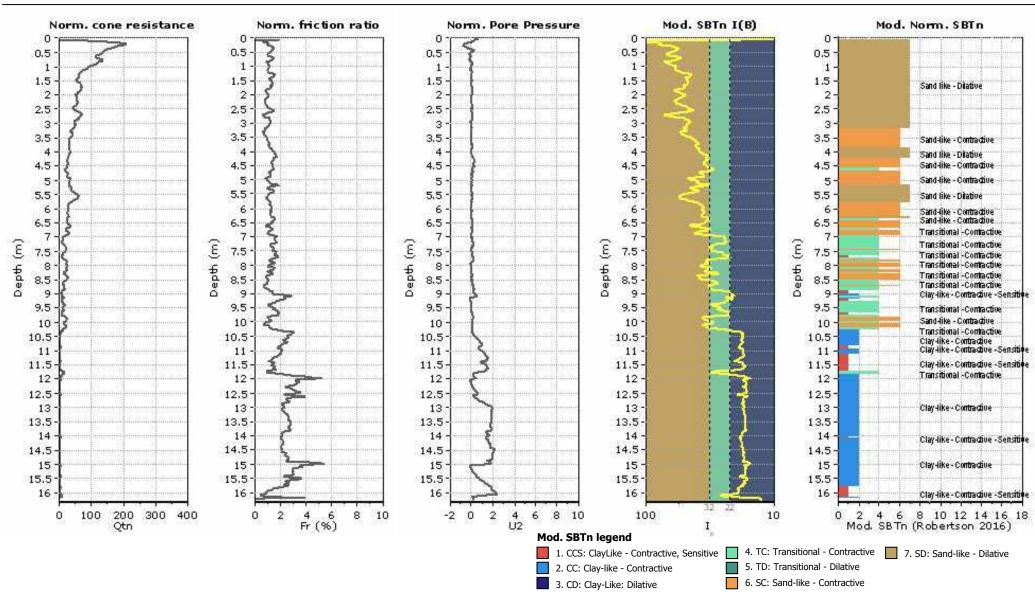
Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT 02

Total depth: 16.24 m, Date: 3/09/2020 Coords: lat 305871.8° lon 6880733.7°

Cone Type: 180911



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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

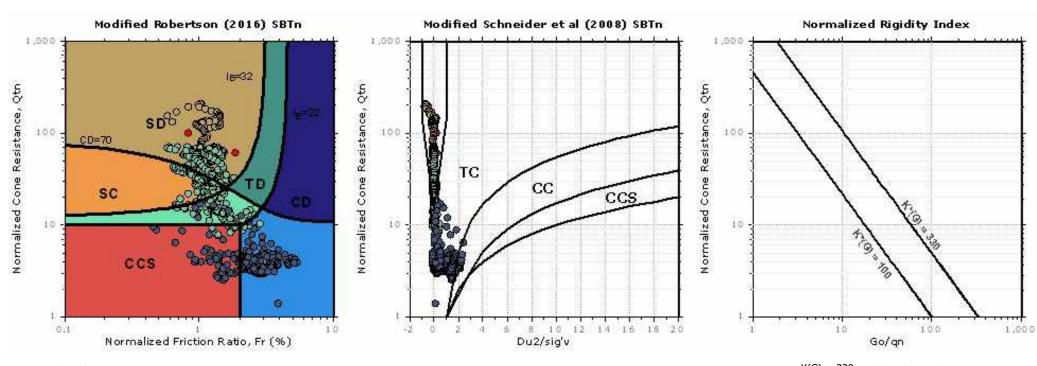
CPT: KP CPT 02

Total depth: 16.24 m, Date: 3/09/2020 Coords: lat 305871.8° lon 6880733.7°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

Updated SBTn plots



K(G) > 330: Soils with significant microstructure (e.g. age/cementation)

CCS: Clay-like - Contractive - Sensitive

CC: Clay-like - Contractive CD: Clay-like - Dilative

TC: Transitional - Contractive

TD: Transitional - Dilative

SC: Sand-like - Contractive

SD: Sand-like - Dilative

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Project: Thunderbox Gold Mine

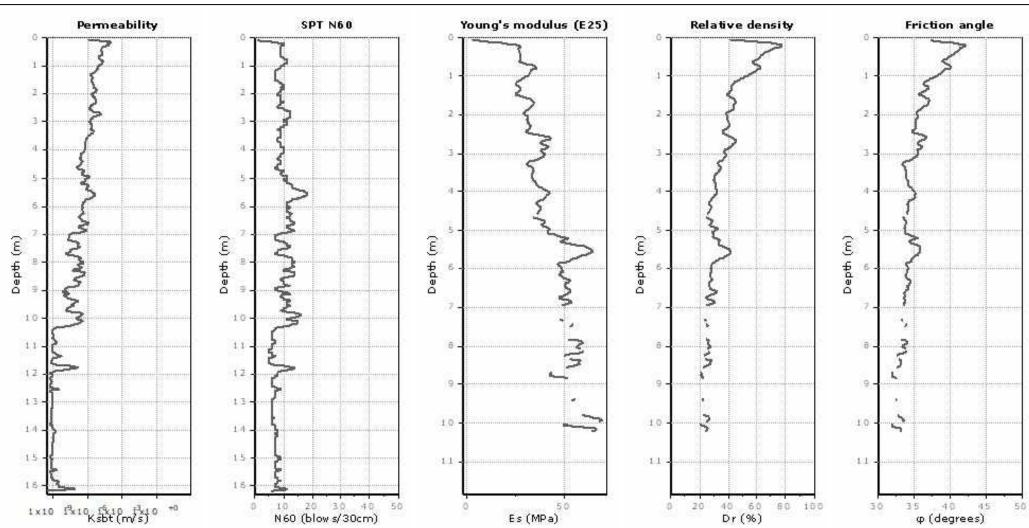
Location: Leinster, Western Australia, Australia

CPT: KP CPT 02

Total depth: 16.24 m, Date: 3/09/2020 Coords: lat 305871.8° lon 6880733.7°

Cone Type: 180911

Cone Operator: Hagstrom Drilling



Calculation parameters

Permeability: Based on SBT_n SPT N₆₀: Based on I_c and q_t

Young's modulus: Based on variable alpha using I_c (Robertson, 2009)

Relative density constant, C_{Dr}: 350.0 Phi: Based on Kulhawy & Mayne (1990) User defined estimation data

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Project: Thunderbox Gold Mine

Calculation parameters

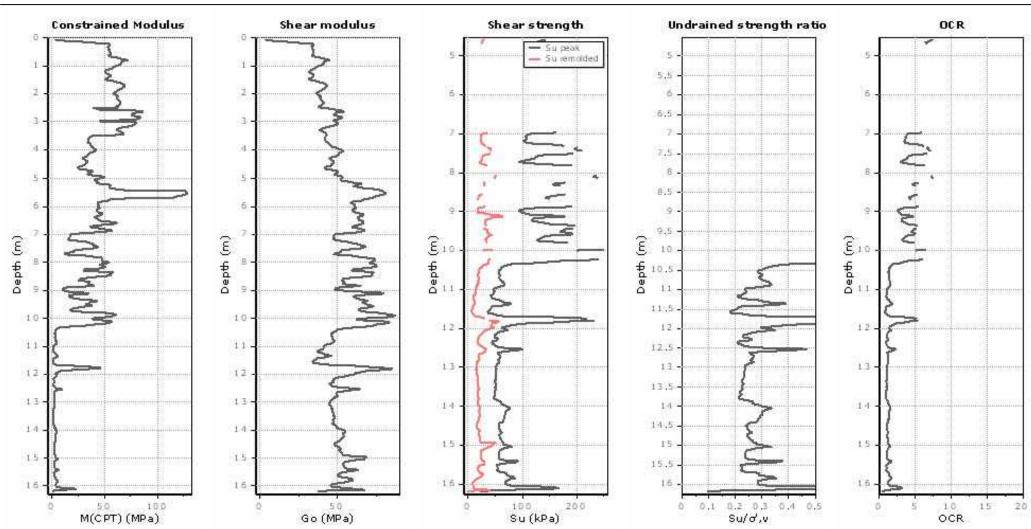
Location: Leinster, Western Australia, Australia

CPT: KP CPT 02

Total depth: 16.24 m, Date: 3/09/2020 Coords: lat 305871.8° lon 6880733.7°

Cone Type: 180911

Cone Operator: Hagstrom Drilling



OCR factor for clays, Nkt: 0.33

User defined estimation data

Flat Dilatometer Test data

Constrained modulus: Based on variable alpha using I_c and Q_{tn} (Robertson, 2009)

Go: Based on variable alpha using I_c (Robertson, 2009)

Undrained shear strength cone factor for clays, N_{kt}: 14

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Project: Thunderbox Gold Mine

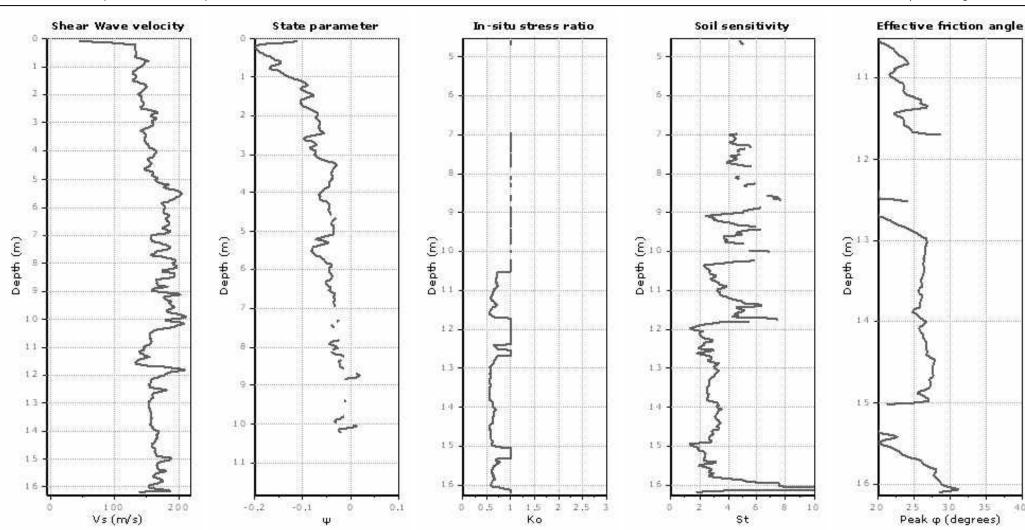
Location: Leinster, Western Australia, Australia

CPT: KP CPT 02

Total depth: 16.24 m, Date: 3/09/2020 Coords: lat 305871.8° lon 6880733.7°

Cone Type: 180911

Cone Operator: Hagstrom Drilling



Calculation parameters

Soil Sensitivity factor, N_S: 7.00

User defined estimation data



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Thunderbox Gold Mine Project:

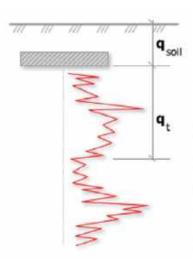
Location: Leinster, Western Australia, Australia

CPT: KP CPT 02

Total depth: 16.24 m, Date: 3/09/2020 Coords: lat 305871.8° lon 6880733.7°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

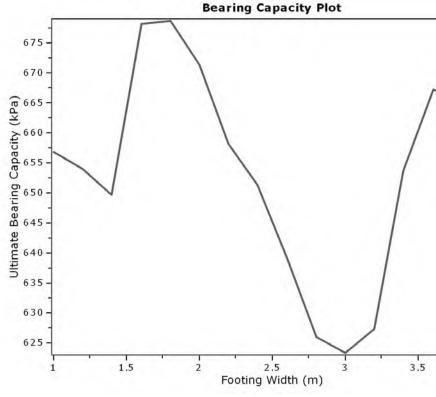


Bearing Capacity calculation is perfromed based on the formula:

$$Q_{ult} = R_k \times q_t + q_{soil}$$

where:

R_k: Bearing capacity factor qt: Average corrected cone resistance over calculation depth q_{soil}: Pressure applied by soil above footing



:: Tabula	ar results ::							
No	B (m)	Start Depth (m)	End Depth (m)	Ave. q _t (MPa)	R _k	Soil Press. (kPa)	Ult. bearing cap. (kPa)	
1	1.00	0.50	2.00	3.24	0.20	9.50	656.79	
2	1.20	0.50	2.30	3.22	0.20	9.50	653.96	
3	1.40	0.50	2.60	3.20	0.20	9.50	649.63	
4	1.60	0.50	2.90	3.34	0.20	9.50	678.14	
5	1.80	0.50	3.20	3.35	0.20	9.50	678.60	
6	2.00	0.50	3.50	3.31	0.20	9.50	671.20	
7	2.20	0.50	3.80	3.24	0.20	9.50	658.05	
8	2.40	0.50	4.10	3.21	0.20	9.50	651.32	
9	2.60	0.50	4.40	3.15	0.20	9.50	638.90	
10	2.80	0.50	4.70	3.08	0.20	9.50	625.88	
11	3.00	0.50	5.00	3.07	0.20	9.50	623.26	
12	3.20	0.50	5.30	3.09	0.20	9.50	627.28	
13	3.40	0.50	5.60	3.22	0.20	9.50	653.66	
14	3.60	0.50	5.90	3.29	0.20	9.50	667.05	
15	3.80	0.50	6.20	3.28	0.20	9.50	666.42	
16	4.00	0.50	6.50	3.28	0.20	9.50	666.05	

Total depth: 16.24 m, Date: 3/09/2020 Coords: lat 305871.8° lon 6880733.7°

Cone Type: 180911

CPT: KP CPT 02

Cone Operator: Hagstrom Drilling

Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

Dissipation Tests Results

Dissipation tests

Dissipation tests consists of stopping the piezocone penetration and observing porepressures (u) with elapsed time (t). The data are automatic recorded by the field computer and should take place until a minimum of 50% dissipation.

The porepressures are plotted as a function of square root of (t). The graphical technique suggested by Robertson and Campanella (1989), yields a value for t_{50} , which corresponds to the time for 50% consolidation.

The value of the coefficient of consolidation in the radial or horizontal direction c_h was then calculated by Houlsby and Teh's (1988) theory using the following equation:

$$c_h = \frac{T \times r^2 \times I_r^{0.5}}{t_{50}}$$

where:

T: time factor given by Houlsby and Teh's (1988) theory corresponding to the porepressure position

r: piezocone radius

I.: stiffness index, equal to shear modulus G divided by the undrained strength of clay (S_u).

t₅₀: time corresponding to 50% consolidation

Permeability estimates based on dissipation test

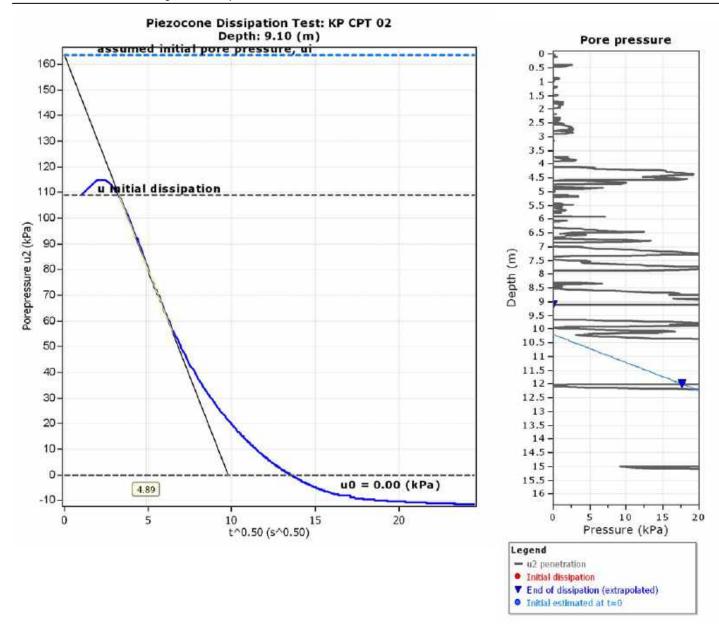
The dissipation of pore pressures during a CPTu dissipation test is controlled by the coefficient of consolidation in the horizontal direction (c_h) which is influenced by a combination of the soil permeability (k_h) and compressibility (M), as defined by the following:

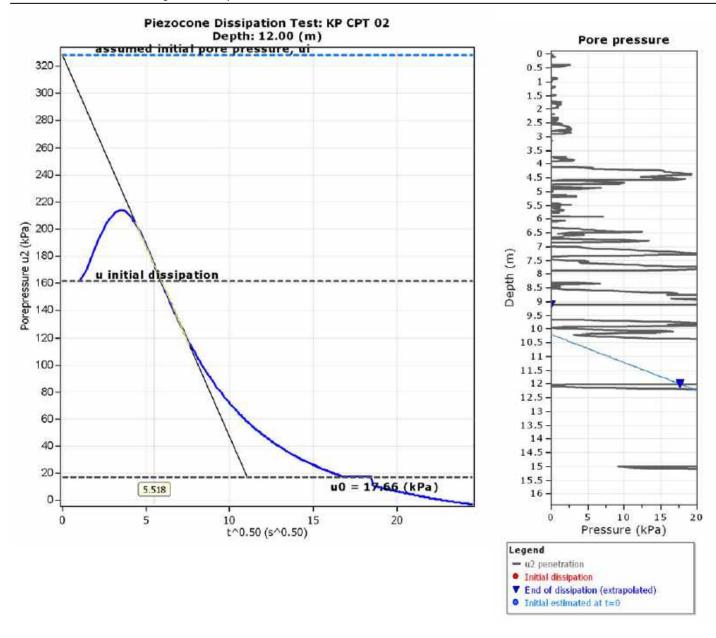
$$k_h = c_h \times \gamma_w / M$$

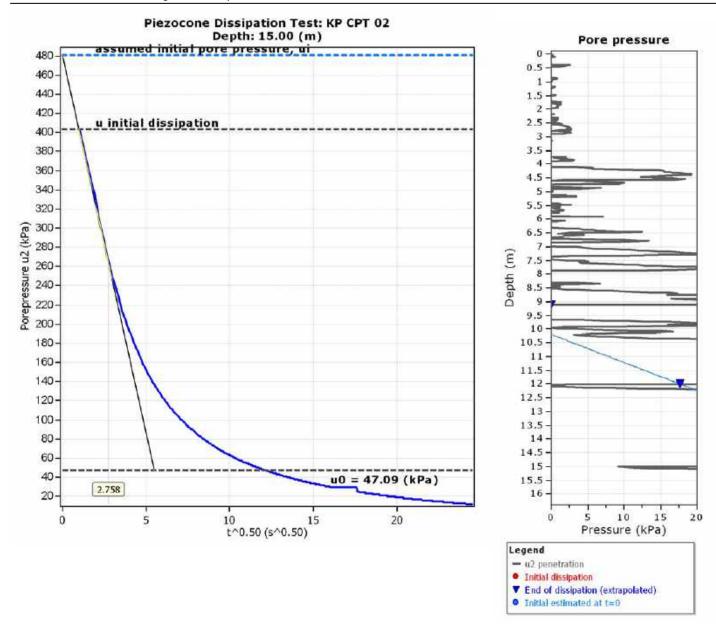
where: M is the 1-D constrained modulus and γ_w is the unit weight of water, in compatible units.

Tabular results

CPTU Borehole	Depth (m)	(t ₅₀) ^{0.50}	t ₅₀ (s)	t ₅₀ (years)	G/S _u	C _h (m²/s)	C _h (m²/year)	M (MPa)	k _h (m/s)
KP CPT 02	9.10	4.9	24	7.58E-007	100.00	3.25E-005	1024	30.96	1.03E-008
KP CPT 02	12.00	5.5	30	9.65E-007	100.00	2.55E-005	804	7.08	3.53E-008
KP CPT 02	15.00	2.8	8	2.41E-007	100.00	1.02E-004	3218	6.03	1.66E-007









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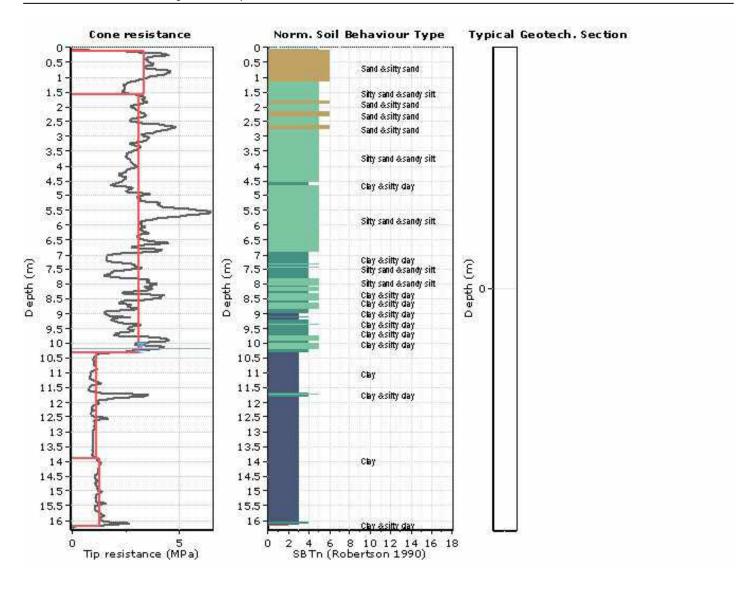
Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT 02

Total depth: 16.24 m, Date: 3/09/2020 Coords: lat 305871.8° lon 6880733.7°

Cone Type: 180911



Tabular results

.:: Layer No: 1 ::.

Code: Layer_1 Start depth: 0.10 (m), End depth: 1.56 (m)

Description: Sand & silty sand

Basic results

Total cone resistance: 3.35 ± 0.73 MPa Sleeve friction: 39.16 ± 9.99 kPa

Ic: 1.94 ±0.16 SBT_n: 6

SBTn description: Sand & silty sand

Estimation results

Permeability: 1.99E-05 ±2.34E-05 m/s Constrained Mod.: 55.21 ±7.98 MPa

Unit weight: $19.00 \pm 0.00 \text{ kN/m}^3$

CPT name: KP CPT 02

.:: Layer No: 2 ::.

Start depth: 1.56 (m), **End depth:** 10.30 (m) Code: Layer_2

Description: Silty sand & sandy silt

Basic results Estimation results

Total cone resistance: 3.09 ±0.89 MPa Permeability: 1.20E-06 ±1.92E-06 m/s Constrained Mod.: 46.69 ±21.24 MPa

Sleeve friction: 35.79 ±10.28 kPa Go: 56.40 ±12.10 MPa N_{60} : 10.72 ±2.33 blows Ic: 2.42 ±0.24 Es: 44.23 ±10.11 MPa Su: $0.00 \pm 0.00 \text{ kPa}$ SBT_n: 5 Su ratio: 0.00 ±0.00 Dr (%): 31.94 ±6.29

SBTn description: Silty sand & sandy silt O.C.R.: 0.00 ±0.00 φ (degrees): 34.37 ±1.09 °

Unit weight: 19.00 ±0.00 kN/m³

.:: Layer No: 3 ::.

