

# **APPENDIX J**

Geotechnical Assessment Reports for Pit and WRS



# **Pells Sullivan Meynink**

**Engineering Consultants Rock-Soil-Water**

> G3 56 Delhi Road North Ryde NSW 2113 P: 61-2 9812 5000 F: 61-2 9812 5001 mailbox@psm.com.au www.psm.com.au

Our ref: PSM71-224R

16 February 2018

OceanaGold Corporation Golden Point Road RD3, Macraes Flat 9483 East Otago NEW ZEALAND

ATTENTION: LOUIE HERRERO

Dear Louie

## **RE: GEOTECHNICAL REVIEW OF DEEPDELL PIT STAGE 3 DESIGN**

We are pleased to submit our report providing a geotechnical review of Deepdell pit design for operations and closure.

We trust it is in keeping with your requirements, but should you have any queries please do not hesitate to contact us.

For and on behalf of PELLS SULLIVAN MEYNINK

ROBERT BERTUZZI

Distribution: pdf copy emailed to [Louie.Herrero@oceanagold.com](mailto:Louie.Herrero@oceanagold.com) Original held by PSM

# **OceanaGold**

# **GEOTECHNICAL REVIEW DEEPDELL STABILITY ASSESSMENT**

**PSM71-224R FEBRUARY 2018**



## **EXECUTIVE SUMMARY**

This report presents a geotechnical stability assessment carried out by PSM to accompany a resource consent application for the proposed Deepdell Stage 3 pit development. This study was requested by OceanaGold Corporation to confirm slope performance during the following phases.

- Operational conditions during excavation This assumes the water table is drawn down below the ultimate pit floor to approximately 372 mRL.
- Closure planning Scenario  $1 A$  pit lake forms to the lowest intersection point with the topography at the southern end of the pit, approximately 445 mRL.
- Closure planning Scenario  $2 A$  pit lake fills to the lowest intersection point at the northern end of the pit, approximately 475 mRL.

The typical slope design for Deepdell Stage 3 comprises:

- North and east walls comprising 15 m high, 60° batters and 4.3 to 7.5 m wide berms producing and 90 to 125 m high, 40 to 48° inter-ramp slopes.
- West wall (orientated towards 090°) comprising variable batter and berm configurations, producing a 145 m high, 23 to 38° inter-ramp slopes.

Geotechnical stability assessments undertaken have shown to have a minimum factor of safety  $\geq$  1.5. This exceeds the general acceptance level for mine slope stability. A sensitivity analysis was carried out on the east wall to test the design for seismic loading. A horizontal load of 0.13g (equivalent to a M 4.5 earthquake 0 km from the mine) was applied in the analysis. The resultant factor of safety  $= 1.2$  indicating that the design would remain stable under the earthquake.

The proposed waste rock stacks to the north and south of the pit will have no effect on stability.



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## <span id="page-6-0"></span>**1 INTRODUCTION**

This report presents a geotechnical stability assessment carried out by PSM to accompany a resource consent application for the proposed Deepdell Stage 3 pit development. This study was requested by Mr Pieter Doelman ([1\)](#page-6-1) of OceanaGold Corporation (OceanaGold) to confirm the stability of the proposed slope design during mining and at closure where an in pit lake is to be established.

OceanaGold has been mining at Macraes Flat since 1990 targeting mineralisation within the Hyde-Macraes Shear Zone (HMSZ), Figure 1. Deepdell is one of eleven open pits within OceanaGold's operations and is located in the northern part of the operation between the Coronation and Golden Point pits.



**Figure 1: Schematic view of the open pit which develops along strike of the Hyde-Macraes Shear Zone**

<span id="page-6-1"></span><sup>&</sup>lt;sup>(1)</sup> Email from Pieter Doelman, dated 13 December 2017



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Deepdell has been mined in three phases prior to the current development, Figure 2. This includes smaller Stage 1 and 2 satellite excavations plus the more prominent Deepdell south pit (located on the opposite side of Deepdell Creek). Currently the Stage 1 and 2 pits are back filled with a waste dump overlying the excavations.

The following aspects will be covered to provide an understanding of the geological environment and geotechnical analysis included as part of this study:

- Data sources,
- Geotechnical Model,
- Kinematic and Statistical Stability Analysis,
- Limit Equilibrium Stability Analysis,
- Slope Design Conformance, and
- Recommendations.

All figures are included at the end of the text. Some figures are reproduced within the text, where they assist in the narrative.

#### <span id="page-7-0"></span>**2 SCOPE**

Oceanagold requested PSM conduct a geotechnical stability assessment of the Deepdell Stage 3 pit development, Figure 3, to confirm slope performance during the following phases.

- 1. Operational conditions during excavation This assumes the water table is drawn down below the ultimate pit floor to approximately 372 mRL.
- 2. Closure planning Scenario  $1 A$  pit lake forms to the lowest intersection point with the topography at the southern end of the pit, approximately 445 mRL.
- 3. Closure planning Scenario 2 A pit lake fills to the lowest intersection point at the northern end of the pit, approximately 475 mRL.

To achieve the design water level and provide adequate freeboard for Scenario 2 an inpit embankment will be constructed in the southern end of the pit. This option provides redundancy if OceanaGold cannot ensure the lake water quality is suitable for discharge, or if the erosion risk of the discharge on the south side cannot be managed due to steep topography.

## <span id="page-7-1"></span>**3 AVAILABLE DATA**

The following data sources have been made available for this review.

#### **Pit shell and topographic wireframes**

• Deepdell Stage 3 pit,



- As built topographic surface Current backfilled topography,
- As mined topographic surface Mining extent during Stage 1 and 2, and
- Waste rock stack surfaces (Proposed) North and South of Deepdell.

## **Geology**

PSM has a long standing relationship with OceanaGold having contributed to the Macraes operations for more than twenty years. The geological understanding is well established with numerous publications detailing the origin and genesis of the area's geology.

## **Geological surfaces**

- Three dimensional fault wireframe of the site persistent "Footwall Fault".
- Three dimensional fault wireframes of inferred geological structures within the northern wall. "DD\_Fault\_plane\_1–3.dxf".

## **Maps and Plans**

- Two dimensional surface traces (.dxf) of the Horse Flat, Table Top and Cyanide Creek Faults.
- Failure outlines and observed surface cracking (.dxf) Deepdell North: DD02, DD03, DD04, DD05, DD06.

## <span id="page-8-0"></span>**3.1 Geotechnical Investigations**

PSM collected geotechnical data during the development of Stage 1, 2 and Deepdell South. This information includes:

## **Trench Excavations**

Exploratory trenches were excavated to establish an early understanding of the geological conditions at Deepdell prior to establishment of the initial workings. A trench was orientated parallel to the east wall of the Stage 1 pit with three additional trenches excavated normal to this. Each trench was geotechnically mapped and data recorded in accordance with OceanaGold logging standards. Mapped defects are presented on stereoplots to provide an overview of structural orientation in the area. Summary histograms capture key defect characteristics including length, spacing, surface roughness infill and water. All of the above data is included in Appendix A.

## **Borehole data**

Seven diamond cored boreholes located within the Stage 1 pit. Geotechnical parameters logged include weathering, infill type and thickness. Two holes (DDH39 and 42) include fractures per metre.

Summary borehole logs and histograms of defect characteristics for the available data is included in Appendix B.



Borehole and trench locations are illustrated spatially in Figure 2. No new geotechnical investigations have been undertaken as part of this study.

## <span id="page-9-0"></span>**4 GEOTECHNICAL MODEL**

### <span id="page-9-1"></span>**4.1 Geology**

The Macraes Flat area is within the extensively deformed and moderately metamorphosed Otago-Haast Schist Belt. The schist comprises a sequence of metamorphosed Otago-Haast Schist Belt. gradational psammitic and pelitic lithologies derived by metamorphism of Mesozoic aged sandstone and mudstone. The rocks are strongly foliated and depending on the origins, are either light grey, quartz rich and laminated (psammite) or dark grey to green, micaceous and finely laminated (< 5mm thick) (pelite).

Mineralisation occurs within the north-south  $(2)$  $(2)$  $(2)$  trending HMSZ (the Shear Zone) has a strike length of at least 35 km, Figure 1. The Shear Zone thickness varies from 5 to 140 m and is defined by the relatively continuous Hanging Wall Shear (HWS) and Footwall Fault (FF) which also approximate its lithological boundaries. Its tectonic displacement has been inferred to be hundreds of metres. The strain associated with that tectonic displacement was probably concentrated within the intra-shear pelite which could more readily absorb strain than the coarsely grained psammite above and below the Shear Zone. The structural geology of the area is dominated by two main orthogonal fault sets, striking to the north and east.

The Shear Zone dips gently to the east from Stoneburn in the south to Coronation in the north but displays a broad bend at Nunns, turning to dip to the northeast, Figure 1. A schematic geological cross section illustrating the adjacent Golden Point development illustrated the typical assemblage throughout the deposit, Figure 4.





<span id="page-9-2"></span> $^{(2)}$  All directions quoted are relative to Macraes' mine grid which is 45 $^{\circ}$  west of true north. The magnetic declination at Macraes (45.38°S, 170.43°E) is approximately +25° [\(https://ww.ncdc.noaa.gov/news/national-centers-environmental-information](https://ww.ncdc.noaa.gov/news/national-centers-environmental-information) 15/3/16)



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The geological model for Deepdell has been refined from the initial trenching and borehole investigations through the mining of Stage 1 and 2. The dominant lithology expected within the Stage 3 excavation is intershear pelite. This unit makes up the majority of the rock mass between the HWS and FF beneath.

Further discussion regarding the pit scale geology is included below in Section 4.4.

## <span id="page-10-0"></span>**4.2 Geological Structure**

Geological structure defines defects within a rock mass resulting from the time of deposition/formation and the sites evolution though changes in stress condition/deformation over geological time. These structures provide an insight into the history of a geological environment and can be analysed to predict future performance of an excavated slope.