Code: Layer_3 Start depth: 10.30 (m), End depth: 13.88 (m)

Description: Clay

Basic results Estimation results

Total cone resistance: 1.12 ±0.45 MPa Permeability: 4.04E-09 ±1.69E-08 m/s Constrained Mod.: 4.65 ±7.15 MPa

Sleeve friction: 20.94 ±9.31 kPa N_{60} : 6.69 ±1.62 blows Go: 48.59 ±9.15 MPa Ic: 3.29 ±0.16 Es: 0.00 ±0.00 MPa Su: 62.32 ±29.48 kPa Dr (%): 0.00 ±0.00 Su ratio: 0.30 ±0.15 SBT_n: 3 SBTn description: Clay φ (degrees): 0.00 ±0.00 ° O.C.R.: 1.37 ±0.68

Unit weight: 19.00 ±0.00 kN/m³

.:: Layer No: 4 ::.

Code 13.88 Start depth: 13.88 (m), End depth: 16.18 (m)

Description: Clay

Sleeve friction: 22.01 ±8.03 kPa

Basic results Estimation results

Permeability: $2.89E-09 \pm 1.02E-08 \text{ m/s}$ Total cone resistance: 1.24 ±0.27 MPa Constrained Mod.: 4.09 ±2.94 MPa

Go: 53.68 ±5.30 MPa

 N_{60} : 7.53 ±0.87 blows Ic: 3.28 ±0.15 Es: 0.00 ±0.00 MPa Su: 68.11 ±18.64 kPa SBT_n: 3 Dr (%): 0.00 ±0.00 Su ratio: 0.29 ±0.07

 ϕ (degrees): 0.00 ±0.00 ° O.C.R.: 1.32 ±0.34 SBTn description: Clay

Unit weight: 19.00 ±0.00 kN/m³



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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT 02

Total depth: 16.24 m, Date: 3/09/2020 Coords: lat 305871.8° lon 6880733.7°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

Summary table of mean values

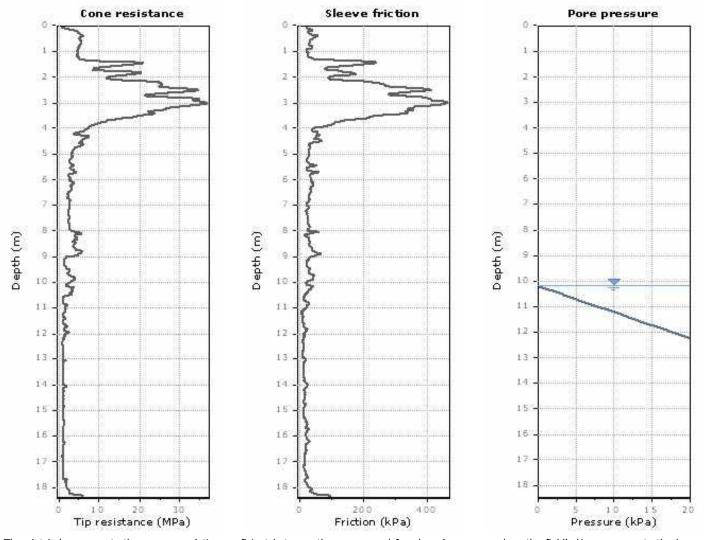
From depth To depth (m)	Thickness (m)	Permeability (m/s)	SPT _{N60} (blows/30am)	E₅ (MPa)	D _r (%)	Friction angle	Constrained modulus, M (MPa)	Shear modulus, G _o (MPa)	Undrained strength, S _U (kPa)	Undrained strength ratio	OCR	Unit weight (kN/m³)
0.10	1.46	1.99E-05	8.6	27.6	55.9	38.7	55.2	34.6	0.0	0.0	0.0	19.0
1.56	1.40	(±2.34E-05)	(±1.5)	(±4.0)	(±11.2)	(±1.9)	(±8.0)	(±5.0)	(±0.0)	(±0.0)	(±0.0)	(±0.0)
1.56	8.74	1.20E-06	10.7	44.2	31.9	34.4	46.7	56.4	0.0	0.0	0.0	19.0
10.30	0.7 1	(±1.92E-06)	(±2.3)	(±10.1)	(±6.3)	(±1.1)	(±21.2)	(±12.1)	(±0.0)	(±0.0)	(±0.0)	(±0.0)
10.30	3.58	4.04E-09	6.7	0.0	0.0	0.0	4.6	48.6	62.3	0.3	1.4	19.0
13.88	3.30	(±1.69E-08)	(±1.6)	(±0.0)	(±0.0)	(±0.0)	(±7.1)	(±9.2)	(±29.5)	(±0.1)	(±0.7)	(±0.0)
13.88	2.30	2.89E-09	7.5	0.0	0.0	0.0	4.1	53.7	68.1	0.3	1.3	19.0
16.18	2.50	(±1.02E-08)	(±0.9)	(±0.0)	(±0.0)	(±0.0)	(±2.9)	(±5.3)	(±18.6)	(±0.1)	(±0.3)	(±0.0)

Depth values presented in this table are measured from free ground surface

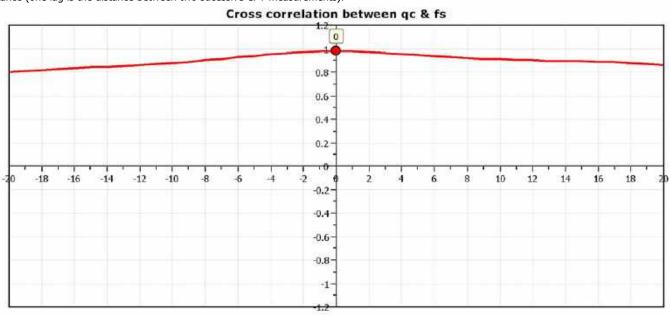
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Level 1, 184 Adelaide Terrace http://www.knightpiesold.com/

Project: Thunderbox Gold Mine Cone Type: 180911 Location: Leinster, Western Australia, Australia Cone Operator: Hagstrom Drilling



The plot below presents the cross correlation coeficient between the raw qc and fs values (as measured on the field). X axes presents the lag distance (one lag is the distance between two sucessive CPT measurements).



CPT: KP CPT03

Total depth: 18.40 m, Date: 3/09/2020

Coords: lat 305916.6° lon 6880462.7°

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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

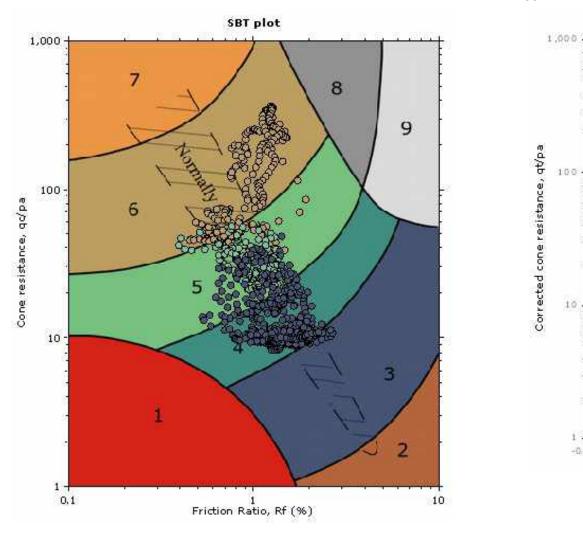
CPT: KP CPT03

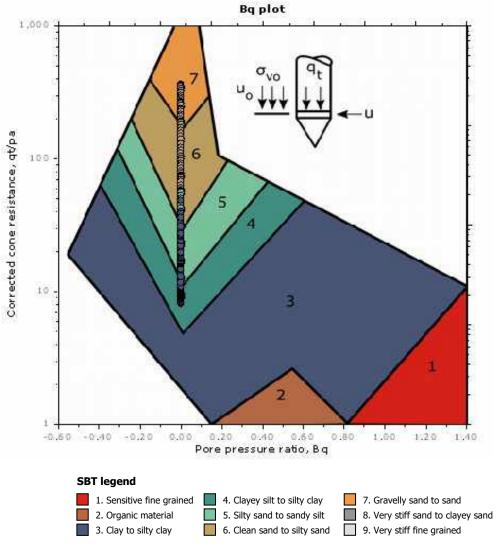
Total depth: 18.40 m, Date: 3/09/2020 Coords: lat 305916.6° lon 6880462.7°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

SBT - Bq plots





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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

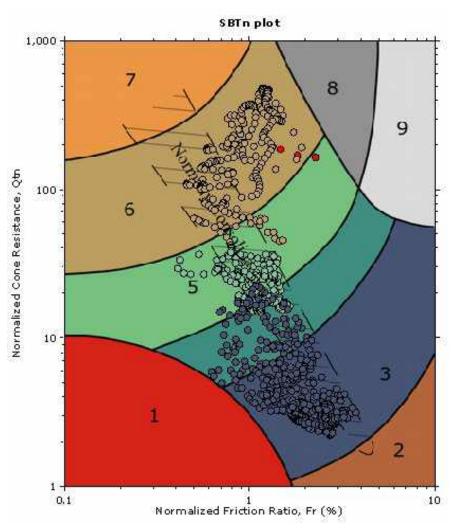
CPT: KP CPT03

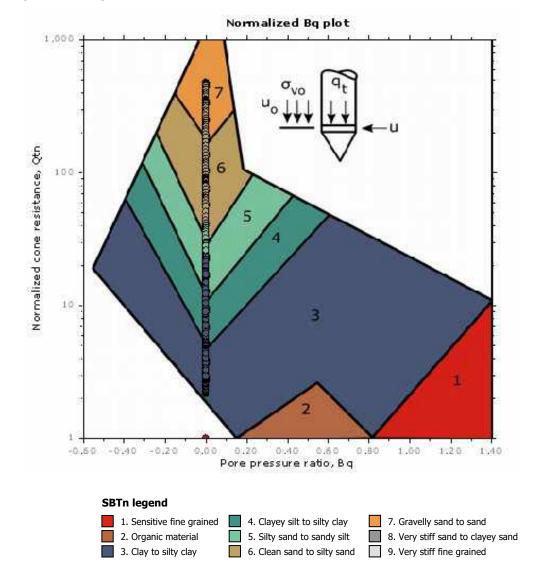
Total depth: 18.40 m, Date: 3/09/2020 Coords: lat 305916.6° lon 6880462.7°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

SBT - Bq plots (normalized)







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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

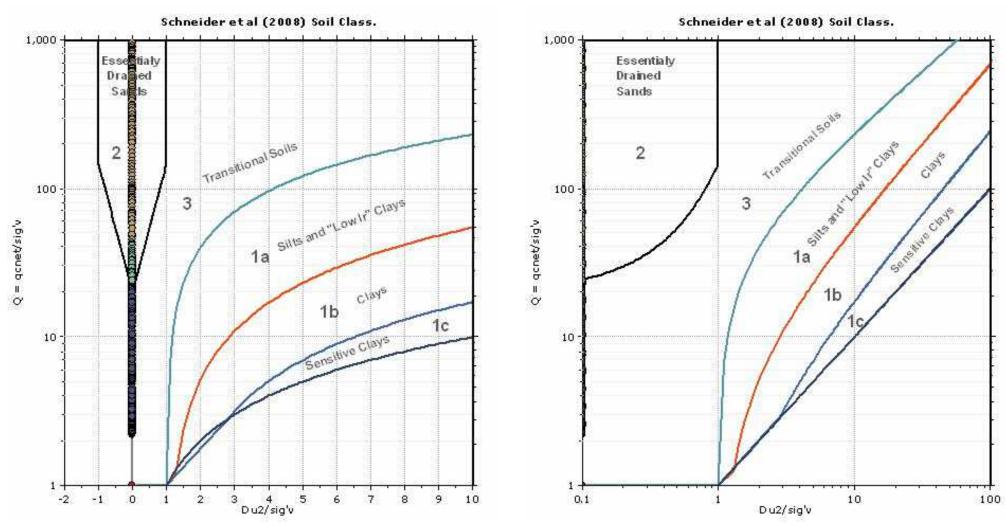
CPT: KP CPT03

Total depth: 18.40 m, Date: 3/09/2020 Coords: lat 305916.6° lon 6880462.7°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

Bq plots (Schneider)







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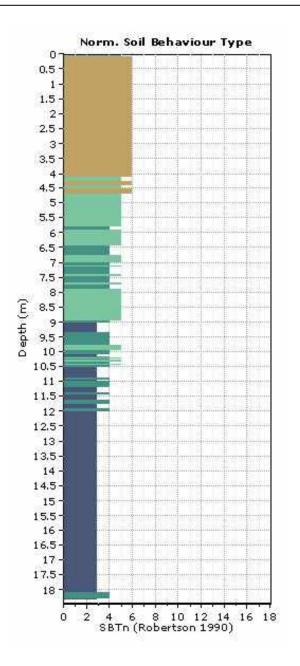
Thunderbox Gold Mine Project:

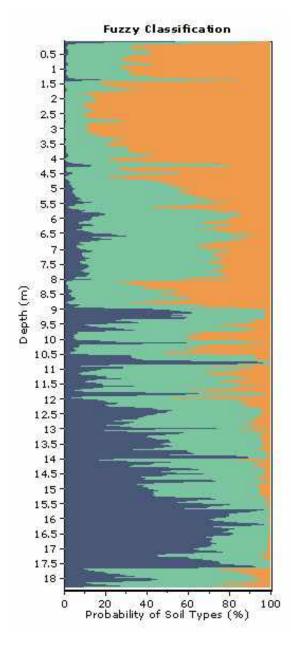
Location: Leinster, Western Australia, Australia

CPT: KP CPT03

Total depth: 18.40 m, Date: 3/09/2020 Coords: lat 305916.6° lon 6880462.7°

Cone Type: 180911





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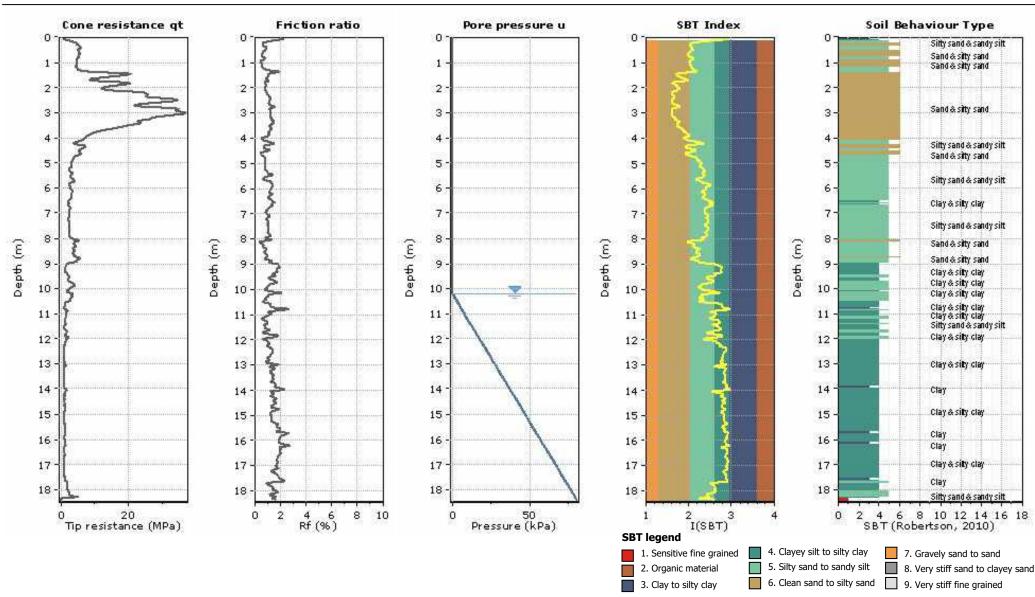
Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT03

Total depth: 18.40 m, Date: 3/09/2020 Coords: lat 305916.6° lon 6880462.7°

Cone Type: 180911



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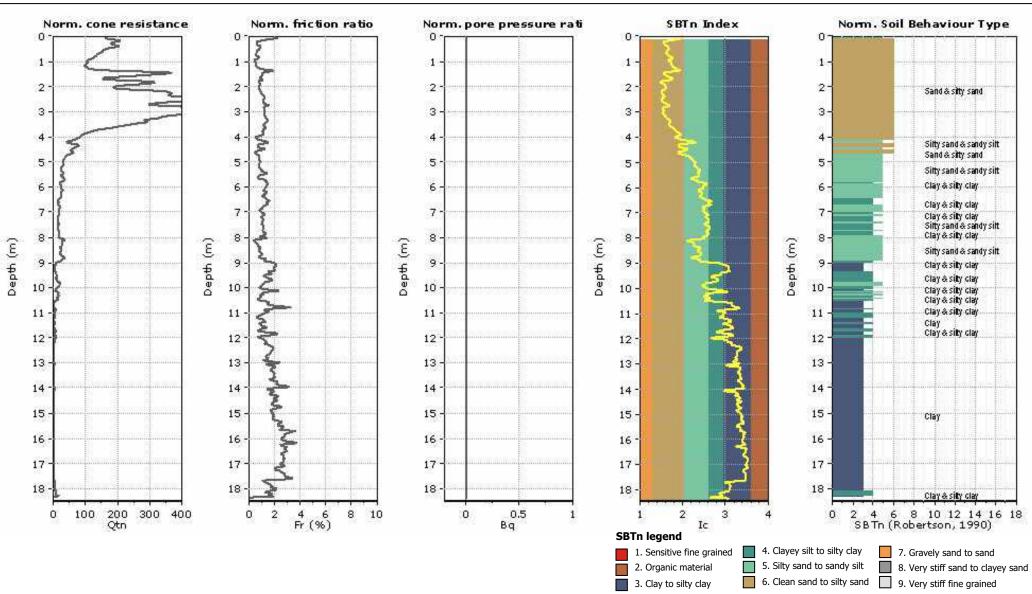
Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT03

Total depth: 18.40 m, Date: 3/09/2020 Coords: lat 305916.6° lon 6880462.7°

Cone Type: 180911



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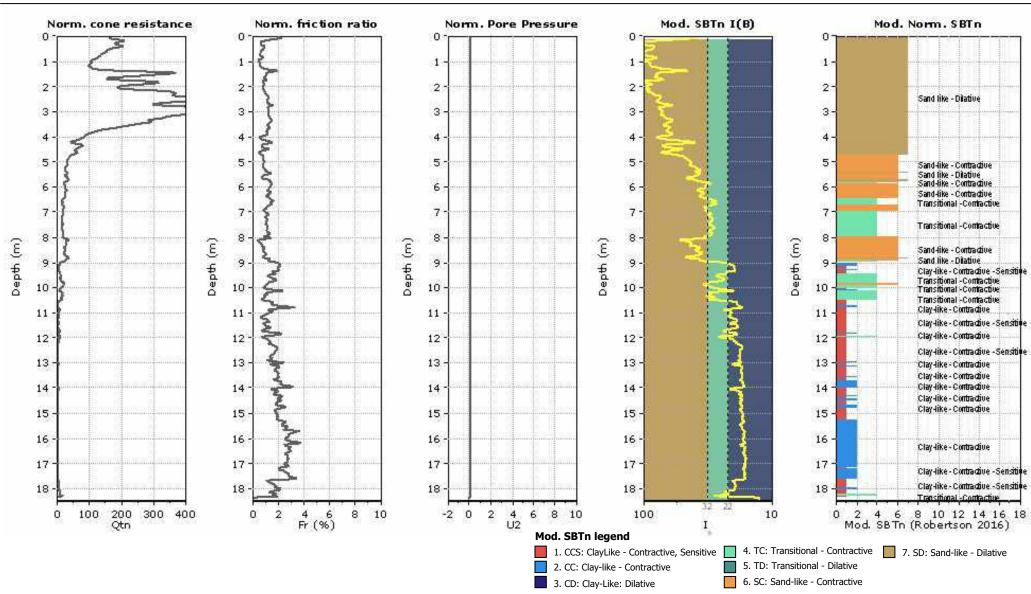
Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT03

Total depth: 18.40 m, Date: 3/09/2020 Coords: lat 305916.6° lon 6880462.7°

Cone Type: 180911



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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

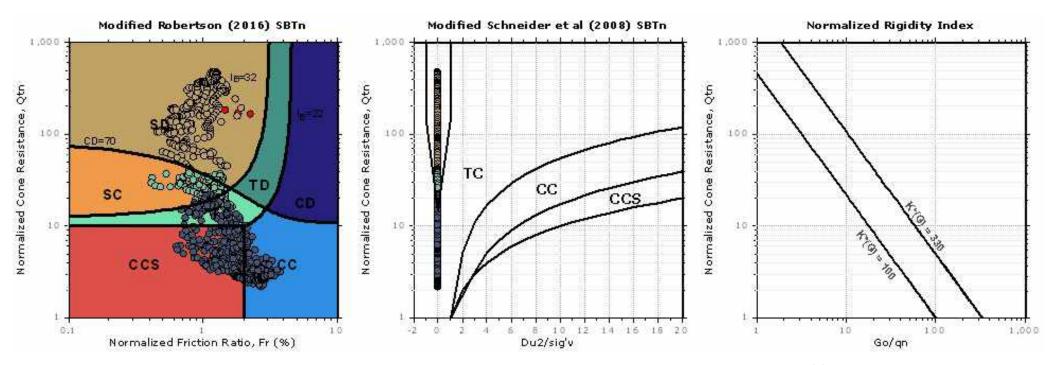
CPT: KP CPT03

Total depth: 18.40 m, Date: 3/09/2020 Coords: lat 305916.6° lon 6880462.7°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

Updated SBTn plots



CCS: Clay-like - Contractive - Sensitive

CC: Clay-like - Contractive
CD: Clay-like - Dilative
TC: Transitional - Contractive
TD: Transitional - Dilative
SC: Sand-like - Contractive
SD: Sand-like - Dilative

K(G) > 330: Soils with significant microstructure (e.g. age/cementation)

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CPT: KP CPT03

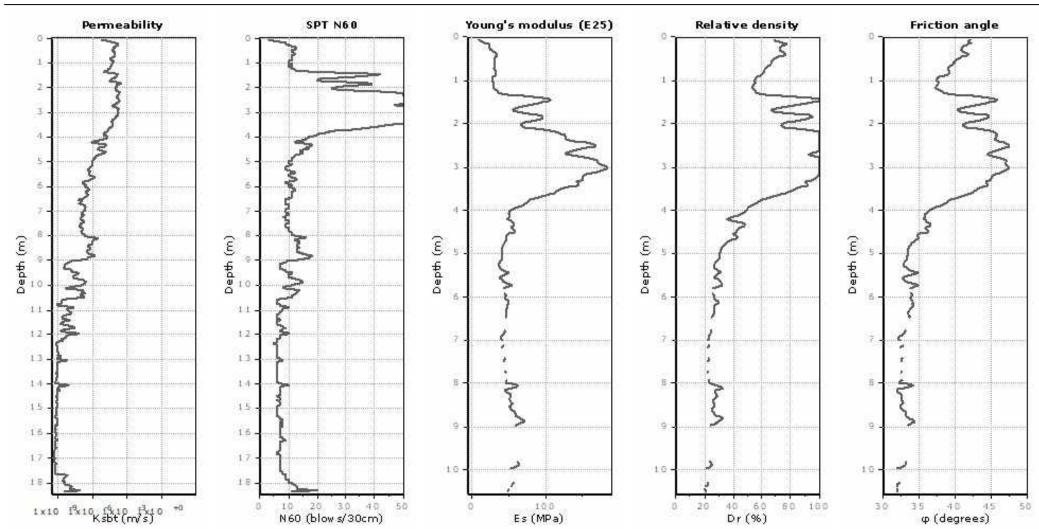
Total depth: 18.40 m, Date: 3/09/2020 Coords: lat 305916.6° lon 6880462.7°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia



Calculation parameters

Permeability: Based on SBT_n SPT N₆₀: Based on I_c and q_t

Young's modulus: Based on variable alpha using I_c (Robertson, 2009)

Relative density constant, C_{Dr}: 350.0 Phi: Based on Kulhawy & Mayne (1990) User defined estimation data

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Project: Thunderbox Gold Mine

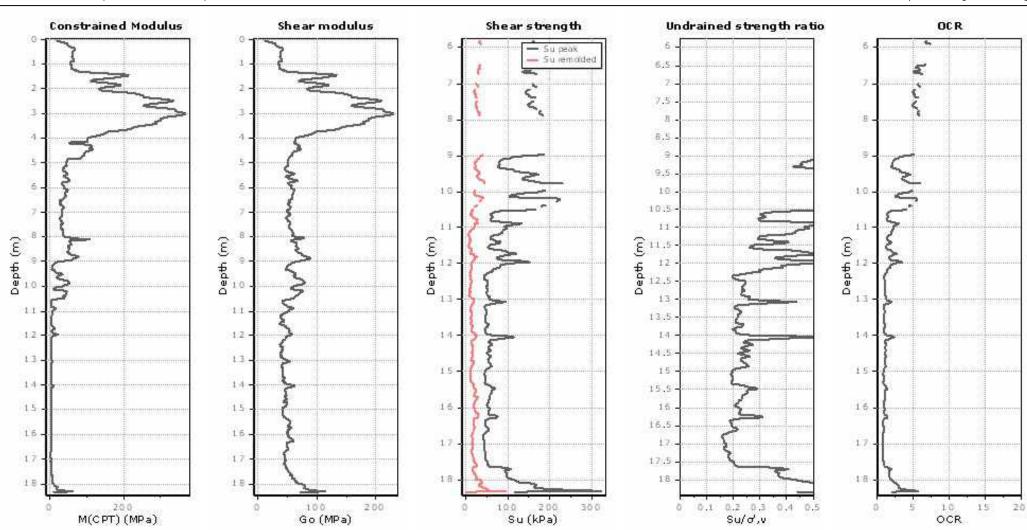
Location: Leinster, Western Australia, Australia

CPT: KP CPT03

Total depth: 18.40 m, Date: 3/09/2020 Coords: lat 305916.6° lon 6880462.7°

Cone Type: 180911

Cone Operator: Hagstrom Drilling



Calculation parameters

Constrained modulus: Based on variable alpha using I_c and Q_{tn} (Robertson, 2009) Go: Based on variable alpha using I_c (Robertson, 2009) Undrained shear strength cone factor for clays, N_{tt} : 14

OCR factor for clays, N_{kt} : 0.33

User defined estimation data

— Flat Dilatometer Test data

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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

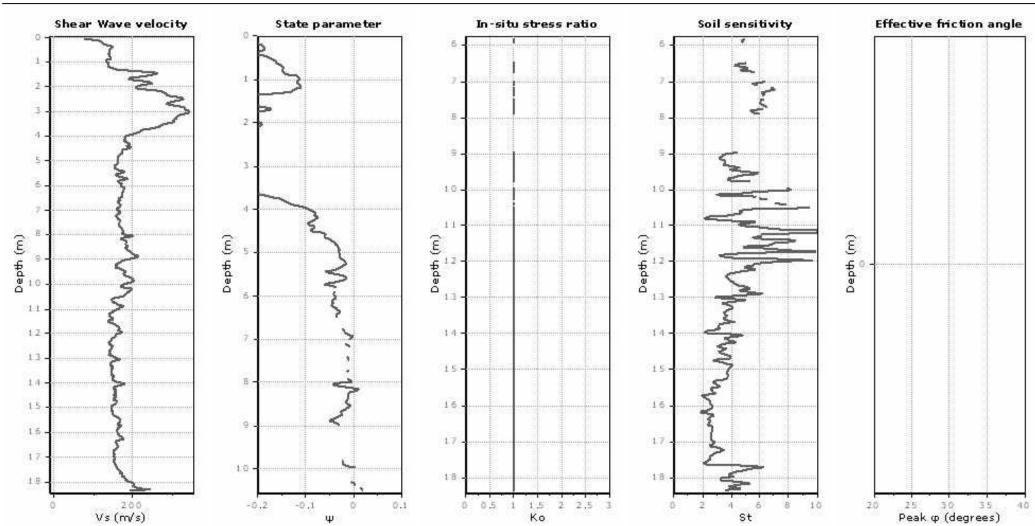
CPT: KP CPT03

Total depth: 18.40 m, Date: 3/09/2020

Coords: lat 305916.6° lon 6880462.7°

Cone Type: 180911

Cone Operator: Hagstrom Drilling



Calculation parameters

Soil Sensitivity factor, N_S: 7.00

User defined estimation data

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Thunderbox Gold Mine Project:

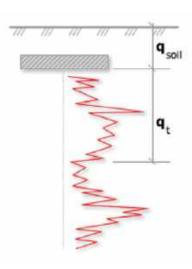
Location: Leinster, Western Australia, Australia

Total depth: 18.40 m, Date: 3/09/2020 Coords: lat 305916.6° lon 6880462.7°

Cone Type: 180911

CPT: KP CPT03

Cone Operator: Hagstrom Drilling

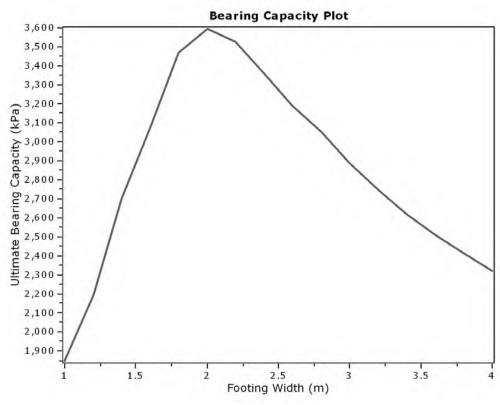


Bearing Capacity calculation is perfromed based on the formula:

$$Q_{ult} = R_k \times q_t + q_{soil}$$

where:

R_k: Bearing capacity factor qt: Average corrected cone resistance over calculation depth $q_{\text{soil}} \text{:} \text{ Pressure applied by soil}$ above footing



:: Tabula	ar results :	:					
No	B (m)	Start Depth (m)	End Depth (m)	Ave. q _t (MPa)	R _k	Soil Press. (kPa)	Ult. bearing cap. (kPa)
1	1.00	0.50	2.00	9.20	0.20	9.50	1849.53
2	1.20	0.50	2.30	10.94	0.20	9.50	2197.03
3	1.40	0.50	2.60	13.45	0.20	9.50	2700.30
4	1.60	0.50	2.90	15.34	0.20	9.50	3076.58
5	1.80	0.50	3.20	17.30	0.20	9.50	3470.19
6	2.00	0.50	3.50	17.92	0.20	9.50	3593.73
7	2.20	0.50	3.80	17.59	0.20	9.50	3527.00
8	2.40	0.50	4.10	16.75	0.20	9.50	3359.22
9	2.60	0.50	4.40	15.90	0.20	9.50	3189.42
10	2.80	0.50	4.70	15.22	0.20	9.50	3052.56
11	3.00	0.50	5.00	14.40	0.20	9.50	2889.34
12	3.20	0.50	5.30	13.69	0.20	9.50	2747.15
13	3.40	0.50	5.60	13.07	0.20	9.50	2623.06
14	3.60	0.50	5.90	12.52	0.20	9.50	2513.83
15	3.80	0.50	6.20	12.03	0.20	9.50	2415.16
16	4.00	0.50	6.50	11.58	0.20	9.50	2324.94

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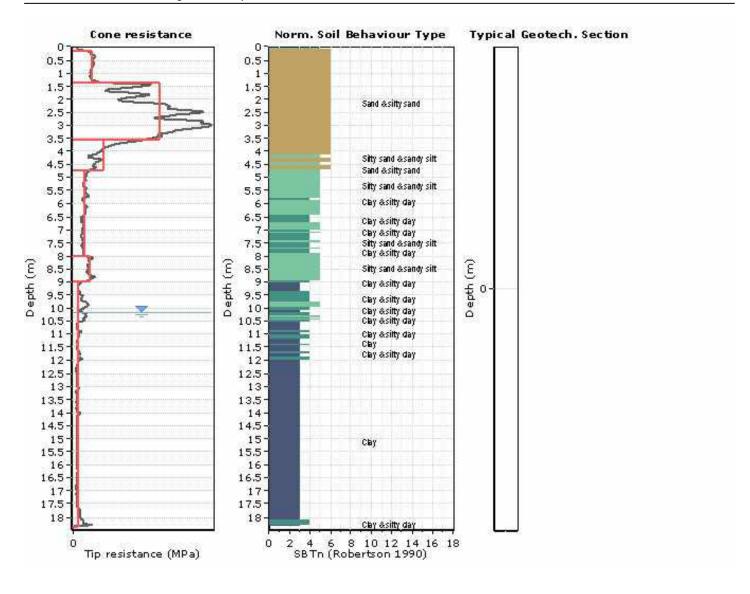
Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT03

Total depth: 18.40 m, Date: 3/09/2020 Coords: lat 305916.6° lon 6880462.7°

Cone Type: 180911



Tabular results

.:: Layer No: 1 ::.

Code: Layer_1 Start depth: 0.14 (m), End depth: 1.36 (m)

Description: Sand & silty sand

Basic results

Total cone resistance: 5.06 ± 0.73 MPa Sleeve friction: 37.94 ± 22.99 kPa

Ic: 1.70 ±0.09 SBT_n: 6

SBTn description: Sand & silty sand

Estimation results

Permeability: 7.14E-05 ±3.95E-05 m/s Constrained Mod.: 63.13 ±14.99 MPa

 N_{60} : 11.21 ±2.07 blows Go: 39.56 ±9.40 MPa Es: 31.57 ±7.50 MPa Su: 0.00 ±0.00 kPa Dr (%): 63.95 ±7.83 Su ratio: 0.00 ±0.00 ϕ (degrees): 39.47 ±1.57 ° O.C.R.: 0.00 ±0.00

Unit weight: $19.00 \pm 0.00 \text{ kN/m}^3$

.:: Layer No: 2 ::.

Code: Layer_2 **Start depth:** 1.36 (m), **End depth:** 3.54 (m)

Description: Sand & silty sand

Basic results

Estimation results

Total cone resistance: 22.99 ±7.89 MPa

Sleeve friction: 268.59 ±114.24 kPa

Ic: 1.61 ±0.08 SBT_n: 6

SBTn description: Sand & silty sand

Permeability: 1.35E-04 ±6.14E-05 m/s

 N_{60} : 47.55 ±15.26 blows Es: 123.84 ±37.74 MPa Dr (%): 92.09 ±9.93 φ (degrees): 44.74 ±1.95 °

Unit weight: 19.00 ±0.00 kN/m³

Go: 155.22 ±47.30 MPa Su: $0.00 \pm 0.00 \text{ kPa}$

Constrained Mod.: 247.69 ±75.48 MPa

CPT name: KP CPT03

Su ratio: 0.00 ±0.00

O.C.R.: 0.00 ±0.00

.:: Layer No: 3 ::.

Code: Layer_3 Start depth: 3.54 (m), End depth: 4.74 (m)

Description: Sand & silty sand

Basic results

Estimation results

Total cone resistance: 7.92 ±3.75 MPa Sleeve friction: 78.97 ±55.10 kPa

Ic: 1.96 ±0.14 SBT_n: 6

SBTn description: Sand & silty sand

Permeability: 1.45E-05 ±1.22E-05 m/s

 N_{60} : 20.20 ±7.87 blows Es: 63.97 ±20.11 MPa Dr (%): 49.94 ±12.06 φ (degrees): 37.18 ±2.37 °

Unit weight: 19.00 ±0.00 kN/m³

Constrained Mod.: 124.35 ±44.34 MPa

Su: $0.00 \pm 0.00 \text{ kPa}$ Su ratio: 0.00 ±0.00 O.C.R.: 0.00 ±0.00

Go: 80.17 ±25.20 MPa

.:: Layer No: 4 ::.

Start depth: 4.74 (m), **End depth:** 8.00 (m) Code: Layer_4

Description: Silty sand & sandy silt

Basic results

Estimation results

Total cone resistance: 2.83 ±0.49 MPa

Sleeve friction: 30.10 ±6.92 kPa Ic: 2.47 ±0.13

SBT_n: 5 SBTn description: Silty sand & sandy silt Permeability: 4.54E-07 ±4.98E-07 m/s

 N_{60} : 10.20 ±1.07 blows Es: 43.85 ±3.95 MPa Dr (%): 27.59 ±3.27 φ (degrees): 33.37 ±0.67 °

Unit weight: 19.00 ±0.00 kN/m3

Constrained Mod.: 38.89 ±10.55 MPa

Go: 54.55 ±4.41 MPa Su: 0.00 ±0.00 kPa Su ratio: 0.00 ±0.00 O.C.R.: 0.00 ±0.00

.:: Layer No: 5 ::.

Code: Layer_5 **Start depth:** 8.00 (m), **End depth:** 8.96 (m)

Description: Silty sand & sandy silt

Basic results

SBT_n: 5

Estimation results

Total cone resistance: 4.22 ±0.61 MPa

Sleeve friction: 36.18 ±12.36 kPa Ic: 2.34 ±0.10

SBTn description: Silty sand & sandy silt

Permeability: 8.51E-07 ±6.69E-07 m/s

 N_{60} : 14.24 ±1.65 blows Es: 56.46 ±6.68 MPa

Dr (%): 27.87 ±2.17 φ (degrees): 32.97 ±0.67 ° Unit weight: 19.00 ±0.00 kN/m³ Constrained Mod.: 58.89 ±13.25 MPa

Go: 70.76 ±8.37 MPa Su: 0.00 ±0.00 kPa Su ratio: 0.00 ±0.00 O.C.R.: 0.00 ±0.00

CPT name: KP CPT03

.:: Layer No: 6 ::.

Start depth: 8.96 (m), End depth: 18.36 (m)Code: Layer_6

Description: Clay

Basic results

Total cone resistance: 1.41 ±0.70 MPa Sleeve friction: 19.58 ±9.26 kPa

Ic: 3.17 ±0.30 SBT_n: 3

SBTn description: Clay

Estimation results

Permeability: 1.52E-08 ±4.13E-08 m/s

 N_{60} : 7.75 ±2.19 blows Es: 0.00 ±0.00 MPa Dr (%): 0.00 ±0.00 ϕ (degrees): 0.00 ±0.00 ° Constrained Mod.: 8.33 ±11.65 MPa

Go: 51.76 ±11.13 MPa Su: 76.39 ±41.72 kPa Su ratio: 0.35 ±0.22 O.C.R.: 1.63 ±1.02

Unit weight: $19.00 \pm 0.00 \text{ kN/m}^3$



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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT03

Total depth: 18.40 m, Date: 3/09/2020 Coords: lat 305916.6° lon 6880462.7°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

Summary table of mean values

From depth To depth (m)	Thickness (m)	Permeability (m/s)	SPT _{N60} (blows/30cm)	E _s (MPa)	D _r (%)	Friction angle	Constrained modulus, M (MPa)	Shear modulus, G _o (MPa)	Undrained strength, S _U (kPa)	Undrained strength ratio	OCR	Unit weight (kN/m³)
0.14	1.22	7.14E-05	11.2	31.6	63.9	39.5	63.1	39.6	0.0	0.0	0.0	19.0
1.36	1.22	(±3.95E-05)	(±2.1)	(±7.5)	(±7.8)	(±1.6)	(±15.0)	(±9.4)	(±0.0)	(±0.0)	(±0.0)	(±0.0)
1.36	2.18	1.35E-04	47.5	123.8	92.1	44.7	247.7	155.2	0.0	0.0	0.0	19.0
3.54	2.10	(±6.14E-05)	(±15.3)	(±37.7)	(±9.9)	(±1.9)	(±75.5)	(±47.3)	(±0.0)	(±0.0)	(±0.0)	(±0.0)
3.54	1.20	1.45E-05	20.2	64.0	49.9	37.2	124.3	80.2	0.0	0.0	0.0	19.0
4.74		(±1.22E-05)	(±7.9)	(±20.1)	(±12.1)	(±2.4)	(±44.3)	(±25.2)	(±0.0)	(±0.0)	(±0.0)	(±0.0)
4.74	3.26	4.54E-07	10.2	43.8	27.6	33.4	38.9	54.6	0.0	0.0	0.0	19.0
8.00	3.20	(±4.98E-07)	(±1.1)	(±3.9)	(±3.3)	(±0.7)	(±10.6)	(±4.4)	(±0.0)	(±0.0)	(±0.0)	(±0.0)
8.00	0.96	8.51E-07	14.2	56.5	27.9	33.0	58.9	70.8	0.0	0.0	0.0	19.0
8.96	0.90	(±6.69E-07)	(±1.7)	(±6.7)	(±2.2)	(±0.7)	(±13.3)	(±8.4)	(±0.0)	(±0.0)	(±0.0)	(±0.0)
8.96	9.40	1.52E-08	7.8	0.0	0.0	0.0	8.3	51.8	76.4	0.4	1.6	19.0
18.36	5. 4 0	(±4.13E-08)	(±2.2)	(±0.0)	(±0.0)	(±0.0)	(±11.6)	(±11.1)	(±41.7)	(±0.2)	(±1.0)	(±0.0)

Depth values presented in this table are measured from free ground surface

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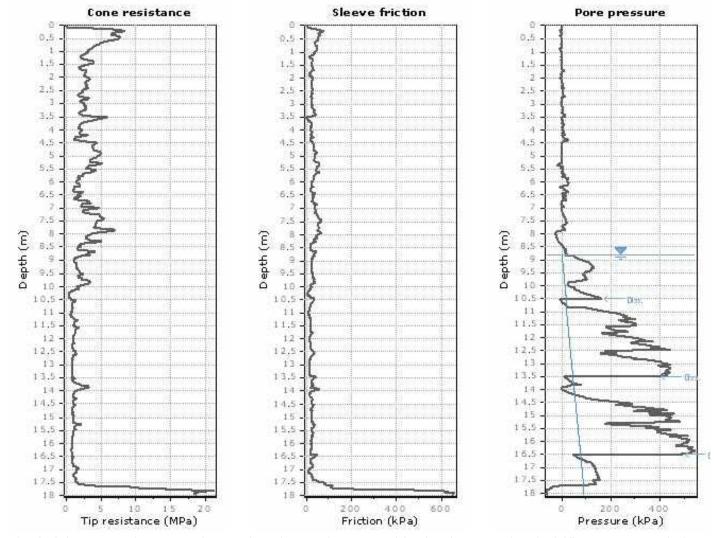
Level 1, 184 Adelaide Terrace http://www.knightpiesold.com/

CPT: KP CPT04

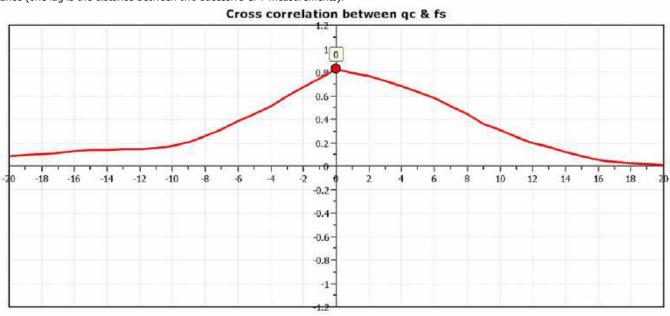
Total depth: 18.04 m, Date: 3/09/2020 Coords: lat 305888.1° lon 6880472.1°

Cone Type: 180911





The plot below presents the cross correlation coeficient between the raw qc and fs values (as measured on the field). X axes presents the lag distance (one lag is the distance between two sucessive CPT measurements).