The geological structure model at Macraes typically comprises:

- Foliation, foliation shears and faults showing a broad range of dips from flat to moderately dipping towards the east (average 15 to 45°/100). This set parallels the FF and HWS of the HMSZ.
- Mine and regional scale faults that moderately to steeply dip towards the east ( $\approx 60^{\circ}/090^{\circ}$ ). These faults are often infilled with clay or breccia to 100mm thick. This fault set includes the Northern Gully Fault and East Fault.
- Batter and mine scale faults and shears showing a broad range of dips from flat to moderately dipping towards the west (average  $\approx$  50°). This fault set includes the Ramp Shears which are typically truncated by the easterly dipping faults, described in Section 4.
- Joints and faults that are moderately to steeply dipping towards the north and south ( $\approx$  55-75°/025°).

Structural data collection at Deepdell is limited to the initial exploratory trench mapping and discrete measurements surrounding individual zones of instability.

The available data was initially analysed to divide it into structural domains, i.e. areas of similar pattern. However, assessment of the stereoplots generated separately for trenches 1, 2 and 3 (Appendix A) indicated no significant spatial difference in structure. Therefore, the area of the proposed pit has been structurally analysed in terms of one domain only. A stereoplot summarising the trench mapping data is shown in Figure 5.





#### **Figure 5: Defect stereonet illustrating geological structure trends from trench mapping at Deepdell. Foliation (Green), Joints (Blue) and Gouge/Shear Zones (Red).**

The major structural trends specific to Deepdell are summarised below.

- Foliation dips at between 5° and 45° to the east/south east.
- Joints are typically sub vertical to steeply dipping to the south and west. Secondary sets dip steeply to sub vertical to the east and north.
- Three zones of gouge/crushed rock and shearing are apparent:
	- The first is flat to moderately dipping to the south to southeast, although a small number of defects are indicated as dipping to the southwest.
	- The second is moderately to steeply dipping to the southeast.
	- The third dips moderately to steeply to the north, but is poorly defined. These structures are favourably orientated with These structures are favourably orientated with respective to the pit walls as they dip approximately perpendicular to the major wall aspects.

## <span id="page-11-0"></span>**4.2.1 Defect Characteristics**

Defect characteristics including length, continuity, roughness and water condition were recorded during trench logging. Information on defect infill and infill thickness was recorded in the borehole logs for gouge, crushed zones and shear zones. Typical descriptions for the major defect types with the exception of foliation (experience indicates that foliation is typically not a controlling defect for pit wall stability) are summarised below.



- Joints: Visible length typically in the range  $0.5 1.5$  m and continuous out of exposure. Spacing between joint planes typically ranges from <0.25 m to 0.75 m. The joints are mostly undulose on the large scale, rough on the small scale and tight. Experience from other pits at Macraes tends to indicate that the typical length of the majority of joints is less than 5 m.
- Gouge/Sheared and Crushed Zones: Visible length typically in the range of 1.0 to 5 m, and continuous out of exposure. The defects are undulose on the large scale and smooth on the small scale. The defects have been logged as being predominantly moist to damp. Borehole logging indicates that the typical infill type is gouge/rock fragments. Infill thicknesses are variable; however appear to be typically in the ranges of 20 to 50 mm and 200 to 500 mm.

## <span id="page-12-0"></span>**4.2.2 Major Geological Structures**

Major geological structures (first order) at Macraes typically refers to the FF and HWS as these features are known to significantly impact slope stability when undercut. The FF maintains a relatively consistent orientation throughout the operation, with a dip and dip direction in the Deepdell area of 12°/075°. This indicates that the fault is located approximately 92 m beneath the Stage 3 pit shell with a relatively shallow dip toward the east. A cross section illustrating this geometry is included in Figure 6.

Geological studies carried out between 1996 and 1999 identified the Deepdell Creek landforms as indicative of an ancient landslide with the FF as its basal plane. This suggested that movement around the Deepdell Creek area had occurred along the FF prior to any mining.

Reviewing second order geological structures (pit scale compared to regional structures) there are a number of features with the potential to negatively affect pit wall stability. This occurs when a feature is unfavourably orientated relative to a specific pit wall aspect and the structure dips out of the slope at an angle shallower than the design wall angle.

Surface traces of second order structures were provided by Oceanagold as mapped during development of the Stage 1 and 2 pits. This information has been combined with defect orientation data by PSM to generate three dimensional fault projections through the as built topography. A detailed explanation of this process is included for reference in Appendix C.

The resultant model is illustrated in Figure 7. Features of note include:

- The intersection of the Horse Flat Fault and inferred Fault Plane "2" within the northern wall. This is an unfavourable geometry as it isolates and potentially decouples a rock block dipping out of the slope.
- The orientation of the Cyanide Creek Fault relative to the existing east wall of the previous development. This fault strikes toward the south east direction which is oblique to the typical north south orientation of other major/first order structures in the area.
- The Table Top Fault is considered to be a benign feature due to its orientation striking towards the east wall of the Stage 3 design. There is potential for an area of structural complexity where this structure



converges with the Cyanide Creek Fault in the lower west wall of the proposed development.



Stability implications relating to these structures are discussed further in Section 8.

**Figure 7: Plan view of Deepdell as built Stage 1 and 2 topography (Grey) superimposed with the proposed Deepdell Stage 3 pit shell (Green). Second order fault structure projections included as noted. Historic instabilities outlined in red appear to be associated with the Cyanide Creek Fault.**



## **4.3 Rock Mass Conditions**

- <span id="page-14-0"></span>The pelitic schist is typically weathered to a depth of 12 to 15 m. This corresponds to the majority of the proposed pit walls being formed in fresh rock. The degree of weathering is noted to vary from highly to slightly weathered.
- Typical fracture spacing is about two defects per metre, ranging between one and seven. Fracture spacing is used to estimate block sized during stability analysis and give an indication to how intact the rock mass may be when excavated.
- Rock mass strength was not recorded as part of the borehole logging. However, based on the trench mapping the weathered pelitic schist has a typical field estimated strength of R3 to R4. This corresponds to an unconfined compressive strength (UCS) of about 25 MPa.
- The strength of fresh pelitic schist is typically higher than that of weathered schist; hence it is interpreted that the fresh rock should typically be at least R4 to R5 (typical UCS greater than 50 MPa). This value is greater than that logged in adjacent boreholes south of Deepdell Creek.

## <span id="page-14-1"></span>**4.4 Groundwater**

The *in situ* schist at Macraes has a very low permeability. However, the numerous faults, shears, shear zones and fractures throughout the rock mass in general provide paths for groundwater movement. The overall permeability therefore within the rock mass is not uniform but rather defined into compartments by the faults and shears.

No active groundwater monitoring piezometers are understood to be available within the Deepdell footprint. The Deepdell south pit has an in pit lake of approximately 370 mRL that can be used to assume a basic standing water level.

## <span id="page-14-2"></span>**5 HISTORIC SLOPE PERFORMANCE**

## <span id="page-14-3"></span>**5.1 Slope Design Parameters**

Recommended slope design parameters for Deepdell were previously derived for the Stage 1, 2 and Deepdell South pits, (References 2 & 3). These are summarised below.

- Deepdell 1 30m high, 75° batters with 10 m wide berms, and
- Deepdell South,
	- 15 m high,  $60^\circ$  batters with 7.5 m berms for the northern and eastern walls. This results in a 43° overall slope angle.
	- 70° batters and 5 m berms for the southern and western walls. This results in a 55° overall slope angle.
	- Batters in the weathered/oxide material should be 10 m high and 60° all around the pit.



# <span id="page-15-0"></span>**5.2 Slope Stability Precedent**

Initial designs for Deepdell aimed to achieve batter face angle of 75° degrees in fresh rock slopes along all slopes aspects. A number of instabilities occurred where persistent structure was exposed in the north and east walls.

Highlighted in Figure 7 are the locations of three multi bench wall instabilities recorded from the early stage Deepdell developments. The two east wall failures are orientated parallel to the Cyanide Creek fault. Exposure of sympathetic structures within the adjacent damaged rock mass are likely to be associated with these instabilities.



Photographs 1 and 2 illustrate the site condition at the time of failure.

**Photograph 1: View of failure head scarp looking south** 





**Photograph 2: View of east wall multi bench failure at Deepdell** 

Slope performance precedent from early developments in Stages 1 and 2 were used to educate a revised set of design angles for Deepdell South. These angles have been adopted for ongoing development of the proposed Deepdell Stage 3.

## <span id="page-16-0"></span>**6 PIT GEOMETRY**

## <span id="page-16-1"></span>**6.1 Changes from current as built design**

Deepdell Stage 3 cutbacks focus on development towards the east. The pit floor deepens by approximately 66 m from 438 mRL to 372 mRL, (Figure 3). The length of the north wall increases significantly but with a favourable slope aspect (180°), this does not undercut major structures of geotechnical concern. The east wall minimises slope aspects orientated between 255 and 295° by using a stepped profile (in plan) alternating between 270° and 320°. This has been implemented to avoid exposure of geological structures along known slope aspects that are more susceptible to cause slope instabilities.

The typical slope design for Deepdell comprises:

- North and east walls comprising 15 m high, 60° batters and 4.3 to 7.5 m wide berms producing and 90 to 125 m high, 40 to 48° inter-ramp slopes.
- West wall (orientated towards 090°) comprising variable batter and berm configurations, producing a 145 m high, 23 to 38° inter-ramp slopes.



# <span id="page-17-0"></span>**7 STABILITY ANALYSIS - GEOLOGICAL STRUCTURE**

Assessment of potential failure mechanisms through intersection of geological structures was carried out using kinematic and statistical methods. Kinematic analysis is a stereographic based technique that considers the mean orientation of defect sets, the slope orientation, the defect shear strength and potential failure modes. The analyses are simply an assessment of the geometrical possibility of simple block movements, i.e. planar sliding, wedge failure and toppling.