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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

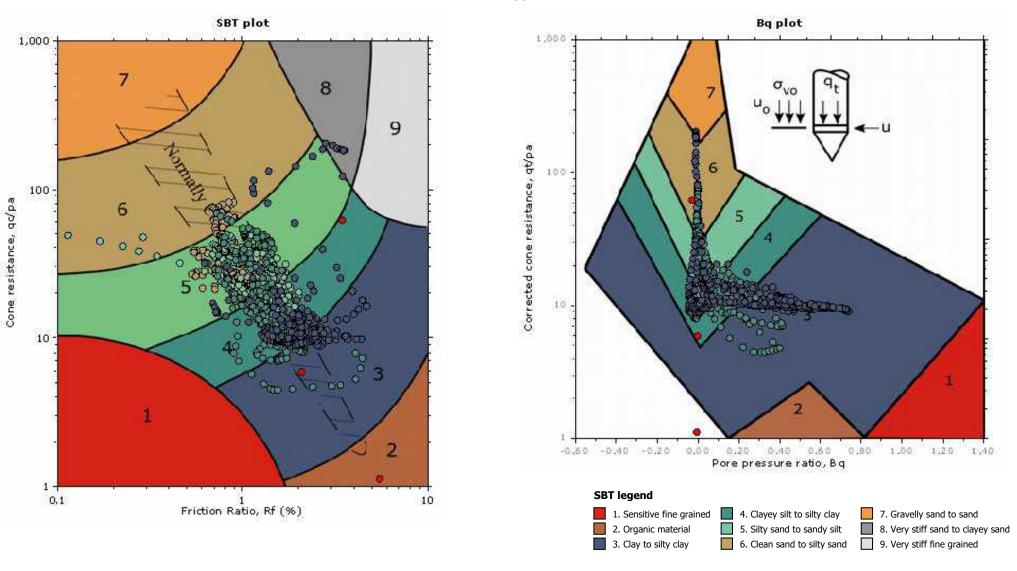
CPT: KP CPT04

Total depth: 18.04 m, Date: 3/09/2020 Coords: lat 305888.1° lon 6880472.1°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

SBT - Bq plots



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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

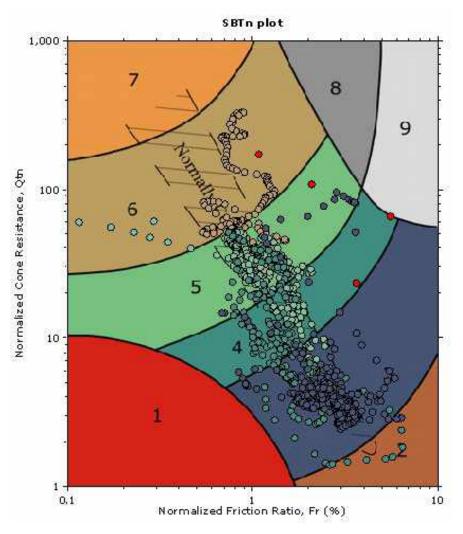
CPT: KP CPT04

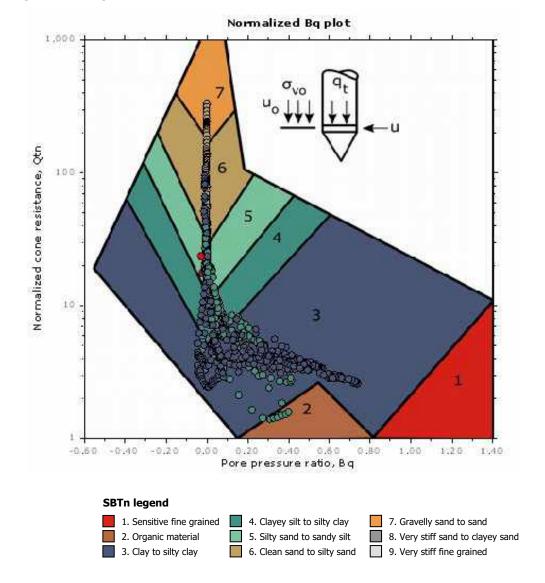
Total depth: 18.04 m, Date: 3/09/2020 Coords: lat 305888.1° lon 6880472.1°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

SBT - Bq plots (normalized)







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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

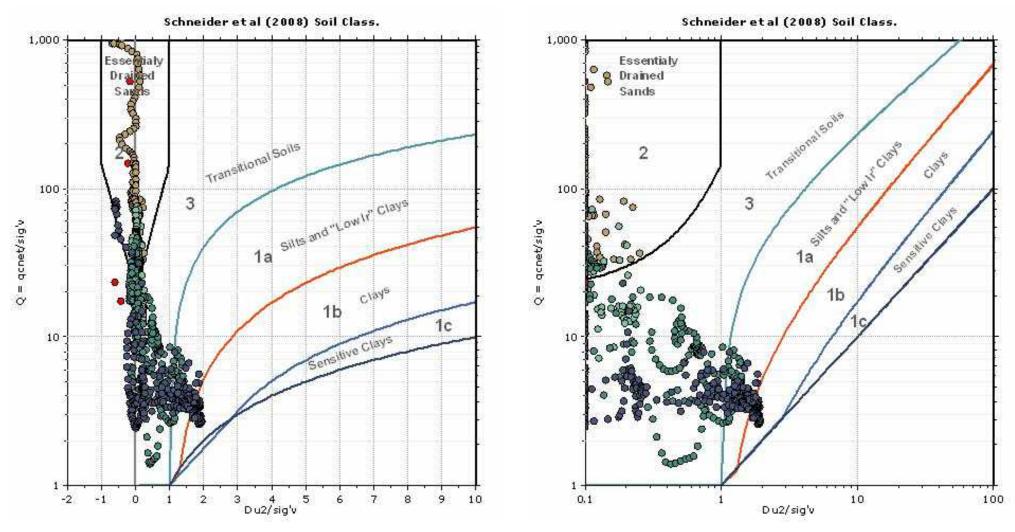
CPT: KP CPT04

Total depth: 18.04 m, Date: 3/09/2020 Coords: lat 305888.1° lon 6880472.1°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

Bq plots (Schneider)







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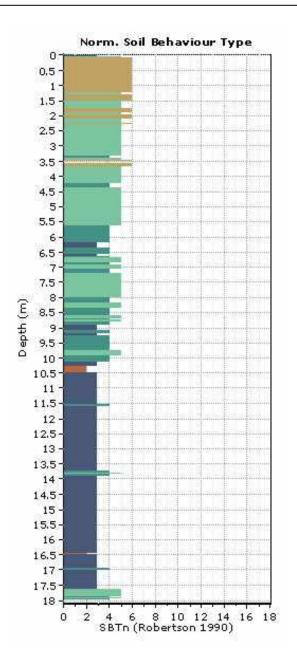
Thunderbox Gold Mine Project:

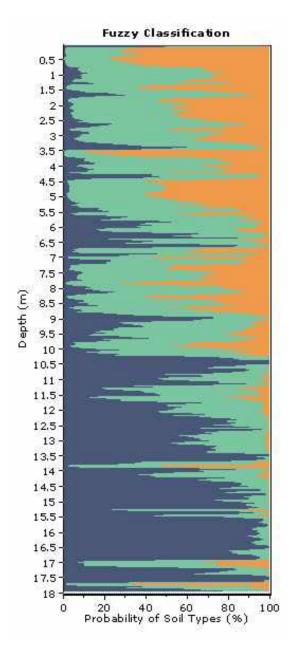
Location: Leinster, Western Australia, Australia

Total depth: 18.04 m, Date: 3/09/2020 Coords: lat 305888.1° lon 6880472.1°

Cone Type: 180911

CPT: KP CPT04





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Project: Thunderbox Gold Mine

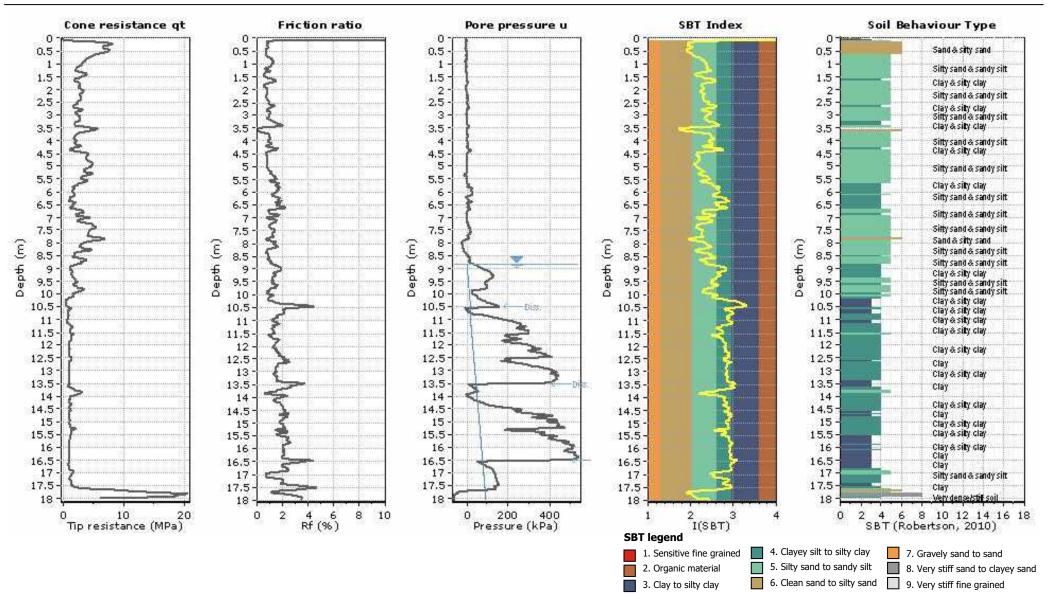
Location: Leinster, Western Australia, Australia

CPT: KP CPT04

Total depth: 18.04 m, Date: 3/09/2020

Coords: lat 305888.1° lon 6880472.1°

Cone Type: 180911



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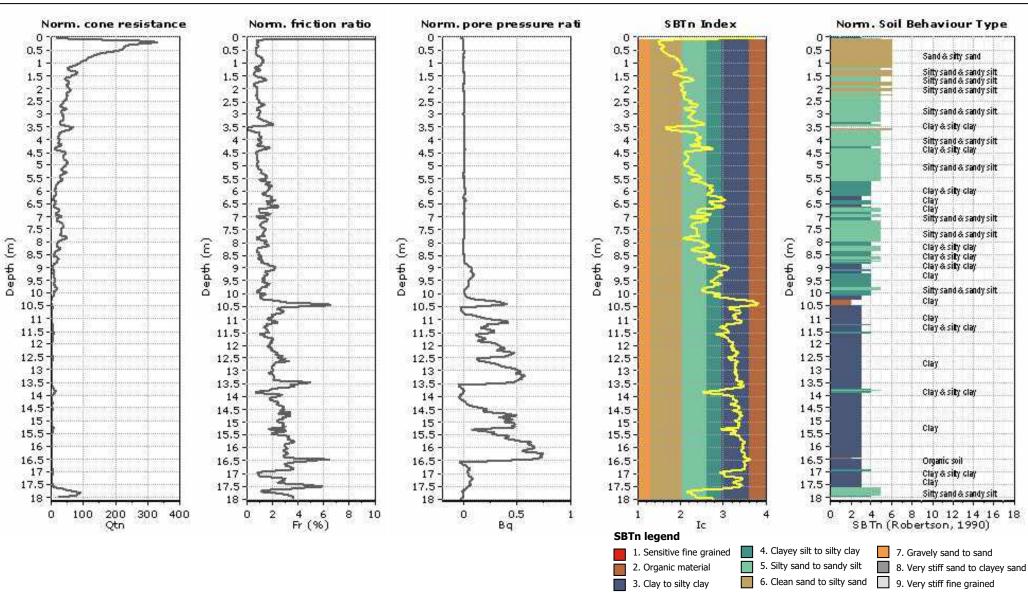
Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT04

Total depth: 18.04 m, Date: 3/09/2020 Coords: lat 305888.1° lon 6880472.1°

Cone Type: 180911



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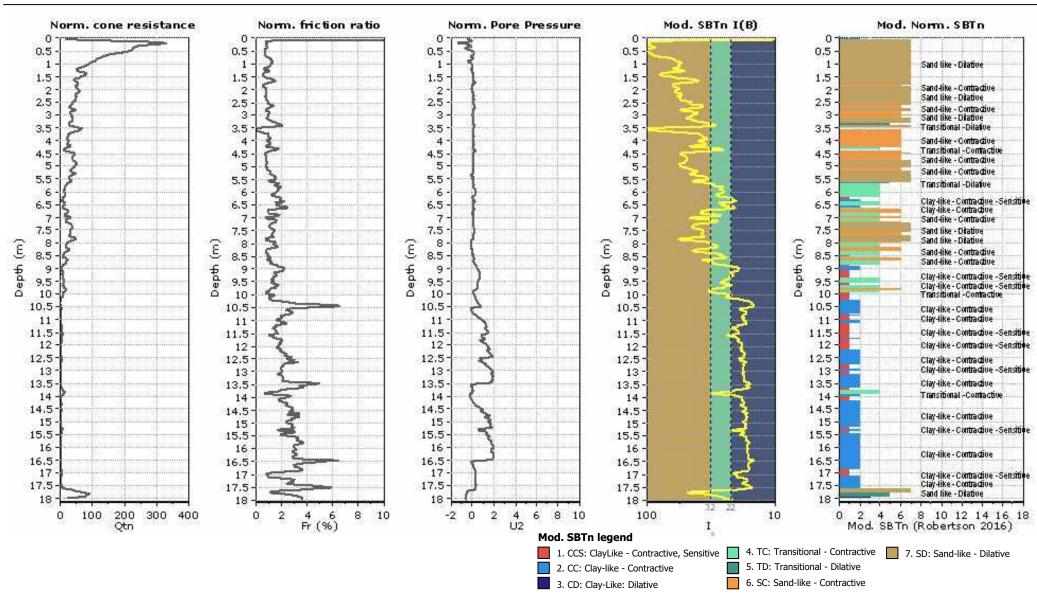
Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT04

Total depth: 18.04 m, Date: 3/09/2020 Coords: lat 305888.1° lon 6880472.1°

Cone Type: 180911



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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

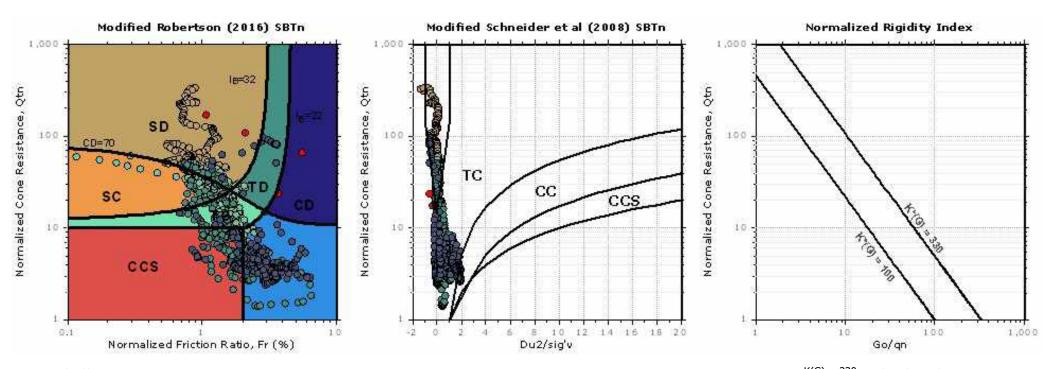
CPT: KP CPT04

Total depth: 18.04 m, Date: 3/09/2020 Coords: lat 305888.1° lon 6880472.1°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

Updated SBTn plots



K(G) > 330: Soils with significant microstructure (e.g. age/cementation)

CCS: Clay-like - Contractive - Sensitive

CC: Clay-like - Contractive CD: Clay-like - Dilative

TC: Transitional - Contractive
TD: Transitional - Dilative
SC: Sand-like - Contractive

SD: Sand-like - Dilative

Knight Piésold Level

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Project: Thunderbox Gold Mine

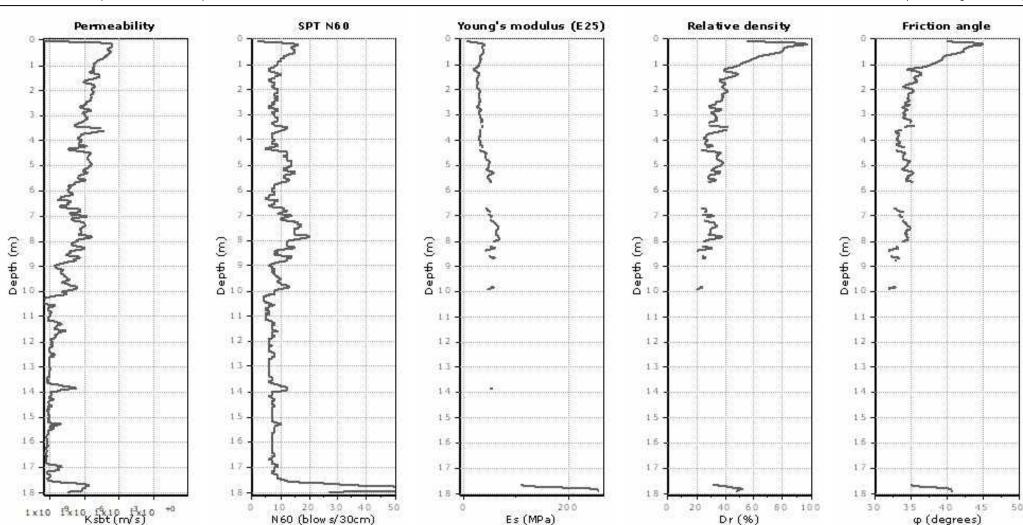
Location: Leinster, Western Australia, Australia

CPT: KP CPT04

Total depth: 18.04 m, Date: 3/09/2020 Coords: lat 305888.1° lon 6880472.1°

Cone Type: 180911

Cone Operator: Hagstrom Drilling



Calculation parameters

Permeability: Based on SBT_n SPT N_{60} : Based on I_c and q_t

Young's modulus: Based on variable alpha using I_c (Robertson, 2009)

Relative density constant, C_{Dr}: 350.0 Phi: Based on Kulhawy & Mayne (1990) — User defined estimation data

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Project: Thunderbox Gold Mine

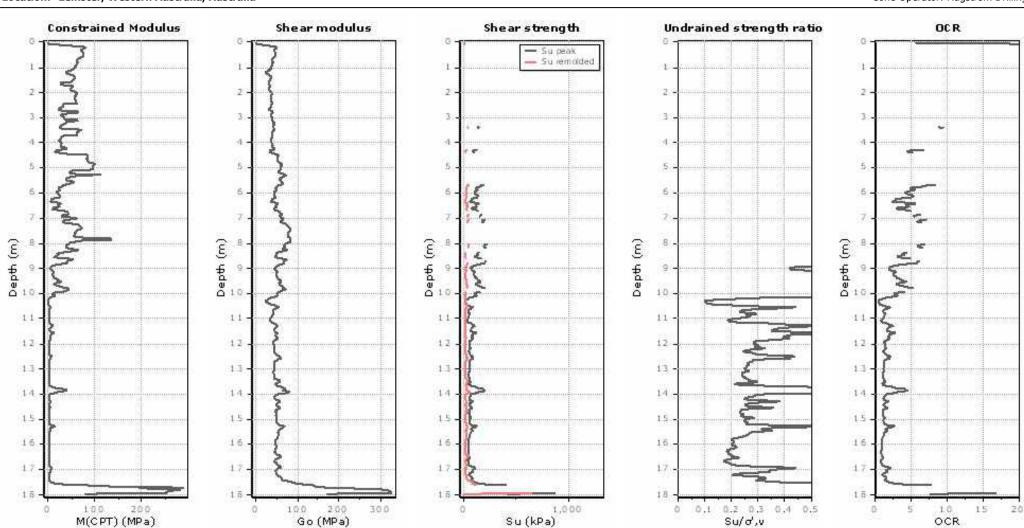
Location: Leinster, Western Australia, Australia

CPT: KP CPT04

Total depth: 18.04 m, Date: 3/09/2020 Coords: lat 305888.1° lon 6880472.1°

Cone Type: 180911

Cone Operator: Hagstrom Drilling



Calculation parameters

Constrained modulus: Based on variable alpha using $\,I_c$ and $\,Q_{tn}$ (Robertson, 2009) Go: Based on variable alpha using $\,I_c$ (Robertson, 2009)

Undrained shear strength cone factor for clays, N_{kt}: 14

OCR factor for clays, N_{kt}: 0.33

User defined estimation data

— Flat Dilatometer Test data

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Project: Thunderbox Gold Mine

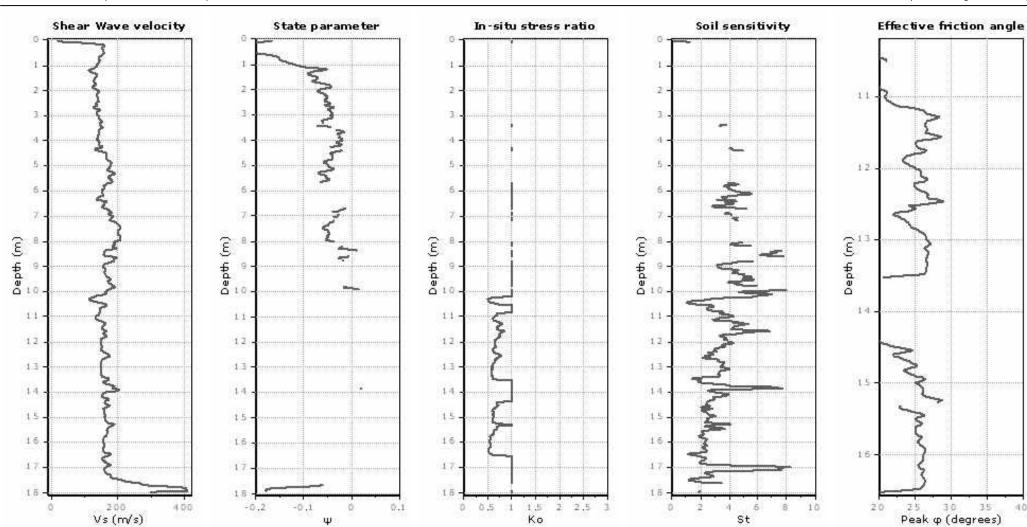
Location: Leinster, Western Australia, Australia

CPT: KP CPT04

Total depth: 18.04 m, Date: 3/09/2020 Coords: lat 305888.1° lon 6880472.1°

Cone Type: 180911

Cone Operator: Hagstrom Drilling



Calculation parameters

Soil Sensitivity factor, N_S: 7.00

User defined estimation data



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Thunderbox Gold Mine Project:

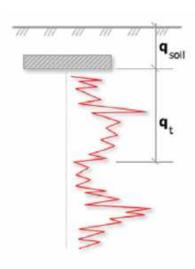
Location: Leinster, Western Australia, Australia

Total depth: 18.04 m, Date: 3/09/2020 Coords: lat 305888.1° lon 6880472.1°

Cone Type: 180911

CPT: KP CPT04

Cone Operator: Hagstrom Drilling

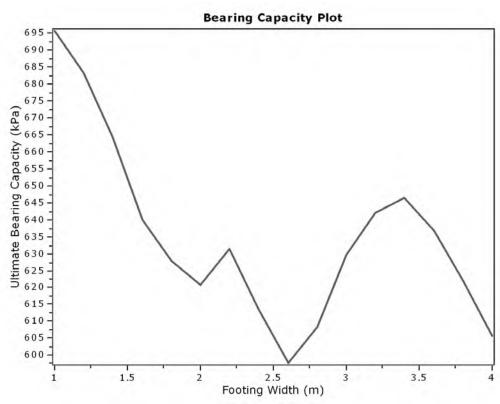


Bearing Capacity calculation is perfromed based on the formula:

$$Q_{ult} = R_k \times q_t + q_{soil}$$

where:

R_k: Bearing capacity factor qt: Average corrected cone resistance over calculation depth q_{soil}: Pressure applied by soil above footing



:: Tabul	ar results ::						
No	B (m)	Start Depth (m)	End Depth (m)	Ave. q _t (MPa)	R _k	Soil Press. (kPa)	Ult. bearing cap. (kPa)
1	1.00	0.50	2.00	3.43	0.20	9.50	695.72
2	1.20	0.50	2.30	3.37	0.20	9.50	683.28
3	1.40	0.50	2.60	3.27	0.20	9.50	664.49
4	1.60	0.50	2.90	3.15	0.20	9.50	639.97
5	1.80	0.50	3.20	3.09	0.20	9.50	627.94
6	2.00	0.50	3.50	3.06	0.20	9.50	620.64
7	2.20	0.50	3.80	3.11	0.20	9.50	631.47
8	2.40	0.50	4.10	3.02	0.20	9.50	613.64
9	2.60	0.50	4.40	2.94	0.20	9.50	597.71
10	2.80	0.50	4.70	2.99	0.20	9.50	608.19
11	3.00	0.50	5.00	3.10	0.20	9.50	629.70
12	3.20	0.50	5.30	3.16	0.20	9.50	641.92
13	3.40	0.50	5.60	3.18	0.20	9.50	646.41
14	3.60	0.50	5.90	3.14	0.20	9.50	636.59
15	3.80	0.50	6.20	3.06	0.20	9.50	621.82
16	4.00	0.50	6.50	2.98	0.20	9.50	605.71

Total depth: 18.04 m, Date: 3/09/2020 Coords: lat 305888.1° lon 6880472.1°

Cone Type: 180911

CPT: KP CPT04

Cone Operator: Hagstrom Drilling

Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

Dissipation Tests Results

Dissipation tests

Dissipation tests consists of stopping the piezocone penetration and observing porepressures (u) with elapsed time (t). The data are automatic recorded by the field computer and should take place until a minimum of 50% dissipation.