These methods can be applied to a range of slope scales depending on the nature of the local geology. However, for Deepdell this method is used to assess individual and multi bench stability due to the relative short persistence of local geological structures.

Statistical analyses extend the results of kinematic analyses and are utilised for two main reasons. Firstly, the scatter in defect orientation is often large which suggests that a traditional kinematic analysis which uses the mean orientation may yield either optimistic or conservative slope angles. Secondly, the sensitivity of defect dips may be qualitatively assessed. The risk of failure on a defect set  $(C_f)$  has commonly been defined as:

 $C_f = C_u \times C_l$ 

Where  $C_{\mu}$  is a measure that a slope is undercut by a defect. The apparent dip of either a defect (planar failure) or an intersection (wedge failure) is steeper than the friction angle and shallower than the batter.  $C_{u}$  is influenced by the size and quality of data set. It assumes the data set is representative of the structural domain.

 $C<sub>l</sub>$  is a measure of the length of the defect compared to the height of the slope. Because the statistical analyses are based on the results of trench mapping, it is not possible to filter out data less than the height of the slope, as in fact all data would be removed. Consequently,  $C_f = C_u$ .

Various published procedures as well as in-house techniques have been used to assess appropriate slope angles. Published data and experience suggests that a  $C_f = 10$  to 30 % generally provides appropriate bench slopes (slopes up to about 30m high) at a level of risk commensurate with open pit mining practice in hard rock.

Results of the kinematic analyses are summarised in Table 7 .1. The result sheets are presented in Appendix D.



## **TABLE 7.1 RESULTS OF KINEMATIC AND STATISTICAL ANALYSES**



The results indicate inter-ramp angles  $3$  of 48 to 50 $^{\circ}$  and batter angles of 54 to 78 $^{\circ}$ . Experience from the other pits and rock exposures at Macraes indicate that joint lengths are typically less than 5 m. As such, the suggested batter angles may be conservative.

Although the chosen slope aspects refer to the wall orientations from the initial mining stages, they are considered reliable for the proposed development of major slope aspects with the proposed Deepdell Stage 3 design.

A tabulated summary of the proposed slope design parameters and a qualitative risk assessment for respective walls is included in Appendix E.

## <span id="page-18-0"></span>**8 LIMIT EQUILIBRIUM ANALYSIS**

Limit equilibrium analysis using RocScience's *Slide* software was used to assess structural and rock mass failure mechanisms for inter ramp and overall wall stability within the proposed Deepdell Stage 3 pit.

Due to the strength of the unweathered rock mass and its relative intact nature (compared to having a closely spaced blocky rock mass), non-circular failure surfaces were analysed to establish the Factor of Safety (FoS) for different scenarios.

The FoS of a slope can be established using the following formula:

$$
FoS = \frac{Shear Strength of a material}{Shear stress required for equilibrium}
$$

The output result is therefore numerical, the following scheme of interpretation being widely accepted in geotechnical engineering. If the FoS was to result in a value ≤ 1 then

<span id="page-18-1"></span> $3$  Toe-to-angles.



it is not deemed to be in an equilibrium state. This would infer that the slope in question is facing imminent failure. Results that exceed 1 are interpreted as stable in terms of slope equilibrium. It should be emphasised that the outputs from such analysis are only as accurate as the input data utilised in the simulation. A number of guidelines have been developed to provide a "buffer" or level of assurance for any given result. In general, for mining applications an acceptable FoS is considered any value  $>1.2$   $^{4}$  $^{4}$  $^{4}$ . This may vary depending on the acceptable level of risk for a given scenario or the design life of the excavation.

The GLE Morgenstern Price method was selected for the analysis as it better suited to capture the potential failure mechanisms applicable to the observed ground conditions. For Deepdell this includes structural sliding along crushed rock/gouge infill associated with first and second order geological structures.

Analysis scenarios considered the following:

- Operational stability,
- Closure stability at 445 mRL (considering a lake filling the Pit),
- Closure stability at 475 mRL (considering a lake filling the Pit), and
- Seismic load applied to the section with the lowest FoS.

#### <span id="page-19-0"></span>**8.1 Material Properties**

This study has adopted established rock mass shear strength properties used in a suite of previous studies. The equivalent Mohr-Coulomb strength parameters are presented in Table 8.1.

#### **TABLE 8.1 SHEAR STRENGTH PARAMETERS**



<span id="page-19-1"></span><sup>4</sup> Read, J and Stacey P., (2009) Guidelines for open pit slope design. *Chapter 9 - Acceptance Criteria*



## <span id="page-20-0"></span>**8.2 Groundwater Conditions**

A key component of this stability review was to test the slopes sensitivity to changes in groundwater level. As discussed earlier, three scenarios have been tested as part of this review:

- 1. Operational conditions during excavation This assumes the water table is drawn down below the ultimate pit floor to approximately 372 mRL.
- 2. Closure planning Scenario  $1 A$  pit lake forms to the lowest intersection point with the topography at the southern end of the pit, approximately 445 mRL.
- 3. Closure planning Scenario 2 A pit lake fills to the lowest intersection point at the northern end of the pit, approximately 475 mRL.

Figure 8 illustrates these three scenarios in plan and provides an indication of the possible water extents.

#### <span id="page-20-1"></span>**8.3 Seismic Loading**

The potential effects of an earthquake on the stability of the Deepdell pit walls at Macraes has been considered. Figure 9 illustrates the mine location relative to the 2016 Kaikoura earthquake site as an example of a recent major seismic event. This clearly identifies the higher earthquake hazard expected at Kaikoura compared with Macraes. The lower seismic hazard at Macraes is the result of the comparatively small amount of relative plate motion accommodated in the east of the Otago Region as evidenced by the region's low seismicity rate.

A comprehensive assessment of the specific seismic hazard at the Macraes Mine was undertaken by the Institute of Geological and Nuclear Sciences (IGNS) (Litchfield et al, 2005). That report assessed the probabilistic seismic hazard of the mine based on the following data and estimated peak ground accelerations (PGA) for different return periods (Table 8.2). This included the following assumptions:

- Knowledge of 21 active faults within 100 km from the mine
- The Seismic hazard associated with the Alpine Fault
- Distributed seismicity sources zones (based on historical earthquake data)
- The mine is a 'rock site'.

#### **TABLE 8.2 MACRAES PGA FOR DIFFERENT RETURN PERIODS (LITCHFIELD ET AL., 2005)**





It is considered that the IGNS 2005 study remains applicable for assessing the seismic hazard of the Macraes as it consistent with the post-Christchurch events New Zealand Seismic Hazard Model (Stirling et al., 2012).

Hence, based on these two sources and Macraes tectonic setting, it is extremely unlikely that an earthquake such as the Kaikoura earthquake could occur close to the mine.



**Figure 9: Seismic hazard model of New Zealand showing the expected PGA values with 2500-year return period. The seismic hazard of the Macraes mine is moderate comparing with the high seismic hazard expected where the M7.8 Kaikoura earthquake occurred (modified from Stirling et al., 2012).**



## <span id="page-22-0"></span>**8.3.1 Earthquake Return Periods**

As part of this analysis a 10-year return period local event is based on a general assessment of the local earthquake records from the GNS/GeoNet database<sup>([5\)](#page-22-1)</sup> considering the following aspects.

- The return period of earthquakes does not correspond to an exact and periodic time of recurrence but rather to an average time of occurrence between earthquakes with similar magnitude measured over a long period of time.
- In earthquake terms, the coverage of the local GNS/GeoNet database is very short (i.e. 52 years) and this may affect the estimation of the return periods, especially for large magnitudes which have longer recurrence.
- The completeness of the earthquake database, which refers to the date from which all earthquakes of certain magnitude have been recorded. According to Litchfield et al (2005) the database is complete:
	- Post 1840 for magnitudes greater than 6.5,
	- Post 1940 for magnitudes greater than 5, and
	- Post 1964 for magnitudes greater than 4.
- The 505 earthquakes of ≤ M3 recorded since 1978, suggest more than 13 earthquakes with these characteristics occur each year. The return period of M3 or less earthquakes is much shorter than the 10-year return period.
- Similarly, the 80 M3 to M4 earthquakes recorded during the last 50 years, indicate return periods shorter than one year.
- 11 earthquakes with magnitudes between M4 and M4.3 have been recorded since 1978. This suggests an average 3 to 4 years return period for these events.
- One earthquake of M4.6 and one earthquake of M5.1 have occurred since the local database began in 1965. This suggests an average 25-year return period for events with similar magnitudes during the short coverage of the database.
- No earthquakes greater than M5.1 have been recorded in the last 52 years, since records began for the area in 1965. The database for events greater than M5 is complete during this period. Consequently, it is considered that earthquakes with magnitudes greater than 5.1 have return periods greater than 10 years.

Therefore, earthquakes with magnitudes between M4.5 and M5 are considered likely to represent the events with return periods of about 10 years based on:

• Consideration of uncertainties associated with the short coverage of the database,

<span id="page-22-1"></span> $<sup>(5)</sup>$  It is acknowledged that the magnitudes in the database are reported in two different scales, M</sup> and  $M<sub>L</sub>$ . It is beyond the scope of this report to determine the equivalence between the reported magnitude scales. It is assumed that the magnitude values were equivalent and therefore comparable for the magnitude range considered in the assessment (i.e. M≤5.1).



- The shorter return period (shorter than 10 years) of earthquakes with magnitudes lower than M4.4, and
- The longer return periods (longer than 10 years) of earthquakes with magnitudes greater than M5.1.