The porepressures are plotted as a function of square root of (t). The graphical technique suggested by Robertson and Campanella (1989), yields a value for t_{50} , which corresponds to the time for 50% consolidation.

The value of the coefficient of consolidation in the radial or horizontal direction c_h was then calculated by Houlsby and Teh's (1988) theory using the following equation:

$$c_h = \frac{T \times r^2 \times I_r^{0.5}}{t_{50}}$$

where:

T: time factor given by Houlsby and Teh's (1988) theory corresponding to the porepressure position

r: piezocone radius

I.: stiffness index, equal to shear modulus G divided by the undrained strength of clay (S_u).

t₅₀: time corresponding to 50% consolidation

Permeability estimates based on dissipation test

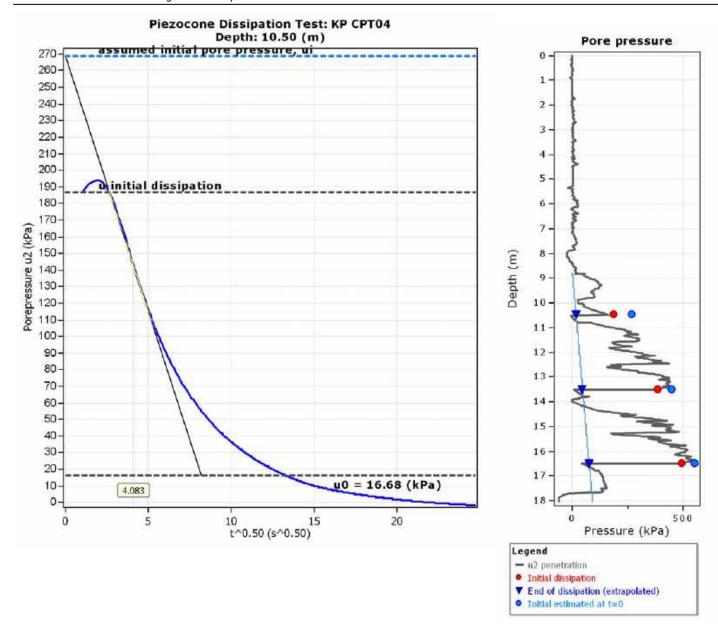
The dissipation of pore pressures during a CPTu dissipation test is controlled by the coefficient of consolidation in the horizontal direction (c_h) which is influenced by a combination of the soil permeability (k_h) and compressibility (M), as defined by the following:

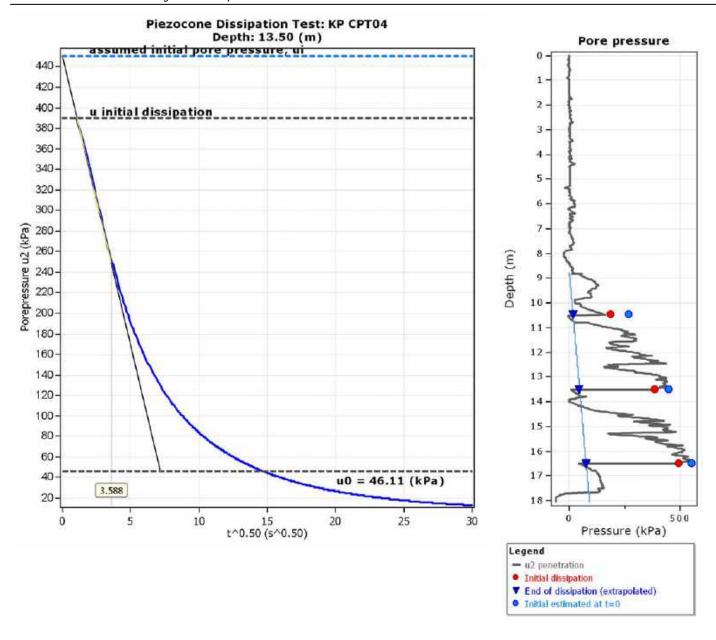
$$k_h = c_h \times \gamma_w / M$$

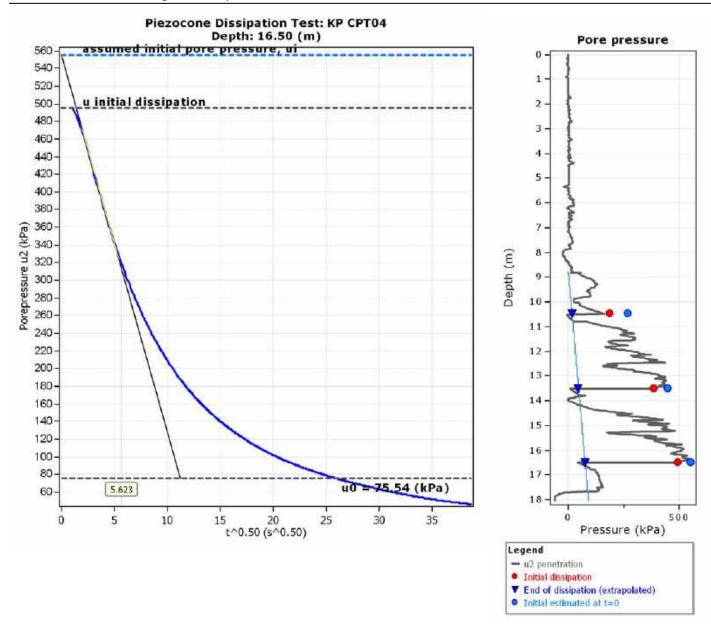
where: M is the 1-D constrained modulus and γ_w is the unit weight of water, in compatible units.

Tabular results

CPTU Borehole	Depth (m)	(t ₅₀) ^{0.50}	t ₅₀ (s)	t ₅₀ (years)	G/S _u	C _h (m²/s)	C _h (m²/year)	M (MPa)	k _h (m/s)
KP CPT04	10.50	4.1	17	5.29E-007	100.00	4.66E-005	1468	6.11	7.48E-008
KP CPT04	13.50	3.6	13	4.08E-007	100.00	6.03E-005	1902	5.06	1.17E-007
KP CPT04	16.50	5.6	32	1.00E-006	100.00	2.45E-005	774	2.73	8.81E-008









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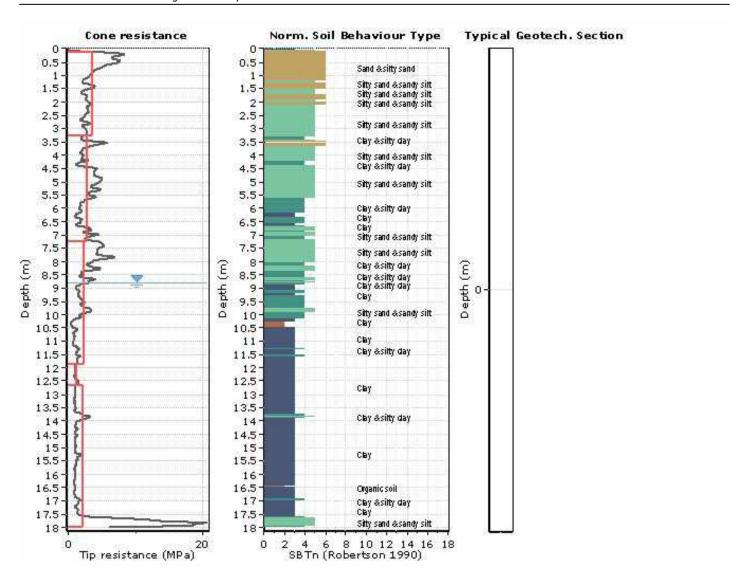
Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT04

Total depth: 18.04 m, Date: 3/09/2020 Coords: lat 305888.1° lon 6880472.1°

Cone Type: 180911



Tabular results

.:: Layer No: 1 ::.

Code: Layer_1 Start depth: 0.10 (m), End depth: 3.26 (m)

Description: Sand & silty sand

Basic results

Total cone resistance: 3.59 ± 1.71 MPa Sleeve friction: 31.35 ± 13.41 kPa

Ic: 2.01 ±0.25 SBT_n: 6

SBTn description: Sand & silty sand

Estimation results

Permeability: $3.46E-05 \pm 6.60E-05 \text{ m/s}$ Constrained Mod.: $53.07 \pm 14.90 \text{ MPa}$

Unit weight: $19.00 \pm 0.00 \text{ kN/m}^3$

.:: Layer No: 2 ::.

Code: Layer_2 **Start depth:** 3.26 (m), **End depth:** 7.22 (m)

Description: Silty sand & sandy silt

Basic results

Estimation results

Total cone resistance: 2.85 ±1.16 MPa

Sleeve friction: 30.45 ±10.82 kPa Ic: 2.43 ±0.28

SBT_n: 5 SBTn description: Silty sand & sandy silt

Permeability: 1.58E-06 ±5.06E-06 m/s N_{60} : 9.70 ±2.58 blows

Es: 41.61 ±8.41 MPa Dr (%): 31.36 ±4.30 φ (degrees): 33.91 ±0.71 °

Unit weight: 19.00 ±0.00 kN/m³

Constrained Mod.: 43.49 ±25.53 MPa Go: 50.19 ±9.82 MPa

CPT name: KP CPT04

Su: $0.00 \pm 0.00 \text{ kPa}$ Su ratio: 0.00 ±0.00

O.C.R.: 0.00 ±0.00

.:: Layer No: 3 ::.

Code: Layer_3

Start depth: 7.22 (m), **End depth:** 11.86 (m)

Description: Clay & silty clay

Basic results

Estimation results

Total cone resistance: 2.28 ±1.48 MPa Sleeve friction: 27.95 ±15.23 kPa

Ic: 2.86 ±0.39 SBT_n: 4

SBTn description: Clay & silty clay

Permeability: 2.30E-07 ±5.66E-07 m/s

 N_{60} : 9.48 ±3.89 blows Es: 0.00 ±0.00 MPa Dr (%): 0.00 ±0.00 φ (degrees): 0.00 ±0.00 °

Unit weight: 19.00 ±0.00 kN/m³

Constrained Mod.: 25.67 ±27.29 MPa Go: 55.86 ±15.02 MPa

Su: 92.70 ±47.12 kPa Su ratio: 0.54 ±0.31 O.C.R.: 2.48 ±1.44

.:: Layer No: 4 ::.

Code: Layer_4

Start depth: 11.86 (m), **End depth:** 12.62 (m)

Description: Clay

Basic results

Estimation results

Total cone resistance: 1.11 ±0.16 MPa Sleeve friction: 19.70 ±6.68 kPa

Ic: 3.23 ± 0.06 SBT_n: 3

SBTn description: Clay

Permeability: $1.47E-09 \pm 5.12E-10 \text{ m/s}$

 N_{60} : 6.59 ±0.82 blows Es: 0.00 ±0.00 MPa Dr (%): 0.00 ±0.00 ϕ (degrees): 0.00 ±0.00 °

Unit weight: 19.00 ±0.00 kN/m3

Constrained Mod.: 4.01 ±1.51 MPa

Go: 47.08 ±6.13 MPa Su: 62.88 ±11.42 kPa Su ratio: 0.32 ±0.06 O.C.R.: 1.46 ±0.25

.:: Layer No: 5 ::.

Code: Layer_5

Start depth: 12.62 (m), **End depth:** 17.96 (m)

Description: Clay

Basic results

Estimation results

Total cone resistance: 2.10 ±3.57 MPa

Sleeve friction: 49.37 ±112.24 kPa Ic: 3.25 ± 0.30 SBT_n: 3

SBTn description: Clay

Permeability: 7.01E-08 ±3.00E-07 m/s

 N_{60} : 10.24 ±10.99 blows Es: 0.00 ±0.00 MPa

Dr (%): 0.00 ±0.00 ϕ (degrees): 0.00 ±0.00 °

Unit weight: $19.00 \pm 0.00 \text{ kN/m}^3$

Constrained Mod.: 18.30 ±54.72 MPa

Go: 65.50 ±51.92 MPa Su: 75.47 ±98.42 kPa Su ratio: 0.33 ± 0.43 O.C.R.: 1.55 ±1.99



http://www.knightpiesold.com/

Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT04

Total depth: 18.04 m, Date: 3/09/2020 Coords: lat 305888.1° lon 6880472.1°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

Summary table of mean values

From depth To depth (m)	Thickness (m)	Permeability (m/s)	SPT _{N60} (blows/30cm)	E _s (MPa)	D _r (%)	Friction angle	Constrained modulus, M (MPa)	Shear modulus, G _o (MPa)	Undrained strength, S _u (kPa)	Undrained strength ratio	OCR	Unit weight (kN/m³)
0.10	3.16	3.46E-05	9.2	29.8	47.8	36.7	53.1	37.4	0.0	0.0	0.0	19.0
3.26	5.10	(±6.60E-05)	(±2.6)	(±4.4)	(±18.0)	(±3.2)	(±14.9)	(±5.5)	(±0.0)	(±0.0)	(±0.0)	(±0.0)
3.26	3.96	1.58E-06	9.7	41.6	31.4	33.9	43.5	50.2	0.0	0.0	0.0	19.0
7.22		(±5.06E-06)	(±2.6)	(±8.4)	(±4.3)	(±0.7)	(±25.5)	(±9.8)	(±0.0)	(±0.0)	(±0.0)	(±0.0)
7.22	4.64	2.30E-07	9.5	0.0	0.0	0.0	25.7	55.9	92.7	0.5	2.5	19.0
11.86	7.07	(±5.66E-07)	(±3.9)	(±0.0)	(±0.0)	(±0.0)	(±27.3)	(±15.0)	(±47.1)	(±0.3)	(±1.4)	(±0.0)
11.86	0.76	1.47E-09	6.6	0.0	0.0	0.0	4.0	47.1	62.9	0.3	1.5	19.0
12.62	0.70	(±5.12E-10)	(±0.8)	(±0.0)	(±0.0)	(±0.0)	(±1.5)	(±6.1)	(±11.4)	(±0.1)	(±0.3)	(±0.0)
12.62	5.34	7.01E-08	10.2	0.0	0.0	0.0	18.3	65.5	75.5	0.3	1.5	19.0
17.96	5.51	(±3.00E-07)	(±11.0)	(±0.0)	(±0.0)	(±0.0)	(±54.7)	(±51.9)	(±98.4)	(±0.4)	(±2.0)	(±0.0)

Depth values presented in this table are measured from free ground surface

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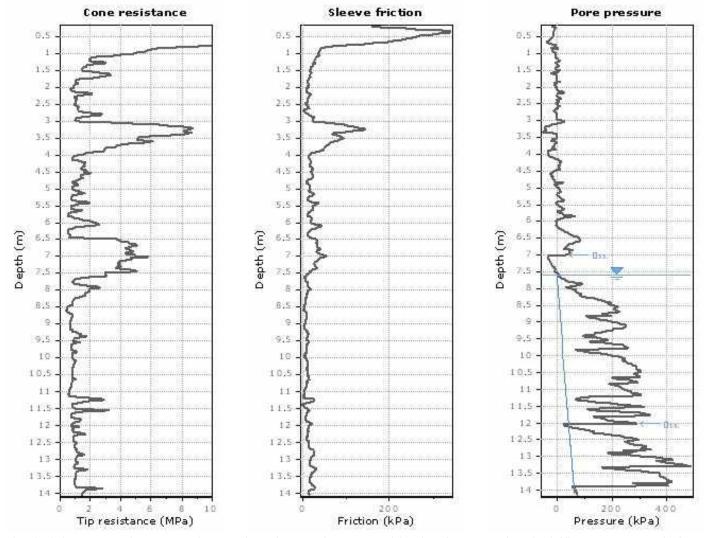
Level 1, 184 Adelaide Terrace

CPT: KP CPT05A

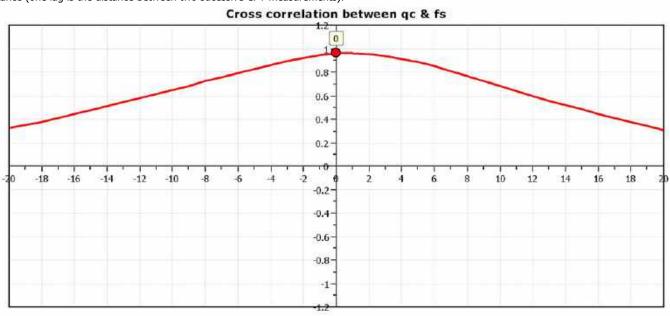
Total depth: 14.12 m, Date: 4/09/2020 Coords: lat 305799° lon 6879943°

Cone Type: 180911





The plot below presents the cross correlation coeficient between the raw qc and fs values (as measured on the field). X axes presents the lag distance (one lag is the distance between two sucessive CPT measurements).



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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT05A

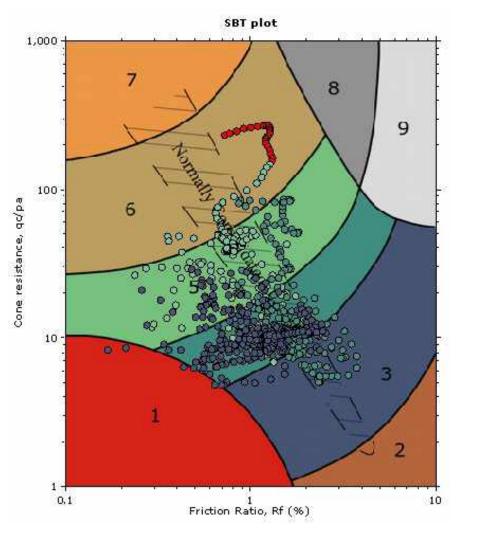
Total depth: 14.12 m, Date: 4/09/2020

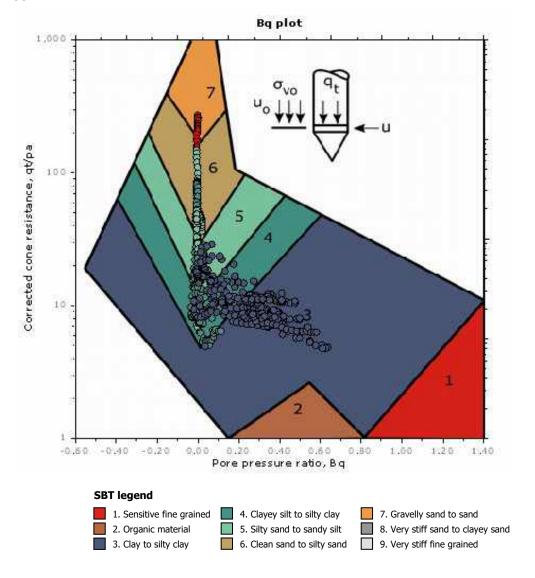
Coords: lat 305799° lon 6879943°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

SBT - Bq plots





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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT05A

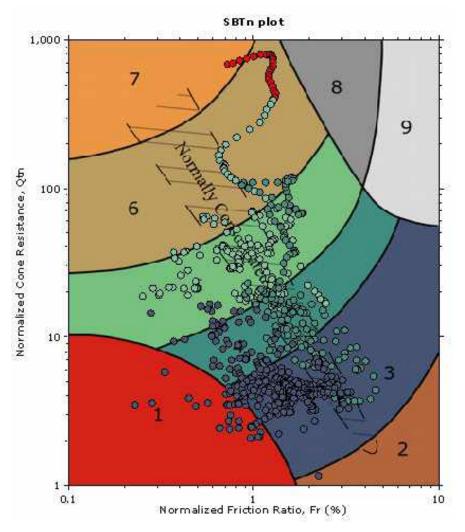
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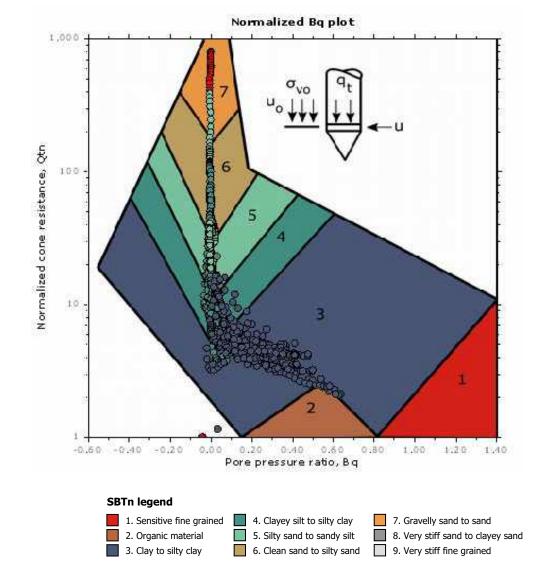
Coords: lat 305799° lon 6879943°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

SBT - Bq plots (normalized)







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Total depth: 14.12 m, Date: 4/09/2020

Coords: lat 305799° lon 6879943°

Cone Type: 180911

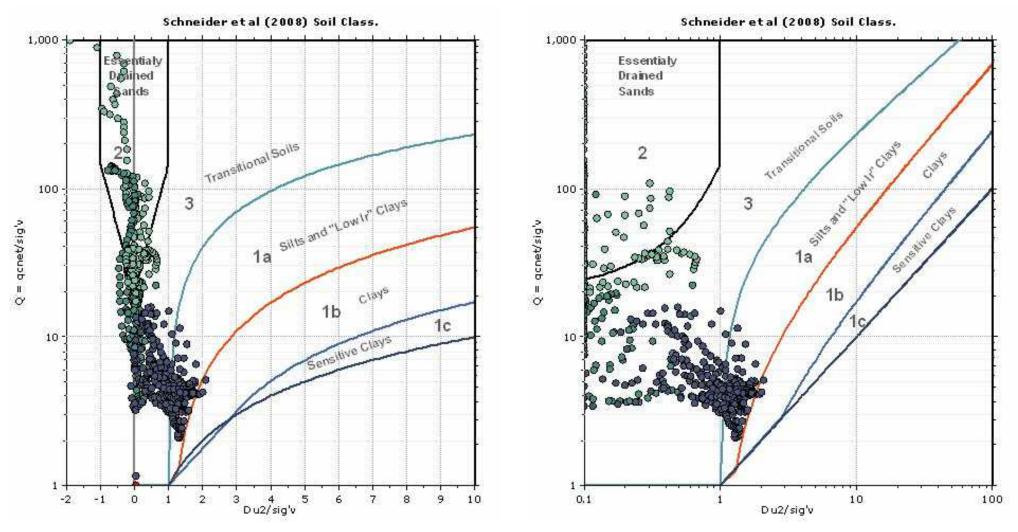
CPT: KP CPT05A

Cone Operator: Hagstrom Drilling

Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

Bq plots (Schneider)