This analysis did not consider the likelihood of the events with 10-year return period being associated with a particular seismic source. It conservatively assumes that this type of event may occur at any location around Macraes even though the actual earthquake records show earthquakes with magnitudes similar to the 10-year return period event (M4.5 to 5) occurred more than 30 km from Macraes.

## <span id="page-23-0"></span>**8.3.2 Scenarios**

The effects on Macraes of an earthquake depend, among other factors, on its distance from the mine. To further consider the possible effects of the 10-year return period earthquake on Macraes and the influence of earthquake location, two hypothetical scenarios were assessed:

- Scenario 1 M5.0 earthquake at 30 km from the mine. This scenario was selected from the database which shows that all earthquakes with magnitudes greater than M4 have occurred more than 30 km.
- Scenario 2 M4.5 earthquake on the Billy's Ridge Fault Macraes Fault.

This hypothetical scenario was assessed to estimate the effect of the 10-year return period earthquake occurring at the mine (i.e. 0 km). This scenario assumes a rupture on the Macraes Fault which crosses the mine site and is considered the northwest extension of the Billy's Ridge Fault (Litchfield et al., 2005). The location of the fault was extracted from the GNS active fault database.

This scenario is hypothetical, it does not consider the actual probability of rupture of the Billy's Ridge Fault – Macraes Fault and is only evaluated to illustrate the worst case associated with the 10-year return period earthquake.

#### <span id="page-23-1"></span>**8.3.3 Peak Ground Accelerations**

As the earthquake scenarios (Section 8.3.2) are hypothetical, there is no record of the ground motion at the mine. Hence, the PGA values at the mine site are estimated by ground motion prediction models (GMPE).

The GMPE developed for New Zealand by McVerry et al. (2006) and a global GMPE developed from data from tectonic regions similar to New Zealand (i.e. NGA West 2 models, 2012) were used for this assessment. The input parameters for the GMPE include:

- Distance to earthquake rupture: Taken for this assessment as 30 km (Scenario 1) and 0 km (Scenario 2), respectively.
- Type of fault: Assumed to be a reverse fault (in agreement with the mine's tectonic setting and data from the nearby faults, including the Billy's Ridge Fault).



Site conditions: The mine is a rock site ( $Vs_{30} \approx 1000$  m/s).

Table 8.3 shows the average PGA at the mine site estimated for each scenario and those adopted for limit equilibrium sensitivity analysis as part of this study.

Note that for the first earthquake scenario (i.e. M5.0 at 30 km) all the GMPE (i.e. McVerry et al., 2006 and NGA West 2 models) were used. However, for the second scenario (i.e. M4.5 at 0 km) only NGA West 2 models were used because the magnitude 4.5 of this scenario is outside the magnitude range applicable to the New Zealand models.

#### **TABLE 8.3 ESTIMATED PGA**



## <span id="page-24-0"></span>**8.4 Modelled Sections**

Critical analysis sections were selected based on wall aspect, scale (final height and length), exposure of critical infrastructure (haul roads accessing the pit) and geological condition, as such the north, east and west walls were considered analysis. The topographic depression at the southern end of the pit results in a partial height wall with no major structure intersections, therefore the southern wall was not analysed. A plan of all sections is included in Figure 3.

The north wall intersection of the Horse Flat Fault and the inferred "Fault Plane Two" (as outlined in Section 4.3.2) creates an adverse geometry with the potential to expose the haul road. Based on review of the design, this wall has two benches above the haul road, should failure occur it will likely happen during the excavation and can be managed using operational controls. This scenario will not have an ongoing affect on stability and was therefore not included as part of the stability analysis.

A typical analysis setup is included for reference in Figure 10, this shows the upper bench is formed within weathered pelite schist, beneath the base of oxidation between 12 and 15 m depth is fresh rock for the remaining benches at depth.





**Figure 10: Typical model setup for LE modelling conducted as part of the Deepdell Stage 3 stability assessment.**

## <span id="page-25-0"></span>**8.5 Results**

The limit equilibrium stability analysis indicates all critical pit sections, result in a FoS greater than 1.5 as outlined in Table 8.4. The east wall recorded the lowest FoS at the intermediate pit lake level. Two horizontal seismic loads (as outlined in Table 8.3) have been applied to this section as a sensitivity run to replicate the earthquake response within the pit. The resultant FoS value of 1.2 indicates that the design is suitable for the current PGA at Macraes.

A complete summary of modelled sections are included in Appendix F.



#### **TABLE 8.4 LIMIT EQUILIBRIUM RESULTS**



# <span id="page-26-0"></span>**8.5.1 Effects of Pit Lake Filling**

Groundwater and surface run-off water will enter the void created by mining the Deepdell Stage 3 pit. Over time the water collecting in the void will form a lake. The impact of that lake on the stability of the west wall is assessed in this section.

The rise in the level of the lake increases the pore water pressures within the rock mass near the base of the void's walls. This increase in pore water pressures effectively reduces the strength of the rock mass and also adds to the force which drives movement of the slopes, as illustrated in the equations below  $^{(6)}$  $^{(6)}$  $^{(6)}$ .



<span id="page-26-1"></span><sup>&</sup>lt;sup>(6)</sup> Modified from Wyllie & Mah *Rock slope engineering*, 4<sup>th</sup> edition Spon Press



Hence, the initial water level rise decreases the stability of the void's walls. Analyses carried out for similar conditions in Deepdell's final void suggest a lake level 60 m above the base of the void decreases wall stability by approximately 12%.

Rises in the lake level, while increasing the pore water pressures acting in the rock mass, also add weight to the base of the void and to the toe of the void's walls. This additional weight increases the resisting force, by effectively increasing the term Wcos<sub>Φp</sub>tanφ in the above equations. The analyses of the Deepdell's final void suggest the balance is in favour of increasing stability with lake level rises higher than 90 m above the base of the void. The overall increase in wall stability is greater than 10%.

Figure 11 presents a chart of the results to illustrate the effects the rising water level has on stability at Deepdell.



#### **Figure 11: Stability results illustrating the FoS response to changes in pit lake level during closure filling.**

## <span id="page-27-0"></span>**8.5.2 Proposed Waste Rock Stacks**

To accommodate existing backfill, and waste generated from development of the Stage 3 pit design two waste rock stacks are proposed. This material will fill the existing Deepdell South pit and require a new site located approximately 125 m to the north of the final pit boundary (Figure 3). The northern offset is sufficient that no additional loads will be placed on the adjacent pit wall; as such this will not have an effect on stability.



## **9 CONCLUSIONS**

Geotechnical stability assessments undertaken by PSM has shown to have a minimum FoS  $\geq$  1.5. This exceeds the general acceptance level for mine slope stability. A sensitivity analysis was carried out on the east wall for test the design for seismic loading. A horizontal load of 0.13g (equivalent to a M 4.5 earthquake 0 km from the mine) was applied in the analysis that provided a resultant  $FoS = 1.2$  indicating that the overall slope would remain stable.

Filling of an in pit lake at the cessation of mining has been analysed to show while the relative stability reduces while the water level is in at intermediate depth, the ultimate stability of the slopes increases beyond operational levels at final filling.

The proposed waste rock stacks to the north and south of the pit will have no effect on stability.

## **10 CLOSING**

We trust this is in keeping with your immediate requirements. Should you have any queries please do not hesitate to contact the undersigned.

For and on behalf of PELLS SULLIVAN MEYNINK

RICHARD BREHAUT ROBERT BERTUZZI

Principal

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<mark>As Mined Design Contour</mark><br>Proposed Stage 3 Design LEGEND<br>
As Mined Design Cont<br>
Proposed Stage 3 Des<br>
Pit Lake Filling Extents

69600E 69800E 17600N  $\frac{1}{\sqrt{17200}}$ 1700 N 1680<br>1680a 1600



As Mined Design Contours









**PSM** 幽

**APPENDIX A**

**GEOTECHNICAL TRENCH EXCAVATIONS – STEREONETS AND HISTOGRAMS**





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**APPENDIX B**

**BOREHOLE LOGS**



# $PSM$

## Pells Sullivan Meynink Pty Ltd<br>
Engineering Consultants<br>
Rock-Soil-Water

### **Summary Borehole Log**

Hole No: DDH39



## $PSM$

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#### **Pells Sullivan Meynink Pty Ltd**

Engineering Consultants<br>Rock-Soil-Water

A.C.N. 061447621

### **Summary Borehole Log**

Hole No: DDH42














































**APPENDIX C**

**GEOLOGICAL STRUCTURE MODELLING**





# **Pells Sullivan Meynink**

**Engineering Consultants Rock-Soil-Water**

> G3 56 Delhi Road North Ryde NSW 2113 P: 61-2 9812 5000 F: 61-2 9812 5001 mailbox@psm.com.au www.psm.com.au

# *MEMORANDUM*



#### **RE: MODELLING OF MAJOR GEOLOGICAL STRUCTURES - DEEPDELL**

This memorandum summarises PSM's geological interpretation of two dimensional fault structures provided for Deepdell and the development of these into three dimensional surface wireframes for use in analysis.

#### **1 AVAILABLE DATA**

OceanaGold provided the following material:

- 2D contour plan fault traces, Figure 1:
	- Horse Flat
	- Table top
	- Cyanide Creek
- Historic failure mapping for Deepdell North, Figure 2
- 3D fault planes as "DD\_Fault\_plane\_1–3.dxf"
- 3D pit geometry surfaces:
	- As built
	- As mined
	- Stage 3 Proposed



**Figure 1: Plan traces of Deepdell faults**





**Figure 2: Example of geological structure mapping from historic failures in Deepdell Stage 1 and 2.**



### **2 METHODOLOGY**

PSM utilised the following process to develop 3D fault projections using Maptek's Vulcan.

- 1. Register contour plan traces to the as mined topography "DD\_As\_Mined\_surface.dxf"
- 2. Geological structure orientations educated from historic failure mapping data, Figure 2 were used to for the Cyanide Creek and Table Top Faults. The Horse Flat Fault orientation was educated from "DD\_Fault\_Plane\_2.dxf"
- 3. The Cyanide Creek fault was extended beyond the original trace shown in Figure 1. The strike of this feature lines up almost directly with failures from Deepdell Stage 1 (Figure 2). It is inferred that the fault is likely to have contributed to these instabilities and has therefore been modelled to fit this hypothesis.
- 4. Respective defect orientations then projected "up and down dip" of the structures intersection to create a three dimensional plane.
- 5. Fault plane projections were relimited against the Footwall Fault for completeness

#### **3 RESULTS**

The resultant fault projections are illustrated in Figure 3. These structures have been overlain against the Deepdell Stage 1 "as mined" topographic surface and the proposed Stage 3 pit shell for context.

PSM note that the Footwall Fault strikes North to South throughout the operation. The modelled Cyanide Creek Fault appears to have been rotated approximately 30 degrees to Southeast. Such a structural arrangement is atypical when compared to general site experience and may indicate further structural complexity in the area.





**Figure 3: 3D Fault surfaces with outline of previous failures (Highlighted with a red line)**

For and on behalf of PELLS SULLIVAN MEYNINK

Richard Brehaut Associate



**APPENDIX D**

### **KINEMATIC STABILITY ASSESSMENTS**





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 $ACN$  061-447-621







KinematicStats Chart for Planar 09-08-2000 12:23 PM



**APPENDIX E** 

**QUALITATIVE RISK ASSESSMENT** 



#### **APPENDIX E QUALITATIVE RISK ASSESSMENT – DEEPDELL**





**APPENDIX F**

**LIMIT EQUILIBRIUM STABILITY ANALYSIS**





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Y:\0000\PSM071\S071\Deepdell\_Sections\East1\_closure\_445mRL\_Seismic013.slim


Y:\0000\PSM071\S071\Deepdell\_Sections\East1\_closure\_475mRL.slim



Y:\0000\PSM071\S071\Deepdell\_Sections\North2.slim



Y:\0000\PSM071\S071\Deepdell\_Sections\North2\_closure\_445mRL.slim



![](_page_112_Figure_0.jpeg)

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![](_page_113_Figure_0.jpeg)

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![](_page_114_Figure_0.jpeg)

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![](_page_115_Figure_0.jpeg)

G3 56 Delhi Road North Ryde NSW 2113 **P** +61-2 9812 5000 **F** +61-2 9812 5001

**E** mailbox@psm.com.au

<span id="page-115-0"></span>**www.psm.com.au**

Our Ref: PSM71-238L

12 June 2019

Planning Engineer OceanaGold Corporation Golden Point Road RD3, Macraes Flat 9483 East Otago NEW ZEALAND Louie.Herrero@oceanagold.com

Attention: Louie Herrero

Dear Louie

## **RE: GEOTECHNICAL REVIEW OF UPDATED DEEPDELL STAGE 3 PIT**

#### **1. Introduction**

This letter presents a geotechnical review of the proposed Deepdell Stage 3 Pit carried out by PSM at the request of Louie Herrero of OceanaGold Corporation (OceanaGold)<sup>(1)</sup>. The proposed pit was recently slightly revised by OceanaGold. This letter therefore supplements our previous review of the pit <sup>(2)</sup>. For ease of reference, sections of that previous review which remain applicable, such as the geological setting and geotechnical model, are reproduced herein.

There have been three prior phases of mining at Deepdell; the Stage 1 and 2 pits plus the more prominent Deepdell south pit. The Stage 1 and 2 pits are partly backfilled with a waste rock. The proposed Stage 3 pit extends the Stage 1 and 2 pits to the east [\(Figure 1\)](#page-116-0).

#### **2. Scope**

The following scope of work was carried out for this review.

- Slope stability analysis of the Deepdell Stage 3 pit design
- Geotechnical assessment of the interaction between the Deepdell pit and waste rock stack
- The analysis and assessment were done under three conditions:
	- ‒ Operational during mining
	- ‒ Closure post mining.
		- $\circ$  Scenario 1 A pit lake forms to the lowest intersection point with the topography at the southern end of the pit, at approximately RL 445m
		- $\circ$  Scenario 2 A pit lake fills to the lowest intersection point of the northern end of the pit, approximately at RL 475m.

 $(1)$  email from Louie Herrero to Robert Bertuzzi dated 15<sup>th</sup> May 2019

<sup>&</sup>lt;sup>(2)</sup> Geotechnical review of Deepdell Pit Stage 3 Design, PSM71-224R, 16<sup>th</sup> February 2018

![](_page_116_Picture_0.jpeg)

# <span id="page-116-0"></span>**Figure 1: Plan view of Deepdell's existing Stages 1 & 2 (in khaki shade) and the proposed Stage 3 pit shell (darker green) which overlies Stages 1 & 2. The adjacent rock waste stack is shown in a light shade to the east of the pit. The major faults encountered in Stage 1 and 2 pits are also shown.**

# **3. Supplied Information**

OceanaGold supplied the following information for this review.

- Pit design (190327\_DD\_pit.dxf)
- Waste Rock Stack design (DDWRS East)
- Existing Surface (As\_Built\_surface.dxf)
- Mined out surface (As-Mined\_surface.dxf).

### **4. Geotechnical Model**

A detailed description of the geological setting and geotechnical model for Deepdell is presented in our previous review<sup>[\(2\)](#page-115-0)</sup>. Below is a summary of the salient points.

The dominant structural trends at Deepdell are presented in the stereonet in [Figure 2.](#page-117-0)

![](_page_117_Figure_3.jpeg)

# <span id="page-117-0"></span>**Figure 2: Stereonet illustrating the geological structure trends at Deepdell: foliation (green), joints (blue) and gouge / shear zones (red).**

• Foliation dips at between 5° and 45° to the east / south east

Macraes' experience is that foliation is typically not a controlling defect for pit wall stability

• Joints are typically sub vertical to steeply dipping to the south and west. Secondary sets dip steeply to sub vertical to the east and north

Visible length typically in the range  $0.5 - 1.5$  m and continuous out of exposure. Spacing between joint planes typically ranges from <0.25 m to 0.75 m. The joints are mostly undulose on the large scale, rough on the small scale and tight. Experience from other pits at Macraes tends to indicate that the typical length of the majority of joints is less than 5 m.

- Three zones of gouge / crushed rock and shearing are apparent:
	- The first is flat to moderately dipping to the south to southeast, although a small number of defects are indicated as dipping to the southwest
	- The second is moderately to steeply dipping to the southeast
	- The third dips moderately to steeply to the north but is poorly defined. These structures are favourably orientated with respective to the pit walls as they dip approximately perpendicular to the major wall aspects.

Visible length typically in the range of 1.0 to 5 m, and continuous out of exposure. The defects are undulose on the large scale and smooth on the small scale. The typical infill type is gouge/rock fragments and is moist to damp. Infill thicknesses are variable; however, appear to be typically in the ranges of 20 to 50 mm and 200 to 500 mm.

Major geological structures at Macraes are the Footwall Fault (FF) and Hanging Wall Shear (HWS). The FF maintains a relatively consistent orientation and has a dip and dip direction in Deepdell of approximately 12° / 075°. The FF will be approximately 90 m beneath the proposed Stage 3 pit. A cross section illustrating this geometry is included in [Figure 3.](#page-118-0)

![](_page_118_Figure_0.jpeg)

## <span id="page-118-0"></span>**Figure 3: West-east section through the deepest part of the proposed Stage 3 pit at approximately 17200 mN. The distance between the FF and the pit is more than 90 m.**

## **5. Slope Design**

The proposed Stage 3 pit focusses on developing the prior Stages 1 and 2 pits towards the east.

- The pit floor deepens by approximately 70 m
- The northern wall lengthens but with a favourable slope aspect (180°)
- The eastern wall minimises orientations which have proved to be more susceptible to slope instability
- There is effectively no southern wall.

The slope design for the Stage 3 pit follows that successfully used in Deepdell South, namely:

- Batters in the weathered / oxide material are 10 m high and 60° all around the pit
- 15 m high, 60° batters with 7.5 m berms for the northern and eastern walls, resulting in a 43° interramp slope angle (3)
- 70° batters and 5 m berms for the southern and western walls, resulting in a 55° inter-ramp slope angle. In practice, the west wall is much shallower (23 to 38°) as it follows the ore zone.

This design is consistent with the kinematic and statistical analyses which were carried out in our previous review [\(2\)](#page-115-0) for the eastern wall and are presented in [Table 1.](#page-118-1)

Kinematic analysis is a stereographic based technique that considers the mean orientation of defect sets, the slope orientation, the defect shear strength and potential failure modes. It is an assessment of the geometrical possibility of simple block movements, i.e. planar sliding, wedge failure and toppling. Statistical analysis extends this by considering the scatter in defect dip and orientation.

<span id="page-118-1"></span>![](_page_118_Picture_240.jpeg)

![](_page_118_Picture_241.jpeg)

<sup>(3)</sup> inter-ramp slope angles are defined as toe-to-toe

### **6. Material Properties**

#### $6.1$ **Rock mass**

There are two main approaches to estimate rock mass strengths; back-analyse failures and rock mass classification. The rock mass strengths used by PSM in studies at Macraes are based on back-analyses and the GSI<sup>(4)</sup> / Hoek-Brown classification [\(Table 2\)](#page-119-0). Uncertainty in these properties is catered for by including variability. In most analyses this is done by way of normal distributions based on the mean ± standard deviation.