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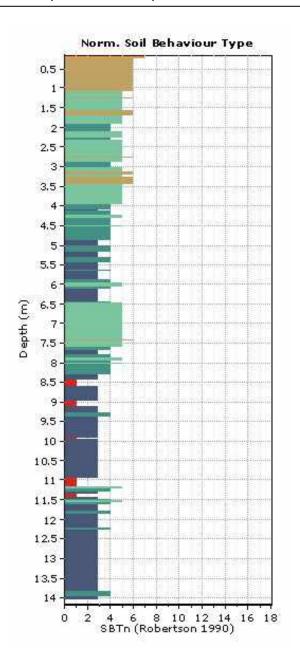
Thunderbox Gold Mine Project:

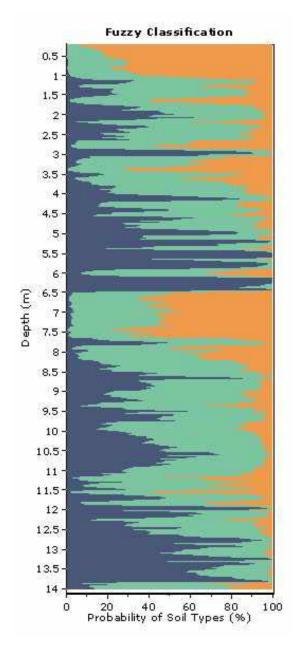
Location: Leinster, Western Australia, Australia

CPT: KP CPT05A

Total depth: 14.12 m, Date: 4/09/2020 Coords: lat 305799° lon 6879943°

Cone Type: 180911





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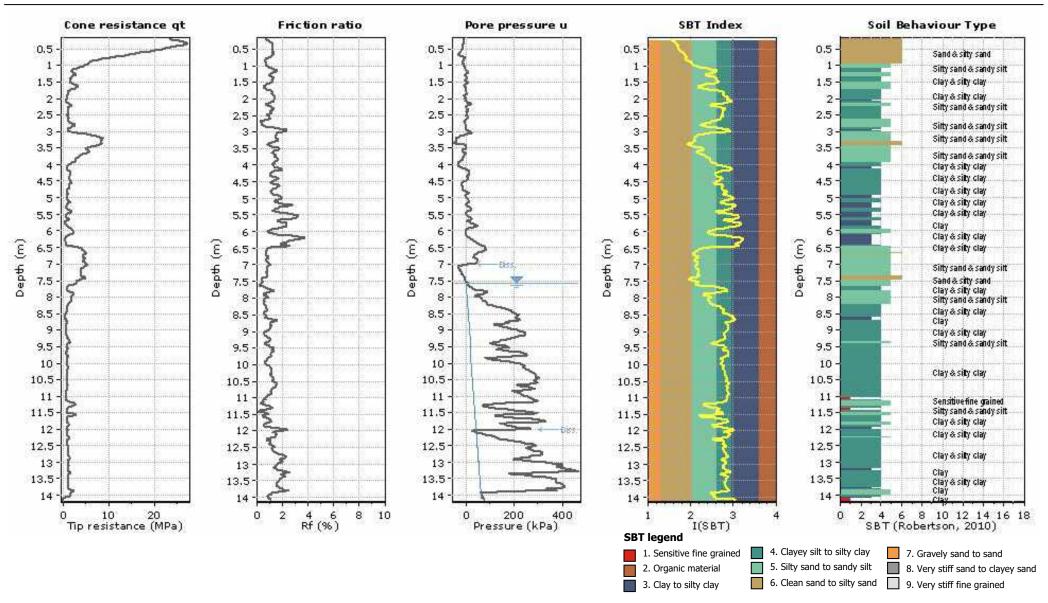
Location: Leinster, Western Australia, Australia

CPT: KP CPT05A

Total depth: 14.12 m, Date: 4/09/2020

Coords: lat 305799° lon 6879943°

Cone Type: 180911



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Project: Thunderbox Gold Mine

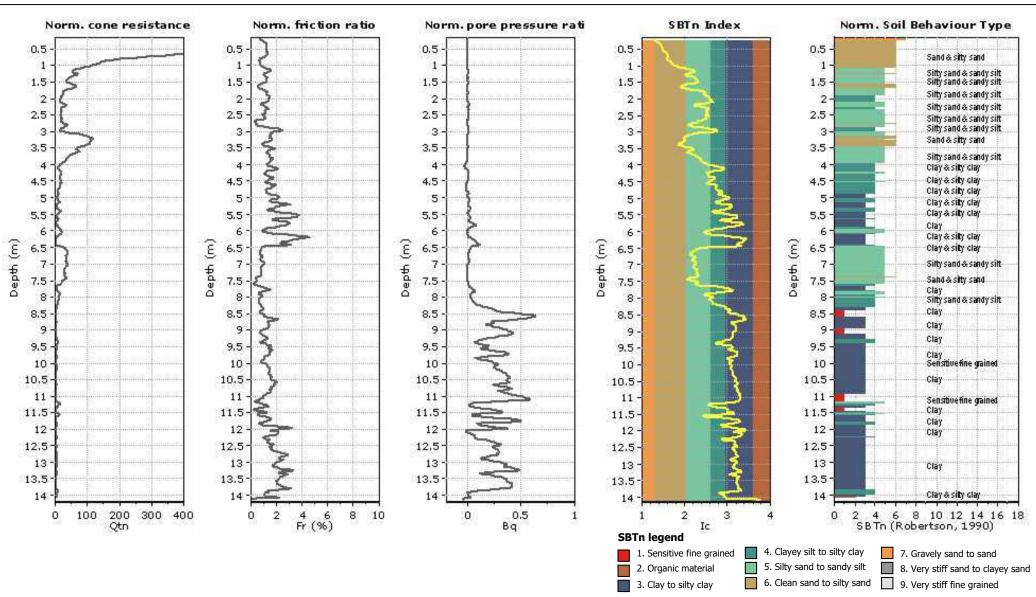
Location: Leinster, Western Australia, Australia

CPT: KP CPT05A

Total depth: 14.12 m, Date: 4/09/2020

Coords: lat 305799° lon 6879943°

Cone Type: 180911



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Project: Thunderbox Gold Mine

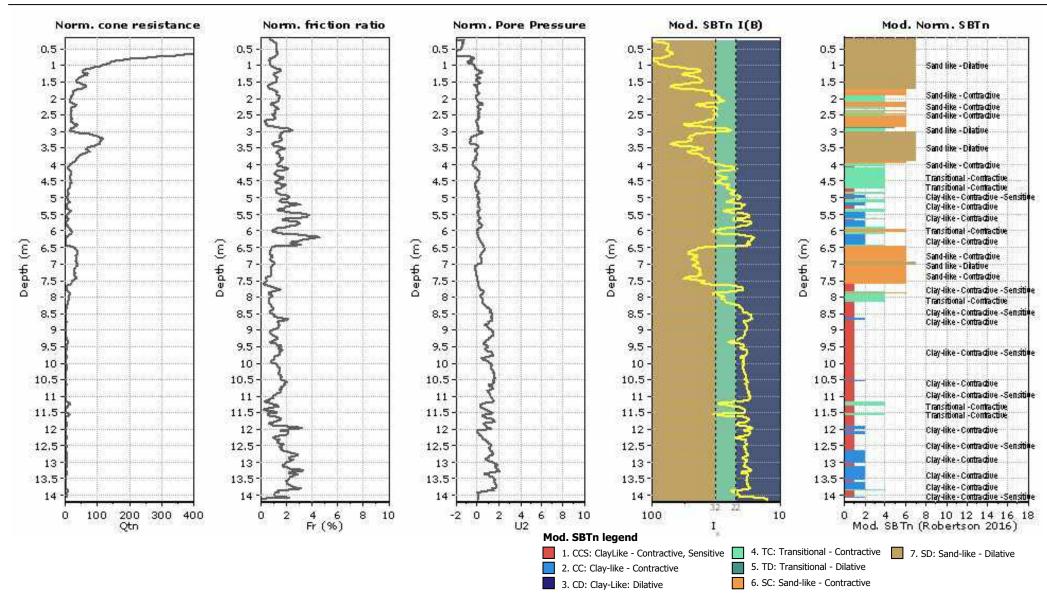
Location: Leinster, Western Australia, Australia

CPT: KP CPT05A

Total depth: 14.12 m, Date: 4/09/2020

Coords: lat 305799° lon 6879943°

Cone Type: 180911



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Total depth: 14.12 m, Date: 4/09/2020

Coords: lat 305799° lon 6879943°

Cone Type: 180911

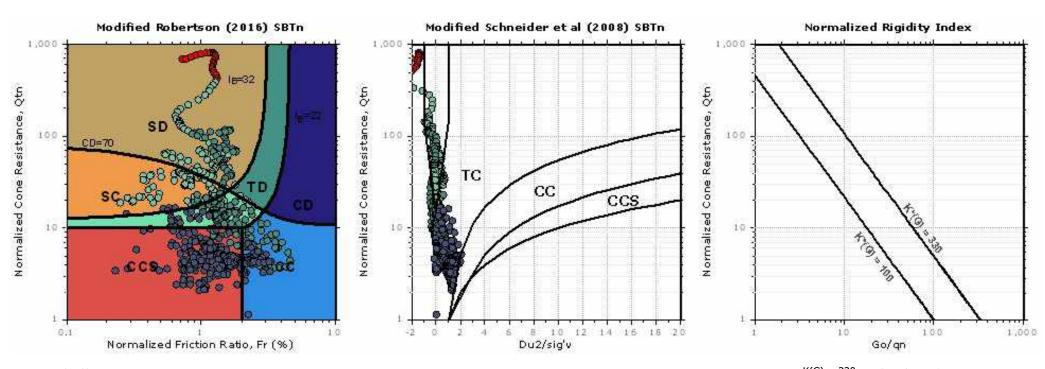
CPT: KP CPT05A

Cone Operator: Hagstrom Drilling

Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

Updated SBTn plots



K(G) > 330: Soils with significant microstructure (e.g. age/cementation)

CCS: Clay-like - Contractive - Sensitive

Clay-like - Contractive Clay-like - Dilative

Transitional - Contractive TD: Transitional - Dilative

SC: Sand-like - Contractive

Sand-like - Dilative

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Total depth: 14.12 m, Date: 4/09/2020

Coords: lat 305799° lon 6879943°

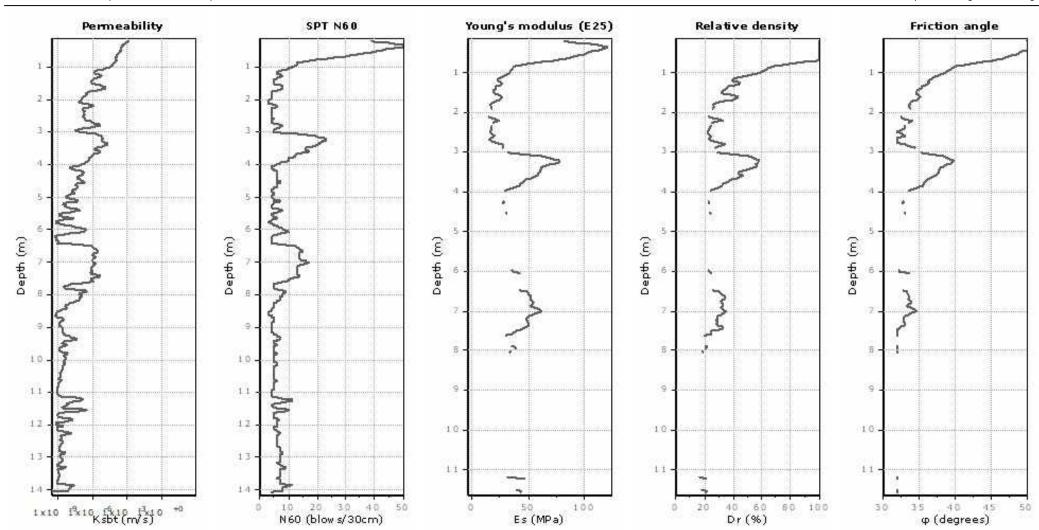
Cone Type: 180911

CPT: KP CPT05A

Cone Operator: Hagstrom Drilling

Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia



Calculation parameters

Permeability: Based on SBT_n SPT N₆₀: Based on I_c and q_t

Young's modulus: Based on variable alpha using I_c (Robertson, 2009)

Relative density constant, C_{Dr}: 350.0 Phi: Based on Kulhawy & Mayne (1990) ____ User defined estimation data

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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

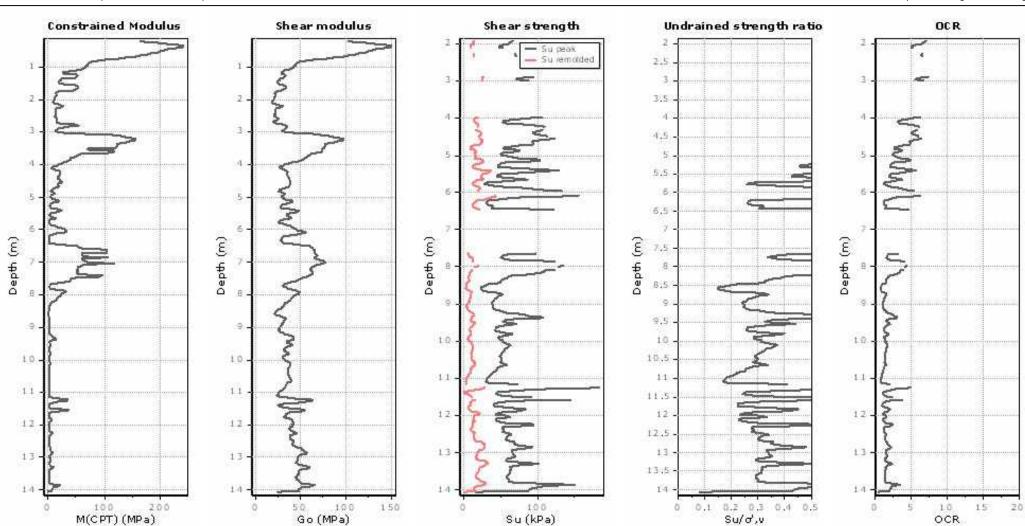
CPT: KP CPT05A

Total depth: 14.12 m, Date: 4/09/2020

Coords: lat 305799° lon 6879943°

Cone Type: 180911

Cone Operator: Hagstrom Drilling



Calculation parameters

Constrained modulus: Based on variable alpha using I_c and Q_{tn} (Robertson, 2009) Go: Based on variable alpha using I_c (Robertson, 2009) Undrained shear strength cone factor for clays, N_{kt}: 14

OCR factor for clays, Nkt: 0.33

User defined estimation data Flat Dilatometer Test data

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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

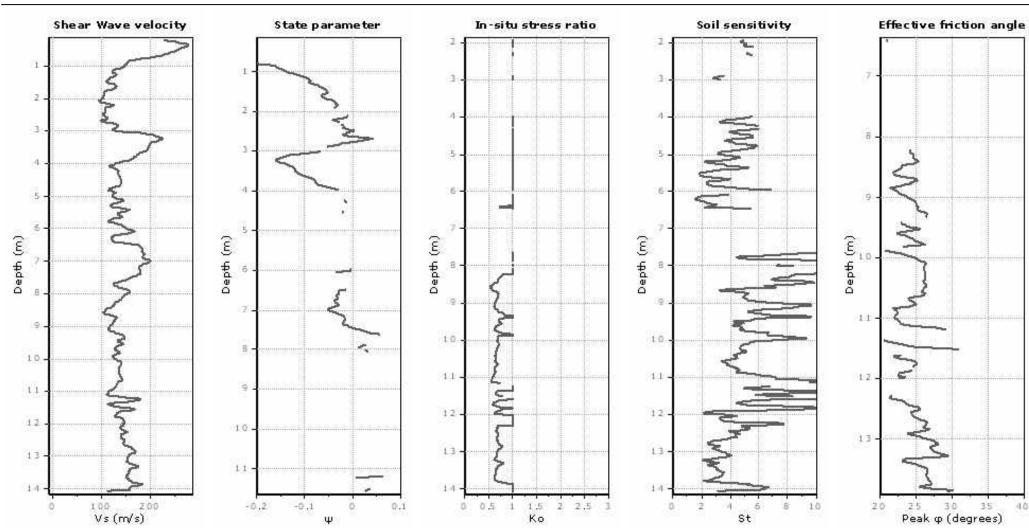
CPT: KP CPT05A

Total depth: 14.12 m, Date: 4/09/2020

Coords: lat 305799° lon 6879943°

Cone Type: 180911

Cone Operator: Hagstrom Drilling



Calculation parameters

Soil Sensitivity factor, N_S: 7.00

User defined estimation data



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CPT: KP CPT05A

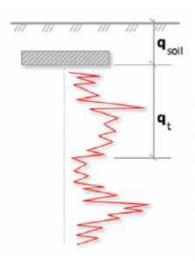
Total depth: 14.12 m, Date: 4/09/2020 Coords: lat 305799° lon 6879943°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

Thunderbox Gold Mine Project:

Location: Leinster, Western Australia, Australia



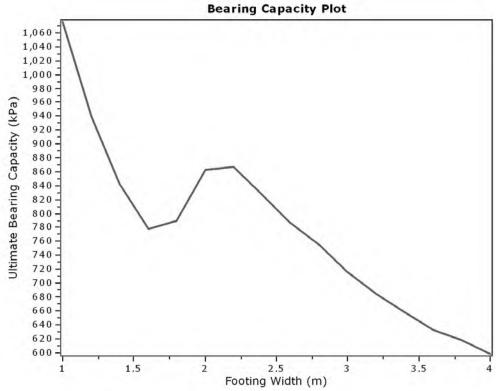
Bearing Capacity calculation is perfromed based on the formula:

$$Q_{ult} = R_k \times q_t + q_{soil}$$

where:

R_k: Bearing capacity factor qt: Average corrected cone resistance over calculation depth $q_{\text{soil}} \text{:} \text{ Pressure applied by soil}$

above footing



:: Tabul	ar results ::	1					
No	B (m)	Start Depth (m)	End Depth (m)	Ave. q _t (MPa)	R _k	Soil Press. (kPa)	Ult. bearing cap. (kPa)
1	1.00	0.50	2.00	5.33	0.20	9.50	1075.72
2	1.20	0.50	2.30	4.65	0.20	9.50	938.68
3	1.40	0.50	2.60	4.17	0.20	9.50	842.83
4	1.60	0.50	2.90	3.84	0.20	9.50	777.97
5	1.80	0.50	3.20	3.90	0.20	9.50	788.94
6	2.00	0.50	3.50	4.26	0.20	9.50	862.05
7	2.20	0.50	3.80	4.29	0.20	9.50	867.64
8	2.40	0.50	4.10	4.09	0.20	9.50	826.92
9	2.60	0.50	4.40	3.88	0.20	9.50	785.75
10	2.80	0.50	4.70	3.73	0.20	9.50	754.82
11	3.00	0.50	5.00	3.53	0.20	9.50	715.87
12	3.20	0.50	5.30	3.38	0.20	9.50	684.77
13	3.40	0.50	5.60	3.24	0.20	9.50	658.33
14	3.60	0.50	5.90	3.11	0.20	9.50	632.14
15	3.80	0.50	6.20	3.04	0.20	9.50	617.64
16	4.00	0.50	6.50	2.94	0.20	9.50	597.87

CPT: KP CPT05A

Total depth: 14.12 m, Date: 4/09/2020 Coords: lat 305799° lon 6879943°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

Dissipation Tests Results

Dissipation tests

Dissipation tests consists of stopping the piezocone penetration and observing porepressures (u) with elapsed time (t). The data are automatic recorded by the field computer and should take place until a minimum of 50% dissipation.

The porepressures are plotted as a function of square root of (t). The graphical technique suggested by Robertson and Campanella (1989), yields a value for t_{50} , which corresponds to the time for 50% consolidation.

The value of the coefficient of consolidation in the radial or horizontal direction c_h was then calculated by Houlsby and Teh's (1988) theory using the following equation:

$$c_h = \frac{T \times r^2 \times I_r^{0.5}}{t_{50}}$$

where:

T: time factor given by Houlsby and Teh's (1988) theory corresponding to the porepressure position

r: piezocone radius

 I_r : stiffness index, equal to shear modulus G divided by the undrained strength of clay (S_u) .

t₅₀: time corresponding to 50% consolidation

Permeability estimates based on dissipation test

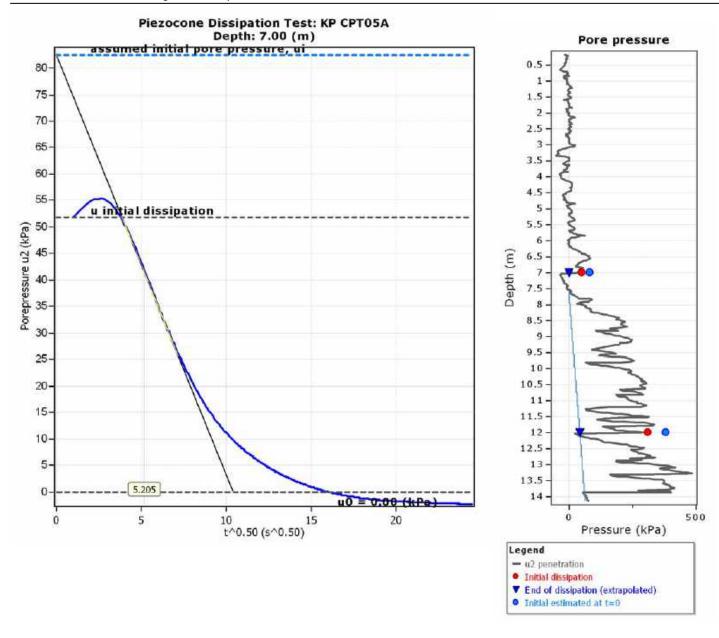
The dissipation of pore pressures during a CPTu dissipation test is controlled by the coefficient of consolidation in the horizontal direction (c_h) which is influenced by a combination of the soil permeability (k_h) and compressibility (M), as defined by the following:

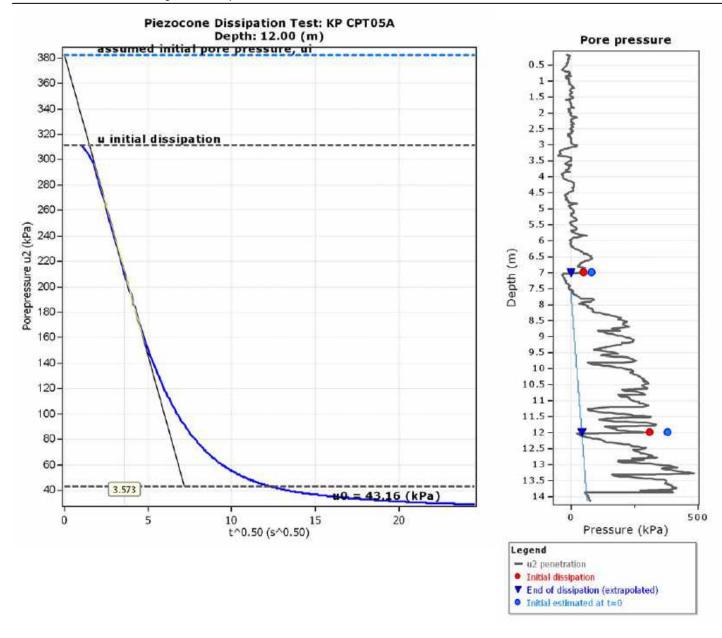
$$k_h = c_h \times \gamma_w / M$$

where: M is the 1-D constrained modulus and γ_w is the unit weight of water, in compatible units.

Tabular results

CPTU Borehole	Depth (m)	(t ₅₀) ^{0.50}	t ₅₀ (s)	t ₅₀ (years)	G/S _u	C _h (m²/s)	C _h (m²/year)	M (MPa)	k _h (m/s)
KP CPT05A	7.00	5.2	27	8.59E-007	100.00	2.87E-005	904	71.47	3.93E-009
KP CPT05A	12.00	3.6	13	4.05E-007	100.00	6.08E-005	1917	4.37	1.37E-007







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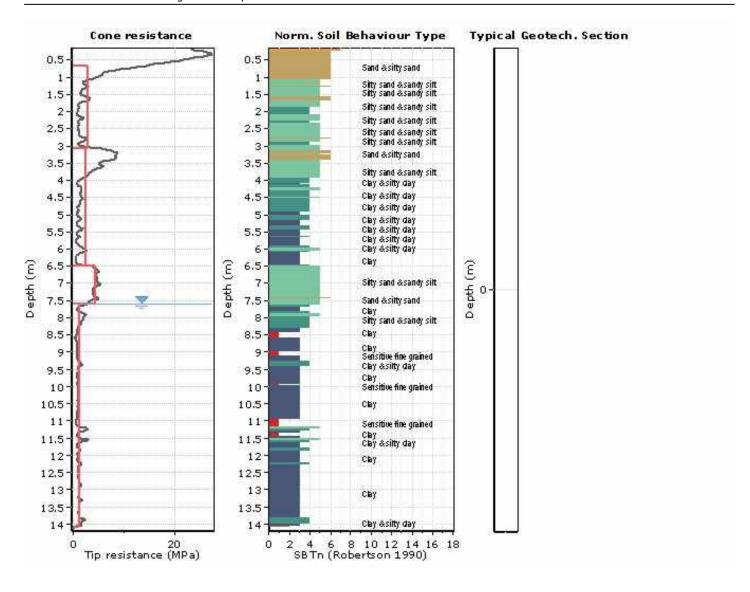
Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT05A

Total depth: 14.12 m, Date: 4/09/2020 Coords: lat 305799° lon 6879943°

Cone Type: 180911



Tabular results

.:: Layer No: 1 ::.

Code: Layer_1 Start depth: 0.66 (m), End depth: 3.06 (m)

Description: Silty sand & sandy silt

Basic results

Total cone resistance: 2.72 ±2.83 MPa Sleeve friction: 27.08 ±30.94 kPa

Ic: 2.24 ±0.32 SBT_n: 5

SBTn description: Silty sand & sandy silt

Estimation results

Permeability: 1.55E-05 ±3.45E-05 m/s Constrained Mod.: 37.12 ±31.54 MPa

 N_{60} : 7.36 ±5.39 blows Es: 26.74 ±13.04 MPa Dr (%): 40.27 ±19.10 ϕ (degrees): 35.54 ±3.33 °

Unit weight: 19.00 ±0.00 kN/m³

Go: 31.92 ±15.58 MPa Su: 0.00 ±0.00 kPa

Su ratio: 0.00 ±0.00 O.C.R.: 0.00 ±0.00

CPT name: KP CPT05A

O.C.R.: 3.41 ±1.50

Go: 38.59 ±8.85 MPa

.:: Layer No: 2 ::.

Start depth: 3.06 (m), **End depth:** 6.48 (m) Code: Layer_2

Description: Clay & silty clay

Basic results Estimation results

Total cone resistance: 2.36 ±2.29 MPa Permeability: 1.36E-06 ±3.47E-06 m/s Constrained Mod.: 34.68 ±43.61 MPa

 φ (degrees): 0.00 ±0.00 °

Sleeve friction: 35.24 ±31.12 kPa Go: 44.57 ±18.67 MPa N_{60} : 8.24 ±5.45 blows Ic: 2.69 ±0.41 Es: 0.00 ±0.00 MPa Su: 71.17 ±28.41 kPa SBT_n: 4 Dr (%): 0.00 ±0.00 Su ratio: 0.74 ±0.32

SBTn description: Clay & silty clay Unit weight: 19.00 ±0.00 kN/m³

.:: Layer No: 3 ::.

Code: Layer_3 Start depth: 6.48 (m), End depth: 7.60 (m)

Description: Silty sand & sandy silt

Basic results Estimation results

Total cone resistance: 4.25 ±0.64 MPa Permeability: 1.57E-06 ±8.60E-07 m/s Constrained Mod.: 65.87 ±21.55 MPa

Sleeve friction: 31.22 ±10.66 kPa N_{60} : 13.39 ±1.68 blows Go: 62.74 ±7.70 MPa Ic: 2.24 ±0.07 Es: 50.05 ±6.14 MPa Su: 0.00 ±0.00 kPa Dr (%): 30.54 ±2.83 Su ratio: 0.00 ±0.00 SBT_n: 5 O.C.R.: 0.00 ±0.00 SBTn description: Silty sand & sandy silt φ (degrees): 33.15 ±0.68 °

Unit weight: 19.00 ±0.00 kN/m³

.:: Layer No: 4 ::.

Code: Layer_4 **Start depth:** 7.60 (m), **End depth:** 14.06 (m)

Description: Clay

Sleeve friction: 12.33 ±6.62 kPa

Basic results Estimation results

Permeability: 1.92E-08 ±7.24E-08 m/s Total cone resistance: 1.11 ±0.43 MPa Constrained Mod.: 5.88 ±6.88 MPa

 N_{60} : 5.93 ±1.51 blows Ic: 3.09 ± 0.21 Es: 0.00 ±0.00 MPa Su: 60.72 ±24.21 kPa SBT_n: 3 Dr (%): 0.00 ±0.00 Su ratio: 0.35 ±0.15

SBTn description: Clay ϕ (degrees): 0.00 ±0.00 ° O.C.R.: 1.62 ±0.68

Unit weight: 19.00 ±0.00 kN/m³



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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT05A

Total depth: 14.12 m, Date: 4/09/2020 Coords: lat 305799° lon 6879943°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

Summary table of mean values

From depth To depth (m)	Thickness (m)	Permeability (m/s)	SPT _{N60} (blows/30cm)	E₅ (MPa)	D _r (%)	Friction angle	Constrained modulus, M (MPa)	Shear modulus, G _o (MPa)	Undrained strength, S _U (kPa)	Undrained strength ratio	OCR	Unit weight (kN/m³)
0.66	2.40	1.55E-05	7.4	26.7	40.3	35.5	37.1	31.9	0.0	0.0	0.0	19.0
3.06	2.70	(±3.45E-05)	(±5.4)	(±13.0)	(±19.1)	(±3.3)	(±31.5)	(±15.6)	(±0.0)	(±0.0)	(±0.0)	(±0.0)
3.06	3.42	1.36E-06	8.2	0.0	0.0	0.0	34.7	44.6	71.2	0.7	3.4	19.0
6.48	3.72	(±3.47E-06)	(±5.4)	(±0.0)	(±0.0)	(±0.0)	(±43.6)	(±18.7)	(±28.4)	(±0.3)	(±1.5)	(±0.0)
6.48	1.12	1.57E-06	13.4	50.1	30.5	33.1	65.9	62.7	0.0	0.0	0.0	19.0
7.60	1.12	(±8.60E-07)	(±1.7)	(±6.1)	(±2.8)	(±0.7)	(±21.5)	(±7.7)	(±0.0)	(±0.0)	(±0.0)	(±0.0)
7.60	6.46	1.92E-08	5.9	0.0	0.0	0.0	5.9	38.6	60.7	0.4	1.6	19.0
14.06	6.46	(±7.24E-08)	(±1.5)	(±0.0)	(±0.0)	(±0.0)	(±6.9)	(±8.8)	(±24.2)	(±0.1)	(±0.7)	(±0.0)

Depth values presented in this table are measured from free ground surface

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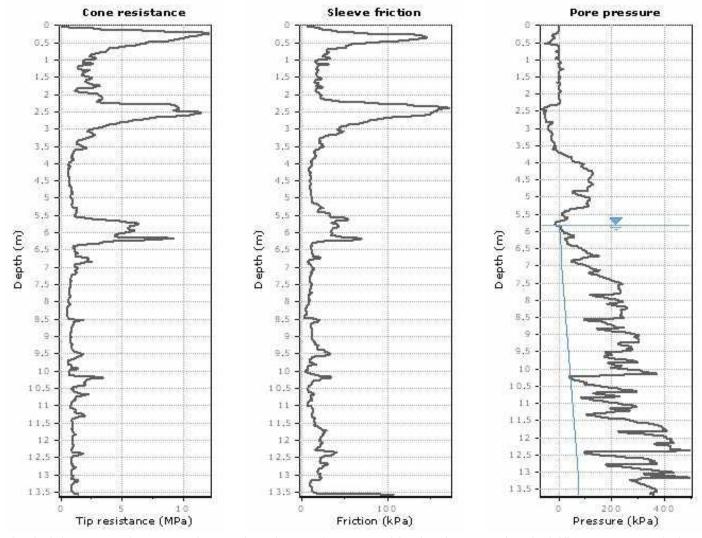
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CPT: KP CPT06

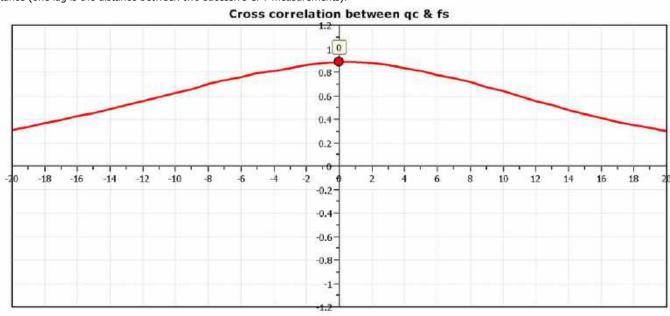
Total depth: 13.64 m, Date: 4/09/2020 Coords: lat 305830.3° lon 6879972.3°

Cone Type: 180911





The plot below presents the cross correlation coeficient between the raw qc and fs values (as measured on the field). X axes presents the lag distance (one lag is the distance between two sucessive CPT measurements).



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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

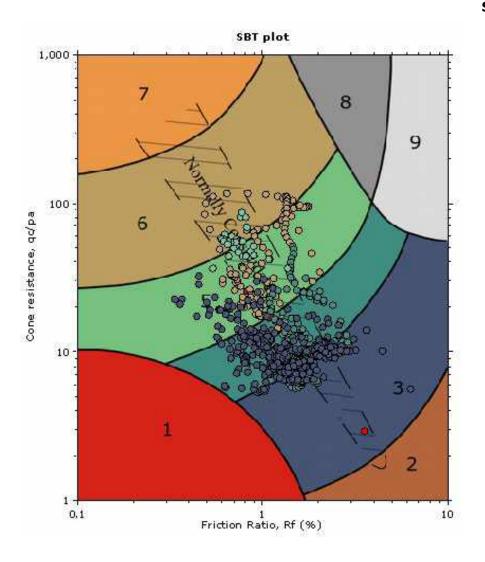
CPT: KP CPT06

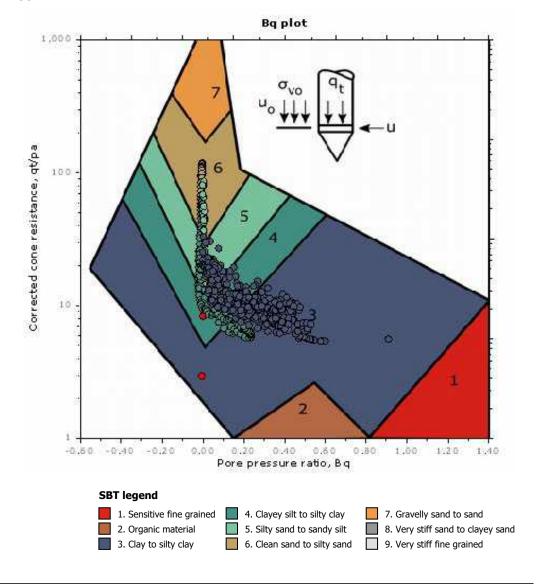
Total depth: 13.64 m, Date: 4/09/2020 Coords: lat 305830.3° lon 6879972.3°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

SBT - Bq plots





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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

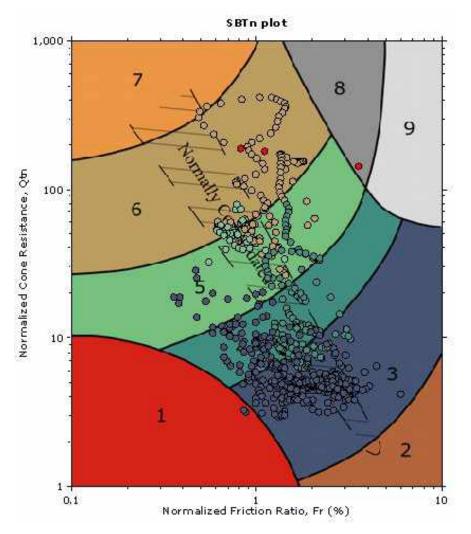
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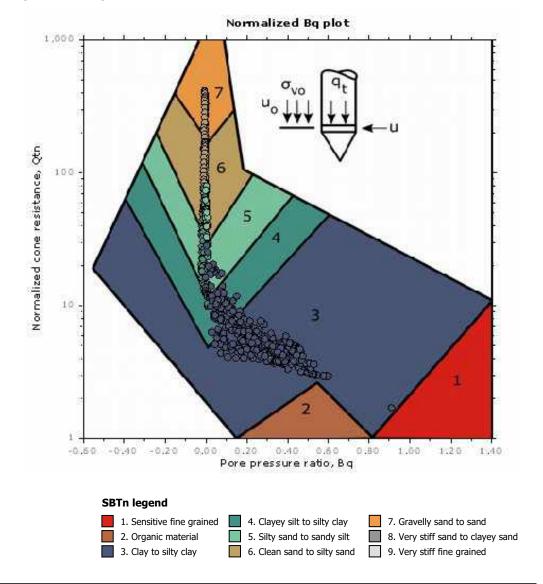
Total depth: 13.64 m, Date: 4/09/2020 Coords: lat 305830.3° lon 6879972.3°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

SBT - Bq plots (normalized)







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Perth WA 6004, Australia http://www.knightpiesold.com/

Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

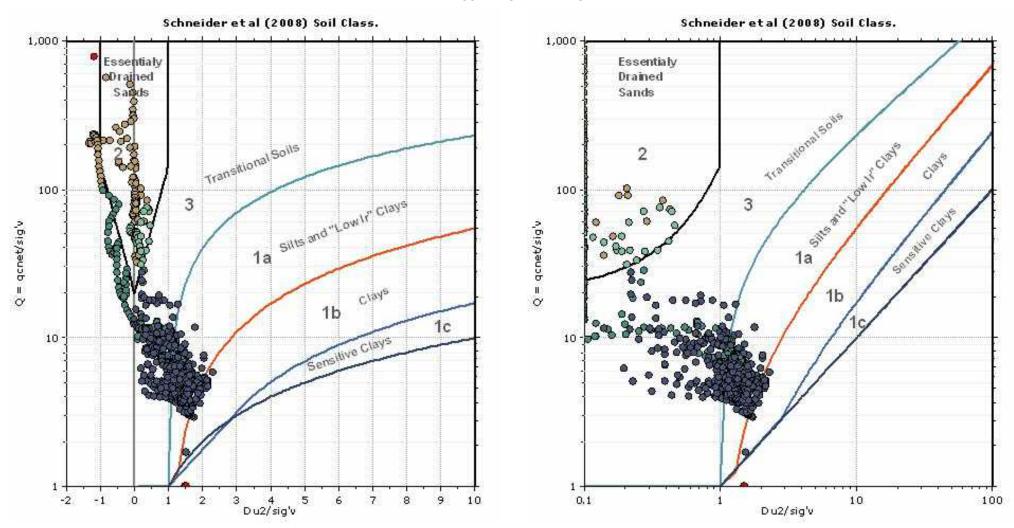
CPT: KP CPT06

Total depth: 13.64 m, Date: 4/09/2020 Coords: lat 305830.3° lon 6879972.3°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

Bq plots (Schneider)







http://www.knightpiesold.com/

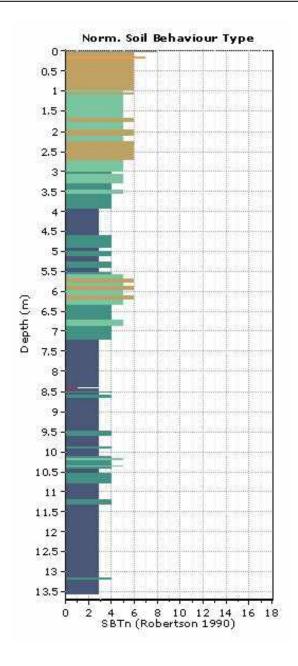
Thunderbox Gold Mine Project:

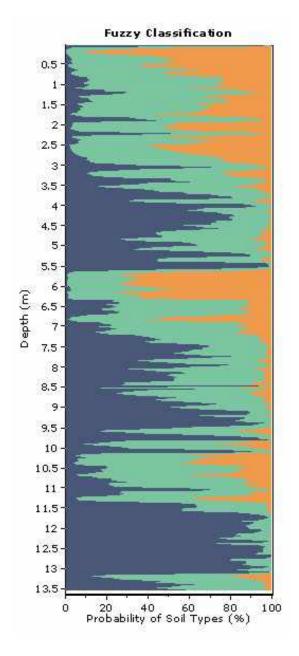
Location: Leinster, Western Australia, Australia

Total depth: 13.64 m, Date: 4/09/2020

CPT: KP CPT06

Coords: lat 305830.3° lon 6879972.3° Cone Type: 180911





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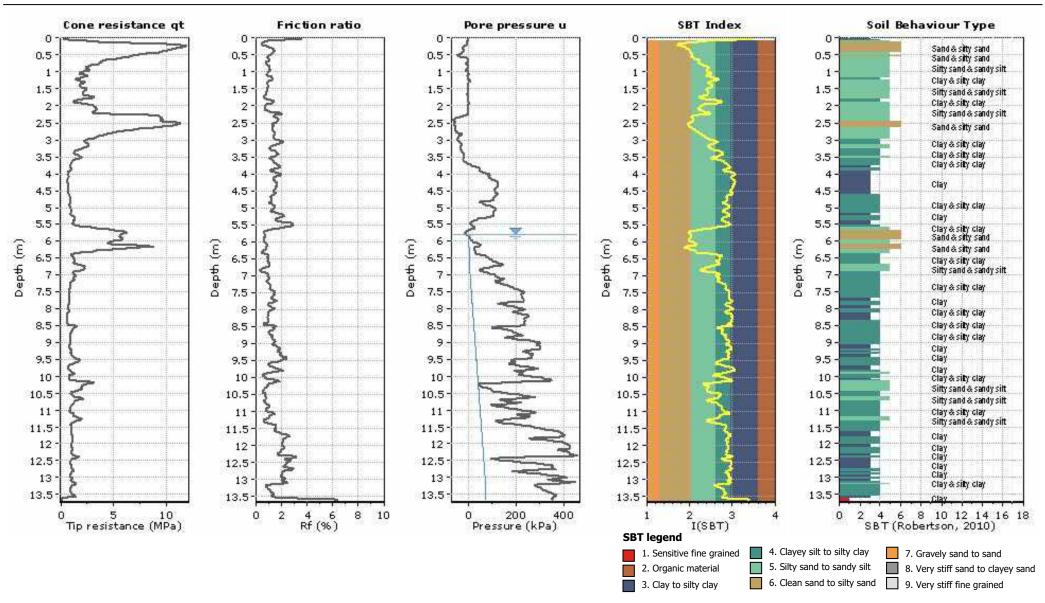
Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT06

Total depth: 13.64 m, Date: 4/09/2020 Coords: lat 305830.3° lon 6879972.3°

Cone Type: 180911



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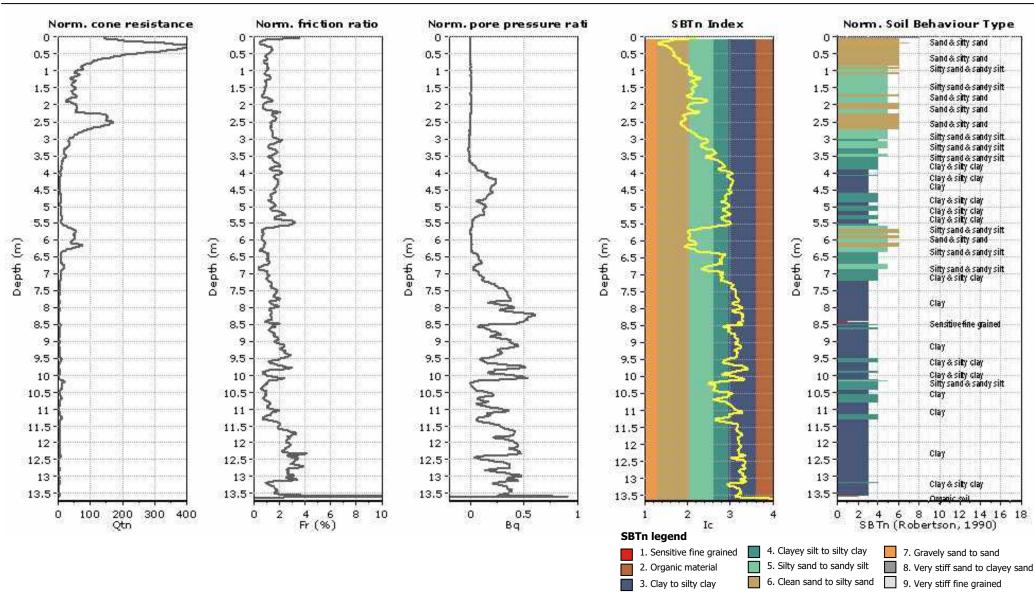
Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT06