<span id="page-119-0"></span>![](_page_119_Picture_322.jpeg)

![](_page_119_Picture_323.jpeg)

For comparison, Bertuzzi et al (2017)<sup>(5)</sup> suggested typical values of GSI<sup>(6)</sup> for the schist encountered at Macraes:

- GSI = 75 (70 to 80): Class A schist lithified rock with frequent defects and rare shearing
- $\bullet$  GSI = 60 (45 to 70): Class B schist fractured rock with frequent defects and some shearing
- GSI = 30 (25 to 35): Class C schist fractured to fragmented rock with frequent shearing
- GSI = 15 (12 to 20): Class D schist fragmented/sheared rock.

The weathered schist can be thought of as equivalent to Class C and the inter-shear pelite as Class B / Class C. The GSI / Hoek-Brown system would lead to the greater strengths than the values used at Macraes (refer to [Table 3\)](#page-119-1).

<span id="page-119-1"></span>![](_page_119_Picture_324.jpeg)

![](_page_119_Picture_325.jpeg)

<sup>(4)</sup> Geological Strength Index

<sup>&</sup>lt;sup>(5)</sup> Bertuzzi R, Douglas K, Mostyn G. Comparison of quantified and chart GSI for four rock masses, Engineering Geology 202 (2016), pp 24–35

<sup>(6)</sup> Geological Strength Index

#### 6.2 **Waste rock**

In 2013 a study of the waste rock stacks at Macraes was carried out to address the absence of site specific data <sup>(7)</sup>. Rill angles, which can be considered as lower bound estimates of friction angle, were observed as; 35 to 38° for fresh schist, 34 to 35° for sheared pelite and 29 to 34° for oxidised schist. Published literature suggests friction angles in the order of 45 to 35 °, decreasing as a function of increasing normal pressure and increasing cohesion [\(Figure 4\)](#page-120-0). Others for different work at Macraes have utilised such a relationship, e.g. shear strength =  $1.29$  x normal stress  $^{(8)}$ .

![](_page_120_Figure_2.jpeg)

## <span id="page-120-0"></span>**Figure 4: Left chart: Friction angle as a function of normal stress within a rock fill from Linero et al (2007) (9) reported in 2013 study () . Right chart: comparison of adopted shear strength.**

As the waste rock stack is physically only a minor part of the stability analysis for this review, adopting a conservative shear strength for it does not impact on the design. Hence, conservative Mohr-Coulomb parameters of  $c' = 1$  kPa and  $\phi' = 35^\circ$  for the shear strength of waste rock used [\(Figure 4\)](#page-120-0).

#### 6.3 **Summary**

The relevant materials for the proposed Deepdell Stage 3 pit are the inter-shear pelite which is overlain by a thin layer of weathered schist. In this case, as the weathered schist has minor influence on overall stability only typical values are used for it. The properties adopted for this review are presented in [Table 4.](#page-120-1)

<span id="page-120-1"></span>**Table 4 – Adopted rock mass strength properties**

<b>Material</b>	Unit Weight (kN/m <sup>3</sup> ) mean $\pm$ sd, range	<b>Cohesion (kPa)</b> mean $\pm$ sd, range	<b>Friction Angle (°)</b> mean $\pm$ sd, range			
Waste rock	20		35			
Weathered schist	25	120	35			
Inter-shear pelite	$25 - 3 + 2$ , $22 - 27$	$180 - 15 + 25$ , $170 - 205$	$43 - 5 + 2$ , $38 - 45$			

<sup>(7)</sup> Coronation waste rock stack stability, PSM71-135L Rev C dated  $28<sup>th</sup>$  March 2013: Waste dump study, PSM71-145R dated 1<sup>st</sup> May 2013 (8) e.g. Engineering Geology

<sup>(9)</sup> Linero. S., Palma. C., & Apablaza. R. (2007) Geotechnical Characterisation of Waste Material in Very High Dumps with Large Scale Triaxial Testing, Australian Centre for Geomechanics, Perth.

## **7. Earthquake**

Previous earthquake scenario assessments undertaken for the Macraes pits concluded that a hypothetical and conservatively worse-case of an M4.5 earthquake on the Macraes Fault (i.e. at a distance of 0 km to the mine), yielded a peak ground acceleration (PGA) of 0.13g <sup>(10,11)</sup>.

<span id="page-121-0"></span>The 0.13g PGA is adopted in stability analyses as the operating basis earthquake (OBE) as it corresponds to an earthquake of 150-year return period at the mine site according to Litchfield et al (2005)<sup>(12)</sup>. The maximum design earthquake (MDE) has an equivalent peak ground acceleration of 0.65g, which is applicable to 1 in 2500-year return event <sup>[\(12\)](#page-121-0)</sup>.

No amplification factor is applied. Earthquake amplification depends on many factors including topography, material stiffness and thickness. Normally, amplification factors are introduced when the structure is founded on soils. In this case, the waste rock stack is founded on *in situ* rock and the open pit slope is within *in situ* rock. Hence, no amplification factor is warranted.

### **8. Acceptance Criteria**

#### $8<sub>1</sub>$ **Static**

Minimum values of the Factor of Safety (FoS) were adopted as the acceptance criteria as provided in Table 9.9 of Read & Stacey 2009, Open Pit Slope Design (13).

- <span id="page-121-1"></span>• FoS ≥ 1.2 as commonly accepted for mining applications
- <span id="page-121-3"></span><span id="page-121-2"></span> $FoS \ge 1.5$  for pit closures.

#### 8.2 **Earthquake**

A pseudo-static FoS ≥ 1.0 is the target when assessing earthquakes. Should the pseudo-static FoS be less than 1.0, additional analyses will be carried out to assess the magnitudes of displacements predicted to occur under the seismic loading. These additional analyses will follow the method of Bray and Travasarou (14, 15) which was developed from a large suite of numerical simulations calibrated against observed seismic slope performance.

During an earthquake, the seismic force may act in one direction for only a few tenths of a second before its direction is changed. The result is a series of pulses. For every pulse that exceeds a critical value, which is dependent on the site conditions, some small amount of permanent displacement will occur. In the Bray and Travasarou method, it is acceptable to select a seismic coefficient (k) that is a fraction of the maximum seismic demand (PGA) because exceeding the maximum seismic resistance for a few instances leads only to minor seismic displacement. Satisfactory performance is therefore determined by whether or not the accumulated displacement is tolerable.

The California Geological Survey (2008) guidelines for mitigating seismic hazards (16) state that for natural slopes, predicted displacements:

- Less than 15 cm are unlikely to correspond to serious landslide movement and damage
- Between 15 100 cm could be serious enough to cause stress loss and failure
- Greater than 100 cm are very likely to correspond to damaging slide movement.

 $(10)$  Assessment of earthquake effects on pit walls, PSM71-215R dated  $6<sup>th</sup>$  May 2017

<sup>(11)</sup> Assessment of 10-year return period earthquake effects on pit walls, PSM71-216R dated 22nd May 2017

<sup>(12)</sup> Litchfield N, McVerry G, Smith W, Berryman K and Stirling M. Seismic study for tailings embankments at Macraes Gold Project. Institute of Geological and Nuclear Sciences, Client report 2005/135, October 2005

<sup>(13)</sup> Guidelines for open pit slope design, Ed. Read & Stacey, CRC Press, 2009.

<sup>(14)</sup> Bray JD, Travasarou T, 2007. Simplified procedure for estimating earthquake-induced deviatoric slope displacements. Journal of Geotechnical and Geoenvironmental Engineering, ASCE, April 2007, pp. 381-392

<sup>(15)</sup> Bray JD, Travasarou T, 2009. Pseudostatic Coefficient for Use in Simplified Seismic Slope Stability Evaluation. J. of Geotechnical and Geoenv. Engineering, ASCE, 135(9), 2009, 1336-1340

<sup>(16)</sup> Jibson RW, 2011. Methods for assessing the stability of slopes during earthquakes—A retrospective: Engineering Geology, v. 122, p. 43-50

These guidelines are for natural slopes and are likely to be conservative for open pit rock slopes. Published literature indicates that open pit rock slopes are not significantly impacted by earthquakes ([13,](#page-121-1) 17, 18).

Assuming the allowable accumulated displacement during an earthquake is 100 cm for the open pit rock slope, the equivalent seismic coefficient for the MDE is  $k = 0.14q$  following the Bray and Travasarou method.

#### **9. Limit Equilbruim Analysis**

#### **Models**  $9.1$

The limit equilibrium analyses were carried out using RocScience's Slide software. The critical slip surfaces were found by the software.

Five sections through the proposed Deepdell Stage 3 pit are used to assess slope stability of the eastern wall and the interaction between the pit and the waste rock stack. The locations of the sections are shown in [Figure](#page-122-0)  [5.](#page-122-0) The sections themselves are shown in [Figure 6.](#page-123-0) Two of the sections  $- B \& E -$  are also used to assess the overall slope stability of the northern and western walls, respectively.

![](_page_122_Figure_6.jpeg)

<span id="page-122-0"></span>**Figure 5: Plan view of the Deepdell Stage 3 design and adjacent rock waste stack showing sections lines considered for limit equilibrium stability analysis.**

<sup>(17)</sup> Lorig L, 2016. Designing for extreme events in open pit slope stability. Journal of the Southern African Institute of Mining and Metallurgy, 116(5), pp.387-398

<sup>(18)</sup> Azhari A, 2016. Evaluating the effects of earthquakes on open pit mine slopes, PhD Thesis, Colorado School of Mines

![](_page_123_Figure_0.jpeg)

<span id="page-123-0"></span>**Figure 6: Sections A, B, C, D & E (top to bottom) though the Deepdell Stage 3 design and adjacent rock waste stack.**

Groundwater pressures were included in the analyses by adopting conservatively high groundwater surfaces, as shown in [Figure 7](#page-124-0) for example.