Total depth: 13.64 m, Date: 4/09/2020 Coords: lat 305830.3° lon 6879972.3°

Cone Type: 180911



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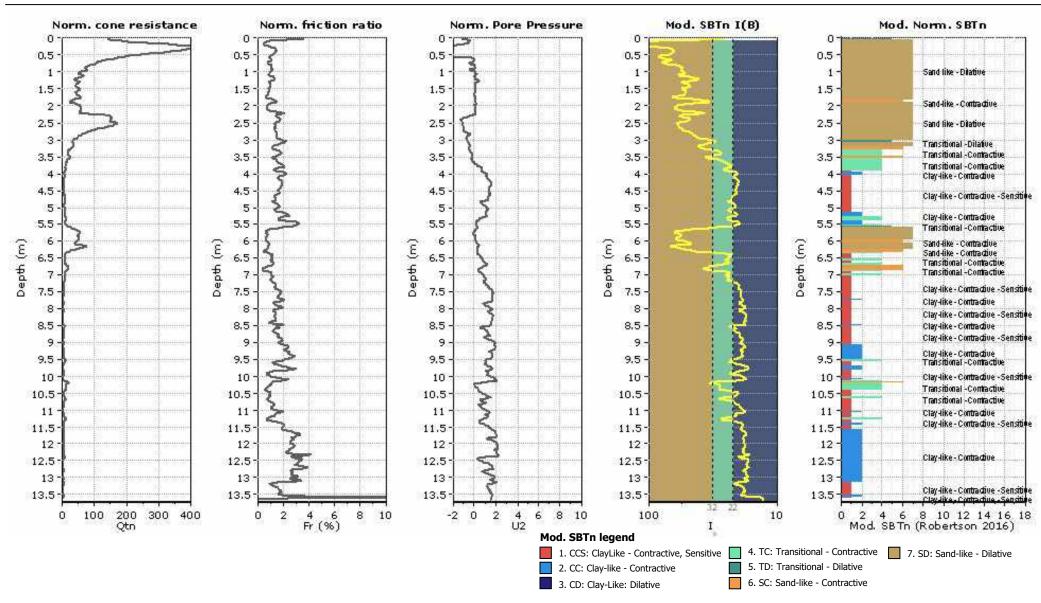
Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT06

Total depth: 13.64 m, Date: 4/09/2020 Coords: lat 305830.3° lon 6879972.3°

Cone Type: 180911



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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

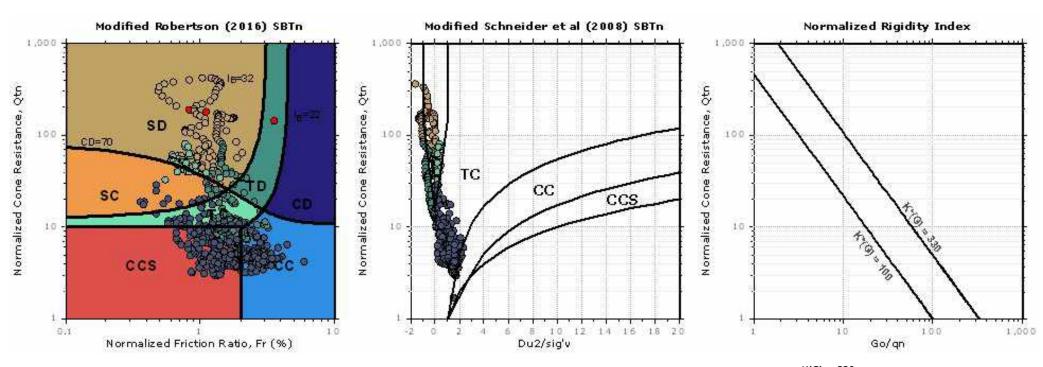
CPT: KP CPT06

Total depth: 13.64 m, Date: 4/09/2020 Coords: lat 305830.3° lon 6879972.3°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

Updated SBTn plots



CCS: Clay-like - Contractive - Sensitive

CC: Clay-like - Contractive

CD: Clay-like - Dilative

TC: Transitional - Contractive

TD: Transitional - Dilative

SC: Sand-like - Contractive

SD: Sand-like - Dilative

K(G) > 330: Soils with significant microstructure (e.g. age/cementation)

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CPT: KP CPT06

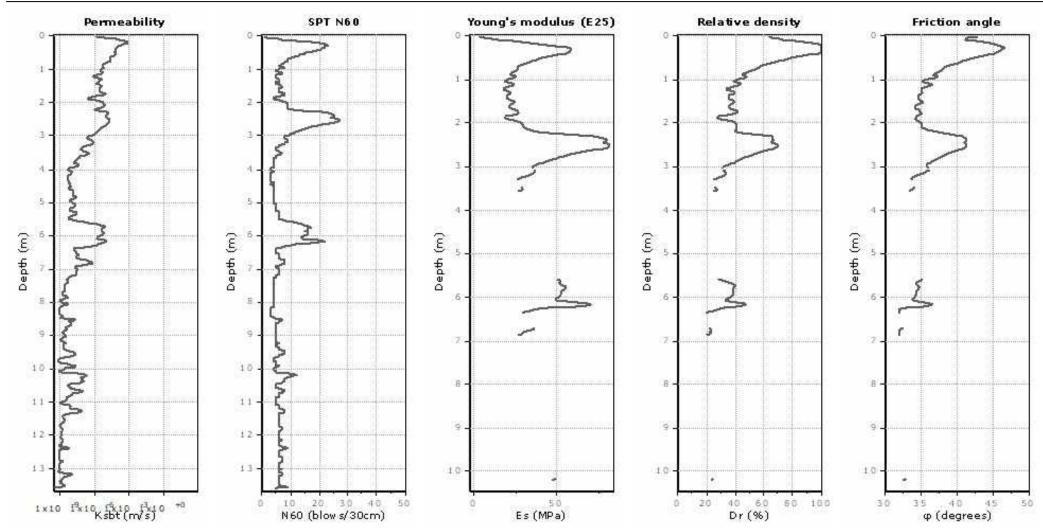
Total depth: 13.64 m, Date: 4/09/2020 Coords: lat 305830.3° lon 6879972.3°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia



Calculation parameters

Permeability: Based on SBT_n SPT N_{60} : Based on I_c and q_t

Young's modulus: Based on variable alpha using I_c (Robertson, 2009)

Relative density constant, C_{Dr}: 350.0 Phi: Based on Kulhawy & Mayne (1990) — User defined estimation data

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Project: Thunderbox Gold Mine

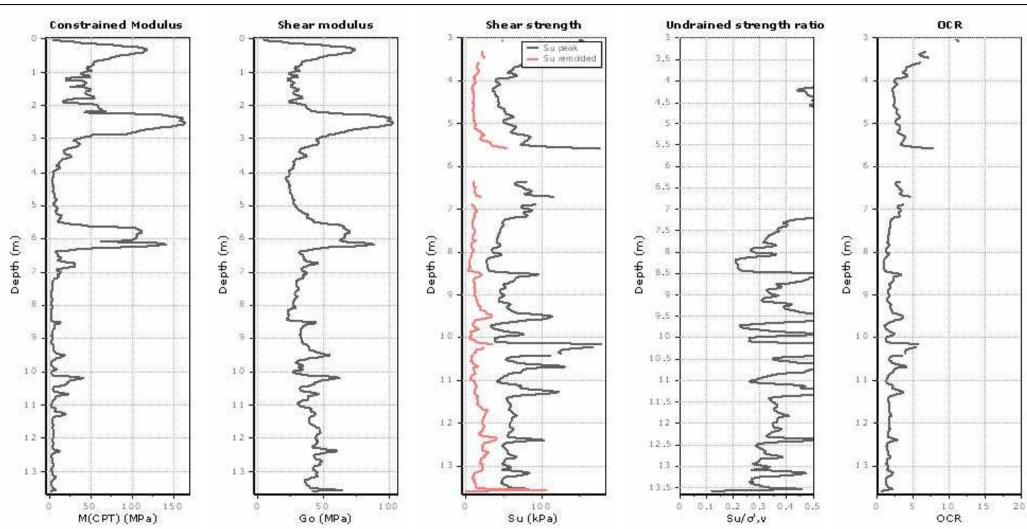
Location: Leinster, Western Australia, Australia

CPT: KP CPT06

Total depth: 13.64 m, Date: 4/09/2020 Coords: lat 305830.3° lon 6879972.3°

Cone Type: 180911

Cone Operator: Hagstrom Drilling



Calculation parameters

Constrained modulus: Based on variable alpha using $\,I_c$ and $\,Q_{tn}$ (Robertson, 2009) Go: Based on variable alpha using $\,I_c$ (Robertson, 2009) Undrained shear strength cone factor for clays, $\,N_{tt}$: 14

OCR factor for clays, N_{kt}: 0.33

User defined estimation dataFlat Dilatometer Test data

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Project: Thunderbox Gold Mine

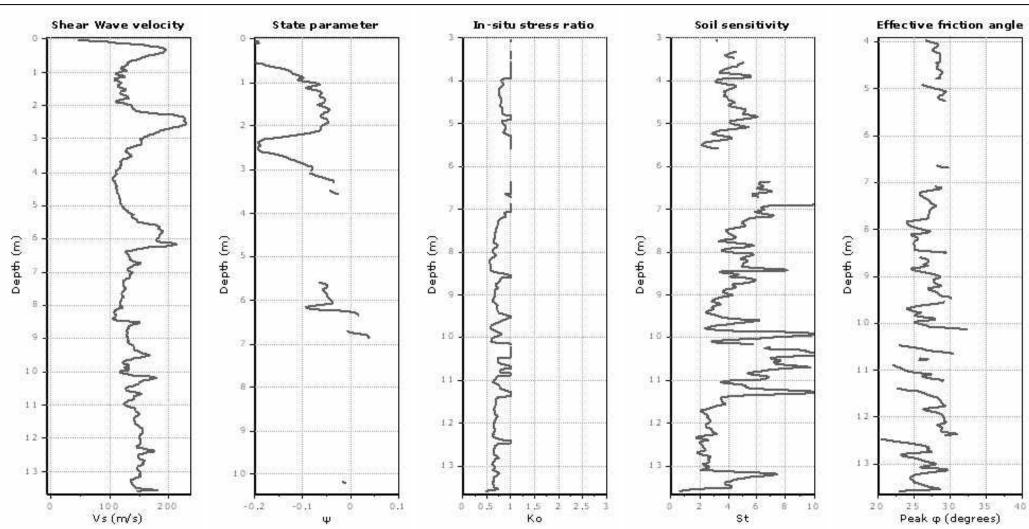
Location: Leinster, Western Australia, Australia

CPT: KP CPT06

Total depth: 13.64 m, Date: 4/09/2020 Coords: lat 305830.3° lon 6879972.3°

Cone Type: 180911

Cone Operator: Hagstrom Drilling



Calculation parameters

Soil Sensitivity factor, N_S: 7.00

User defined estimation data



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Thunderbox Gold Mine Project:

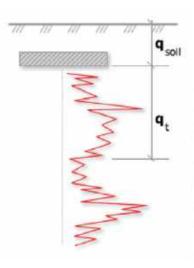
Location: Leinster, Western Australia, Australia

Total depth: 13.64 m, Date: 4/09/2020 Coords: lat 305830.3° lon 6879972.3°

Cone Type: 180911

CPT: KP CPT06

Cone Operator: Hagstrom Drilling

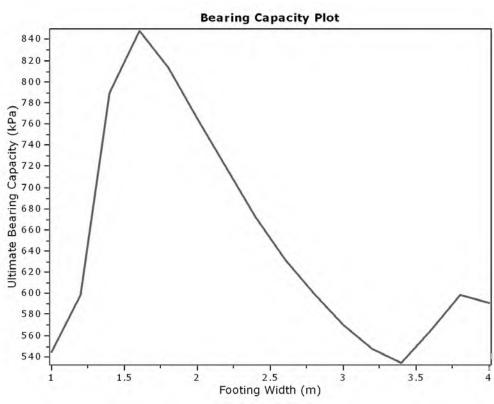


Bearing Capacity calculation is perfromed based on the formula:

$$Q_{ult} = R_k \times q_t + q_{soil}$$

where:

R_k: Bearing capacity factor qt: Average corrected cone resistance over calculation depth q_{soil}: Pressure applied by soil above footing



:: Tabula	ar results ::						
No	B (m)	Start Depth (m)	End Depth (m)	Ave. q _t (MPa)	R _k	Soil Press. (kPa)	Ult. bearing cap. (kPa)
1	1.00	0.50	2.00	2.68	0.20	9.50	544.56
2	1.20	0.50	2.30	2.94	0.20	9.50	597.96
3	1.40	0.50	2.60	3.90	0.20	9.50	789.11
4	1.60	0.50	2.90	4.19	0.20	9.50	848.01
5	1.80	0.50	3.20	4.02	0.20	9.50	812.70
6	2.00	0.50	3.50	3.78	0.20	9.50	764.95
7	2.20	0.50	3.80	3.54	0.20	9.50	718.12
8	2.40	0.50	4.10	3.31	0.20	9.50	672.46
9	2.60	0.50	4.40	3.11	0.20	9.50	631.32
10	2.80	0.50	4.70	2.95	0.20	9.50	599.47
11	3.00	0.50	5.00	2.80	0.20	9.50	569.78
12	3.20	0.50	5.30	2.69	0.20	9.50	547.76
13	3.40	0.50	5.60	2.62	0.20	9.50	534.08
14	3.60	0.50	5.90	2.78	0.20	9.50	565.08
15	3.80	0.50	6.20	2.94	0.20	9.50	598.43
16	4.00	0.50	6.50	2.91	0.20	9.50	590.98



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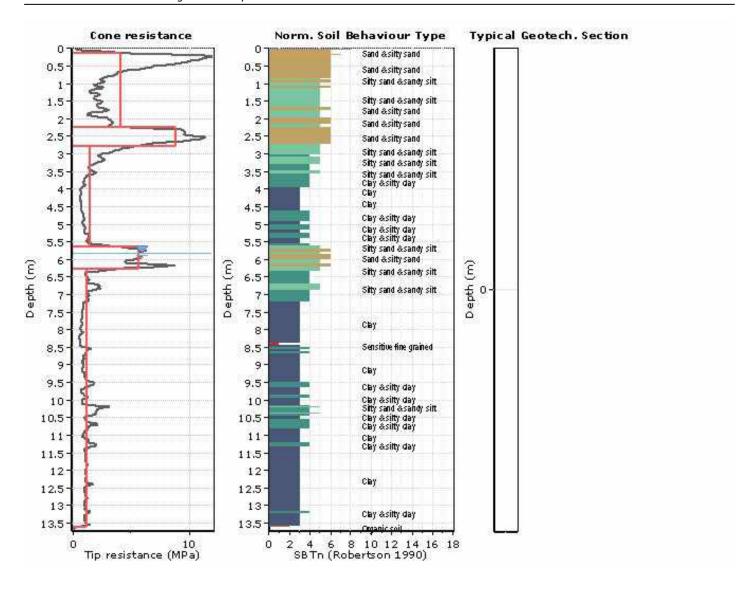
Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT06

Total depth: 13.64 m, Date: 4/09/2020 Coords: lat 305830.3° lon 6879972.3°

Cone Type: 180911



Tabular results

.:: Layer No: 1 ::.

Code: Layer_1 Start depth: 0.10 (m), End depth: 2.24 (m)

Description: Sand & silty sand

Basic results

Total cone resistance: 3.96 ±2.90 MPa Sleeve friction: 40.00 ±35.19 kPa

Ic: 1.97 ±0.26 SBT_n: 6

SBTn description: Sand & silty sand

Estimation results

Permeability: 6.08E-05 ±1.59E-04 m/s Constrain

 N_{60} : 9.57 ±5.15 blows Es: 29.62 ±11.32 MPa Dr (%): 52.85 ±22.39 ϕ (degrees): 37.75 ±3.93 °

Unit weight: $19.00 \pm 0.00 \text{ kN/m}^3$

Constrained Mod.: 56.00 ±25.13 MPa

Go: 37.12 ± 14.19 MPa Su: 0.00 ± 0.00 kPa Su ratio: 0.00 ± 0.00 O.C.R.: 0.00 ± 0.00 .:: Layer No: 2 ::.

Code: Layer_2 **Start depth:** 2.24 (m), **End depth:** 2.76 (m)

Description: Sand & silty sand

Basic results

Estimation results

Total cone resistance: 8.81 ±1.84 MPa

Sleeve friction: 130.19 ±30.21 kPa Ic: 1.93 ±0.07

SBT_n: 6 SBTn description: Sand & silty sand

Permeability: 1.32E-05 ±5.79E-06 m/s

 N_{60} : 22.56 ±3.86 blows Es: 71.80 ±10.02 MPa Dr (%): 62.26 ±6.61 φ (degrees): 40.24 ±1.13 ° Unit weight: 19.00 ±0.00 kN/m³

Su: $0.00 \pm 0.00 \text{ kPa}$ Su ratio: 0.00 ±0.00

Go: 89.98 ±12.56 MPa

Constrained Mod.: 143.59 ±20.04 MPa

CPT name: KP CPT06

O.C.R.: 0.00 ±0.00

.:: Layer No: 3 ::.

Code: Layer_3 **Start depth:** 2.76 (m), **End depth:** 5.60 (m)

Description: Clay & silty clay

Basic results

Estimation results

Total cone resistance: 1.40 ±1.00 MPa Sleeve friction: 22.00 ±15.71 kPa

Ic: 2.76 ± 0.26 SBT_n: 4

SBTn description: Clay & silty clay

Permeability: 2.69E-07 ±7.00E-07 m/s Constrained Mod.: 17.48 ±20.70 MPa

 N_{60} : 5.62 ±2.72 blows Go: 33.40 ±10.89 MPa Es: 0.00 ±0.00 MPa Su: 63.25 ±25.06 kPa Su ratio: 0.76 ±0.35 Dr (%): 0.00 ±0.00 φ (degrees): 0.00 ±0.00 ° O.C.R.: 3.49 ±1.61

Unit weight: 19.00 ±0.00 kN/m³

.:: Layer No: 4 ::.

Code: Layer_4 **Start depth:** 5.60 (m), **End depth:** 6.26 (m)

Description: Silty sand & sandy silt

Basic results

Estimation results

Total cone resistance: 5.52 ±1.21 MPa

Sleeve friction: 42.09 ±11.09 kPa Ic: 2.10 ± 0.12

SBT_n: 5

SBTn description: Silty sand & sandy silt

Permeability: 4.68E-06 ±2.76E-06 m/s

 N_{60} : 15.74 ±2.38 blows

Es: 54.43 ±6.05 MPa Dr (%): 37.57 ±4.15 φ (degrees): 34.54 ±0.78 °

Unit weight: 19.00 ±0.00 kN/m³

Constrained Mod.: 102.91 ±22.22 MPa Go: 68.22 ±7.59 MPa

Su: 0.00 ±0.00 kPa Su ratio: 0.00 ±0.00

O.C.R.: 0.00 ±0.00

.:: Layer No: 5 ::.

Code: Layer_5

Start depth: 6.26 (m), **End depth:** 13.58 (m)

Description: Clay

Basic results

SBT_n: 3

Estimation results

Total cone resistance: 1.12 ±0.45 MPa

Sleeve friction: 14.98 ±8.57 kPa Ic: 3.06 ±0.24

SBTn description: Clay

Permeability: 3.61E-08 ±1.67E-07 m/s

 N_{60} : 5.76 ±1.44 blows

Es: 0.00 ±0.00 MPa Dr (%): 0.00 ±0.00 ϕ (degrees): 0.00 ±0.00 ° Unit weight: $19.00 \pm 0.00 \text{ kN/m}^3$ Constrained Mod.: 7.08 ±8.15 MPa

Go: 37.67 ±7.85 MPa Su: 62.01 ±23.50 kPa Su ratio: 0.42 ±0.17 O.C.R.: 1.96 ±0.80



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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT06

Total depth: 13.64 m, Date: 4/09/2020 Coords: lat 305830.3° lon 6879972.3°

Cone Type: 180911

Cone Operator: Hagstrom Drilling

Summary table of mean values

From depth To depth (m)	Thickness (m)	Permeability (m/s)	SPT _{N60} (blows/30am)	E _s (MPa)	D _r (%)	Friction angle	Constrained modulus, M (MPa)	Shear modulus, G _o (MPa)	Undrained strength, S _u (kPa)	Undrained strength ratio	OCR	Unit weight (kN/m³)
0.10	2.14	6.08E-05	9.6	29.6	52.8	37.7	56.0	37.1	0.0	0.0	0.0	19.0
2.24	2.17	(±1.59E-04)	(±5.1)	(±11.3)	(±22.4)	(±3.9)	(±25.1)	(±14.2)	(±0.0)	(±0.0)	(±0.0)	(±0.0)
2.24	0.52	1.32E-05	22.6	71.8	62.3	40.2	143.6	90.0	0.0	0.0	0.0	19.0
2.76		(±5.79E-06)	(±3.9)	(±10.0)	(±6.6)	(±1.1)	(±20.0)	(±12.6)	(±0.0)	(±0.0)	(±0.0)	(±0.0)
2.76	2.84	2.69E-07	5.6	0.0	0.0	0.0	17.5	33.4	63.2	0.8	3.5	19.0
5.60	2.04	(±7.00E-07)	(±2.7)	(±0.0)	(±0.0)	(±0.0)	(±20.7)	(±10.9)	(±25.1)	(±0.3)	(±1.6)	(±0.0)
5.60	0.66	4.68E-06	15.7	54.4	37.6	34.5	102.9	68.2	0.0	0.0	0.0	19.0
6.26	0.00	(±2.76E-06)	(±2.4)	(±6.1)	(±4.2)	(±0.8)	(±22.2)	(±7.6)	(±0.0)	(±0.0)	(±0.0)	(±0.0)
6.26	7.32	3.61E-08	5.8	0.0	0.0	0.0	7.1	37.7	62.0	0.4	2.0	19.0
13.58	7.52	(±1.67E-07)	(±1.4)	(±0.0)	(±0.0)	(±0.0)	(±8.2)	(±7.9)	(±23.5)	(±0.2)	(±0.8)	(±0.0)

Depth values presented in this table are measured from free ground surface

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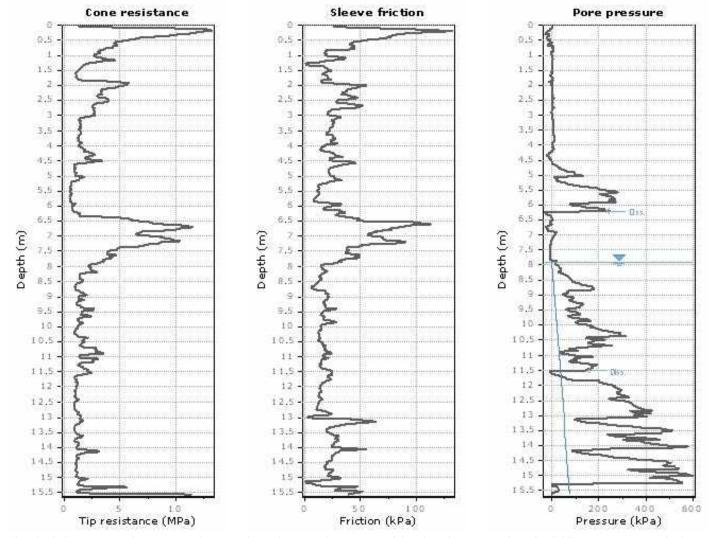
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Project: Thunderbox Gold Mine

Location: Leinster, Western Australia, Australia

CPT: KP CPT07A Total depth: 15.64 m, Date: 5/09/2020

Coords: lat 305898.1° lon 6879692.3° Cone Type: 180911



The plot below presents the cross correlation coeficient between the raw qc and fs values (as measured on the field). X axes presents the lag distance (one lag is the distance between two sucessive CPT measurements).