Safety Factor 0.000 <b>SCR</b> 0.250 0.500 0.750 1.000 1.250 1.500 1.750 2.000 2.250 8 <sup>o</sup> 2.500 2.750 3.000 3.250 3.500 3.750 4.000 4.250 4.500 $rac{400}{400}$ 4.750			1.738		W <u>00 0 00 </u> ═							
5.000 5.250 5.500 5.750			ы	<b>Material Name</b>	Color	<b>Unit Weight</b> (kN/m3)	<b>Strength Type</b>	Cohesion (kPa)	Phi (deg)	<b>Water Surface</b>	Hu Type Hu	
	$6.000+$			<b>Weathered Schist</b>		25	Mohr-Coulomb	120	35	<b>Water Surface</b>	Custom	$\mathbf{1}$
				<b>Intershear Pelite</b>		25	Mohr-Coulomb	180	43	Water Surface	Custom	1
$\frac{8}{3}$				<b>Waste Rock</b>		20	Mohr-Coulomb	$\mathbf{1}$	35	<b>Water Surface</b>	Custom	1
$\frac{8}{2}$												
	2100	2200	2300	2400		2500		2600		2700		

<span id="page-124-0"></span>**Figure 7: Example of Slide analysis showing the assumed high groundwater surface.**

#### $9.2$ **Results**

# **9.2.1 Static**

The results of the limit equilibrium analyses are presented in Appendix A and summarised i[n Table 5.](#page-124-1) Minimum FoS for all the deterministic cases exceeds 1.5.

The minimum FoS of 1.47 and 1.49 for the sensitivity analyses of the pit closure scenario 1 case (RL 425 m lake level), were obtained with the lower bound friction angle of 38°. Considering that these results are for the lower bound sensitivity analyses and the general conservatism built into the model (strength properties, groundwater load) the 2% difference in the minimum FoS from 1.5 is inconsequential.

![](_page_124_Picture_225.jpeg)

### <span id="page-124-1"></span>**Table 5 – Results of LE analyses**

![](_page_125_Picture_289.jpeg)

## **9.2.2 Waste rock stack**

The analyses presented in Appendix A did not identify minimum FoS for failure paths in the waste rock stack. The overall slope of the waste rock stack is less than 18°, typically 15°. A failure path confined to the face of the waste rock stack, is the infinite slope problem with the FoS calculated as the ratio of the friction angle to the slope (ignoring any benefit of cohesion). That is FoS ≥ tan 35° / tan 18° ≥ 2.16. The FoS for slip circles that pass through the waste rock stack, and not just along its surface, will therefore be greater than 2.16, i.e. above the acceptance criteria.

# **9.2.3 Earthquake**

The effects of earthquakes were assessed by applying horizontal accelerations to the models in a pseudo-static analysis. Section C was used for the assessment as it provided the lowest static FoS (all other sections would yield higher FoS). The results are presented in Appendix B and summarised in [Table 6.](#page-125-0)

<span id="page-125-0"></span>![](_page_125_Picture_290.jpeg)

![](_page_125_Picture_291.jpeg)

\* FoS < 1.0 means permanent displacement will occur under this seismic event

The FoS for the operational cases, i.e. OBE (0.13g) and amplified OBE (0.2g), exceed 1.0. The FoS for the closure scenarios under the MDE (0.65g) is less than 1.0 which indicates that permanent displacement of the slopes is expected. Hence, additional analyses were carried out using the Bray and Travasarou method [\(14,](#page-121-2)[15\)](#page-121-3) Accepting a permanent displacement of 100 cm gives a resultant seismic coefficient of 0.14g. The FoS results with 0.14g exceed 1.0 and hence meet the acceptance criterion.

### **10. Conclusions**

This letter presents the results of a geotechnical review of the Deepdell Stage 3 pit design including the interaction between the pit and waste rock stack. Slope stability was assessed for the expected conditions during mining as well conditions post mining. This includes two pit lake levels. The impacts of a 1 in 150-year seismic event occurring during mining and of a 1 in 2500-year seismic event occurring post-mining were assessed.

The results of the analyses for the proposed pit during mining and closure are in keeping with the acceptance criteria.

The results of the analyses suggest that should a 1 in 2500-year earthquake effect Deepdell, then permanent displacement in the order of 100 cm will occur but the slopes will remain stable.

The risk of landslides triggering a wave-overtopping event in the pit lake is negligible because the open pit rock slopes are stable under earthquake events.

We trust this letter is in keeping with your requirements.

For and on behalf of

# **PELLS SULLIVAN MEYNINK**

**ROBERT BERTUZZI PRINCIPAL**

**Appendix A Results of Slide Analyses – Static Cases**

![](_page_128_Figure_0.jpeg)

![](_page_129_Figure_0.jpeg)

![](_page_130_Figure_0.jpeg)

![](_page_131_Figure_0.jpeg)

![](_page_132_Figure_0.jpeg)

![](_page_133_Figure_0.jpeg)

![](_page_134_Figure_0.jpeg)

![](_page_135_Figure_0.jpeg)

![](_page_136_Figure_0.jpeg)

![](_page_137_Figure_0.jpeg)

![](_page_138_Figure_0.jpeg)

![](_page_139_Figure_0.jpeg)

![](_page_140_Figure_0.jpeg)

![](_page_141_Figure_0.jpeg)

![](_page_142_Figure_0.jpeg)

![](_page_143_Figure_0.jpeg)




















































**Appendix B Results of Slide Analyses – Earthquake Cases**













#### **Engineering Geology Ltd**

- $3 + 6494862546$
- × info@egl.co.nz
- Unit 7C, 331 Rosedale Road, Albany, Auckland<br>PO Box 301054, Albany, Auckland 0752  $\bullet$
- www.egl.co.nz

EGL Ref: 8528

## **MACRAES GOLD PROJECT DEEPDELL EAST WASTE ROCK STACK DESIGN REPORT**

Prepared for: 8 November 2019

Oceana Gold (New Zealand) Ltd PO Box 5442 **DUNEDIN 9058**



**CONTENTS**







**Engineering Geology Ltd** 

- $3 + 6494862546$
- info@egl.co.nz  $\mathsf{M}$
- Unit 7C, 331 Rosedale Road, Albany, Auckland PO Box 301054, Albany, Auckland 0752
- to www.eql.co.nz

EGL Ref: 8528

8 November 2019

# **MACRAES GOLD PROJECT DEEPDELL EAST WASTE ROCK STACK DESIGN REPORT**

# <span id="page-178-0"></span>**1.0 INTRODUCTION**

Oceana Gold (New Zealand) Ltd (OceanaGold) operates the gold mine, known as the Macraes Gold Project (MGP), at Macraes Flat in East Otago. The mine is located between Middlemarch and Palmerston as shown in Figure 1. Gold extraction from the current mining operation involves mining of open pits and underground (Frasers Underground). Associated with the MGP are waste rock stacks for disposal of pit overburden material and tailings storage facilities for disposal of tailings.

The Deepdell North Stage III project is located on the northern side of Deepdell Creek as shown in Figure 2. The project involves the following:

- Re-mining of an extension of the Deepdell North Pit located immediately south of Horse Flat Road, to be known as Deepdell North Stage III Pit.
- Construction of a new waste rock stack, to be known as Deepdell East Waste Rock Stack (WRS), immediately south of Horse Flat Road. This will include backfill of the Deepdell South Pit.
- Partial realignment of Horse Flat Road past the WRS.

There is an existing haul road between the Coronation North Project to the north and the MGP Process Plant located on the southern side of Deepdell Creek. The haul road is on the western side of Deepdell North Stage III project (refer Figure 3) and will provide access to the new pit and WRS. No significant length of new haul road will therefore be required for the project other than a short length to access the WRS.

Details of the proposed Deepdell North Project III comprises:

- Deepdell North Stage III Pit will produce 3.5Mt of ore and 53.3Mt of waste rock. The new pit comprises an extension to the existing Deepdell North Pit (Stage 2). The footprint will be 38ha of which 18.7ha was previously disturbed by mining.
- Deepdell East WRS comprises backfilling of the existing Deepdell South Pit and will approximately re-establish the original ground contours, before raising the ground profile to the north. At the WRS northern extent the WRS crosses Horse Flat Road and the road is to be realigned. Overall the WRS has a footprint of 70.8ha and a storage capacity of 21.6Mm<sup>3</sup>, to a crest height of RL540. The design crest level for consent is RL580 and this is the level has been used for design in this report

This design report by Engineering Geology Limited (EGL) is for the Deepdell East WRS. Pells Sullivan Meynink is carrying out the design for Deepdell North Stage III Pit and their





design considers the impact of the open pit on the stability of the WRS and Pit Backfill (Ref. 2). The analyses covered by this report therefore only consider the shallow stability of the WRS and excludes analyses of potential shear failure into the new pit.

All plans grids, references and geological orientations referred to in this report are to mine north, which is approximately 45 degrees anti-clockwise from true north.

### <span id="page-179-0"></span>**2.0 RESOURCE CONSENTS**

Consents held for the existing Coronation North Project are listed in Table 2.1 below and in the Macraes Water Quality Management Plan. It is anticipated that similar consent conditions will apply to the proposed Deepdell East WRS and the design has been carried out on this basis.

### **Table 2.1: Consent Conditions for Existing Coronation North WRS**




Condition 11.2 of land use consent 201.2016.779, 201.2013.360.1, LUC-2016-234 and LUC-2013-225A requires the consent holder to engage a suitably qualified geotechnical engineer to design the waste rock stack and submit a Design Report prior to the construction of the new rock stack.

#### *Condition 11.2.*

*The consent holder shall engage a suitably qualified geotechnical engineer to design the waste rock stack. A construction report shall be prepared for the waste rock stack and this report provided to the Councils prior to the commencement of construction of the waste rock stack. The report shall include details of site formation, design construction, appearance, and testing for stability of the waste rock stack, and shall include evaluation of the long-term stability and performance of the waste rock stack.*

Condition 3 of Resource Consent RM16.138.01 relates to the requirement to construct underdrains in the watercourses that underlie the waste rock stack unless the "Best Practicable Options Report" identifies otherwise.

Condition 11.1 of Land Use Consent 201.2016.779, 201.2013.360.1, LUC-2016-234 and LUC-2013-225A relates to the seismic requirements for the waste rock stack and is addressed within this report.

*The Coronation North waste rock stack shall be designed for operating basis earthquake (OBE) with a recurrence interval of 150 years and maximum design earthquake (MDE) with a recurrence interval of 2,500 years and otherwise shall otherwise be designed in accordance with sound engineering practice.*

# **3.0 SITE AND PROJECT DESCRIPTION**

Deepdell North Stage III is located on a relatively flat plain running east west, which is approximately 1km wide. Deepdell Creek meanders down the southern side of the plain and is incised about 130m lower. The northern side of the plain comprises a mountain range rising up about 200m.

The existing Deepdell South Pit is located on the southern edge of the plain as shown in Figure 3 and extends partway down the steep slope to Deepdell Creek. The sloping ground varies between about 1v:4h to 1v:1.3h. The existing south pit is to be backfilled and merges with the waste rock to be placed north of the pit, on Horse Flat, to form the proposed Deepdell East WRS. On Horse Flat, a local high point lies beneath the centre of the WRS with ground sloping to the east, west and south. To the west of the WRS, the ground slopes relatively gently at approximately 1v:30h to 1v:15h. To the east the ground also slopes relatively gently at 1v:30h to 1v:10h which steepens into the gully side slopes of approximately 1v:3h, down to Deepdell Creek. The northern toe of the WRS approaches the toe of the hills to the north of Horse Flat.

The new Deepdell North Stage III Pit is north west of the existing Deepdell South Pit and located over the central southern area of the plain, immediately south of Horse Flat Road, east of the existing haul road and west of the Deepdell East WRS as shown in Figure 3. This area has been previously mined in part for Deepdell North Pit Stage II. The new pit is to be excavated deeper and extended further than the previous pit.

All stormwater runoff from Deepdell North Stage III project will drain via a series of gullies and creeks to Deepdell Creek, that forms part of the Shag Catchment.

The batter slopes of the WRS have been designed to blend as naturally as possible with the natural landscape. The maximum height of the WRS is approximately 200m and the outer shoulders have an overall slope of about  $1(v):3(h)$ . The WRS and Deepdell South Pit are

<sup>•</sup> *Condition 3. Underdrains shall be constructed in the natural channels that form the unnamed tributaries of Maori Hen Creek, Trimbells Gully, Mare Burn and Coal Creek beneath the footprint of the Coronation North Waste Rock Stack by placement of large rocks covered by appropriately graded material to provide sufficient filtering to prevent blockage of the drains by finer material unless it is identified in the "Best Practicable Options Report" required by Condition 5 of Consent RM16.136.02 that underdrains should be constructed differently or not be constructed at all.*

<sup>•</sup> *Condition 11.1*

shown in Figure 4 with the underlying topography and the location of the cross sections. The design profile of Deepdell East WRS is shown on cross sections (X1-X1' and X2-X2') in Figure 5 and the long sections (L1-L1' and L2-L2') in Figure 6.

# **4.0 GEOLOGY AND GEOTECHNICAL INVESTIGATION**

# **4.1. Regional Geology**

The basement rock in Central and East Otago comprises Otago schist. The Otago schist is primarily composed of psammitic and pelitic grey schist derived from metamorphism of Mesozoic age sandstone and mudstone. In the area of Macraes Flat, the rocks have been metamorphosed to green schist metamorphic facies, giving a strongly foliated fabric of dark grey micaceous and light grey quartz-rich laminations.

From previous geotechnical investigations for the MGP, it is apparent that the prominent geological structure includes a well-developed schistosity with two dominant fault sets. West of the Footwall Fault, that defines the footwall of the Hyde – Macraes Shear Zone (HMSZ). The schistosity is folded and has a varying trend over the project area revealing a series of anticlines and synclines. Foliation dips either to the northwest, north, west or south west. East of the Footwall Fault (Hanging wall) the schistosity has more of an easterly trend. At Coronation the Footwall Fault position is inferred as a subtle feature on the landscape. The WRS is located to the south of the proposed Coronation North Pit and east of the both the Footwall Fault and the Hanging Wall Shear.

The major set of faults has an eastern trend. They exhibit Miocene (recent tectonic) deformations and are related to the formation of the Alpine Fault. This deformation has faulted and folded the surface within Central and East Otago to produce the presentday basin and range topography.

The second set of faults has a northern trend, and the most significant of these is the Hyde-Macraes Shear Zone.

The Hyde–Macraes Shear Zone (HMSZ) comprises a mineralised shear zone which has been mapped for at least 25km by OceanaGold geologists. The HMSZ represents the principal gold bearing ore body exploited by OceanaGold and generally strikes north and dips at about  $15^{\circ}$  to the east. Tectonic displacement associated with the HMSZ is inferred to be in the order of hundreds of metres, with this movement initiating some 120 to 150 million years ago. The ore-schist zone of the HMSZ consists of predominantly pelite and semipelite, but includes blocks of psammite, typically well foliated and containing mineralised quartz veins.

# **4.2. Geotechnical Investigation**

Specific geotechnical investigation for Deepdell East WRS comprised field mapping and test pits. Test pits were excavated on Horse Flat where the schist is mantled by a layer of loess.

# **4.2.1. Fieldwork**

The fieldwork was carried out on the  $27<sup>th</sup>$  and  $28<sup>th</sup>$  of June 2019. The fieldwork comprised a walkover survey, mapping and the excavation and logging of test pits by a senior engineering geologist. The results of the field mapping are shown in Figure 7. The test pit logs are in Appendix A.

# **4.2.2. Soils**

The prevalent rock outcrops and head scarps of shallow slips observed on the sides of gullies and farm tracks, show that there is generally only a thin layer of soil overlying the bedrock on Horse Flat. Soil depths over the extent of the WRS was typically 0.3 to 0.5m with one location (TP4) with 4m of soil and another location (TP3) with 1.3m of soil. The soil comprises loess or residual soil (of the underlying schist) and comprises layers of silt with varying amounts of clay, sand and gravel.

Particle size distributions and Atterberg limit tests were undertaken on three loess soil samples (results are in Appendix B). The results of the three loess samples comprised silt with some clay with a Plasticity Index (PI) of 3 and 12.

# **4.2.3. Schist**

The schist observed on site comprises well foliated, highly to moderately jointed semipsammitic schist.

The foliation is well developed. A recent walk over survey and testpits show that the foliation generally dipping between  $10^{\circ}$  and  $30^{\circ}$  mine south east on Horse Flat. On the northern end of the Deepdell South Pit Backfill, the dip of the foliation varies between  $10^{\circ}$  and  $20^{\circ}$  towards the mine north. On the southern side of the existing Deepdell South Pit Backfill, the foliation dips about  $25^{\circ}$  to the south.

The schist is moderately to highly jointed with joints generally steeply dipping between  $60^{\circ}$  and  $80^{\circ}$  in multiple directions around the Deepdell East WRS. The joints dip between  $60^{\circ}$  and  $80^{\circ}$  to the northwest to northeast on the southern end of the Deepdell South Pit and to the south on the northern end of the Deepdell South Pit (Figure 7). Large tension cracks were observed on the eastern pit wall of the existing pit which were dipping steeply to the north west. Toppling failure was observed across the pit wall in this area.

No strength testing has been undertaken on schist in the Deepdell North Stage III area. However, elsewhere on the Macraes Gold Project, the typical unconfined compressive strength of unweathered schist is between about 20MPa and 40MPa, normal to the foliation. Schist typically has a lower unconfined compressive strength along the direction of foliation. This is reflective of the layered nature of the rock and the presence of weak, mica-rich laminations. It is anticipated that the strength of the schist underlying the proposed WRS will be consistent with that found elsewhere in the Macraes Gold Project area.

# **4.2.4. Inferred Areas of Instability**

No areas of significant historical or incipient instability were observed on site.

Where the creeks dissect the underlying schist bedrock the steepness of the slopes generally vary depending on the local dip of the foliation and discontinuities. Typically, the side slopes to the creeks are steeper where the foliation is dipping into the slope (i.e. governed by block failures) and gentler sloping where the foliation is dipping out of the slope (i.e. governed by block sliding on the foliations and the slopes are typically parallel to the dip).

Localised shallow slumping/instability is evident on the hills to the north of Horse Flat, more than the slopes to the south down to Deepdell Creek. This shallow instability is considered to be associated with erosion and undercutting at the toe of the slopes during heavy rainfall and flow within the creeks.

Soil creep was also inferred on some of the steeper slopes on the hills however this does not effect the WRS. Where localised soil creep was evident, the soil mantle is reasonably thin and can be cleared as part of the foundation stripping.

# **5.0 DESIGN**

# **5.1. Design Life**

The estimated duration of the operation and rehabilitation of the Deepdell East WRS is about 5 years  $(2020 - 2025)$  and will remain in place in perpetuity.

# **5.2. Stability**

# **5.2.1. General**

Engineering Geology Ltd (EGL) has carried out both static and seismic stability analyses for the WRS. The analyses do not include the stability of potential shear failures into the new Deepdell North Stage III Pit. This has been covered by Pells Sullivan Meynink in their design for the pit (Ref.2). They conclude in their report that the eastern offset to the WRS is sufficient that no additional loads will be placed on the adjacent pit wall; as such this will not have an effect on stability.

The stability of the WRS has been analysed using the same design approach and parameters as that used for the existing consented Coronation Project WRS (Ref.3 and Ref.12).

Analyses of long-term static stability of the shoulders of the waste rock stack and stability when subjected to design earthquake loads have been undertaken. Limit equilibrium analyses of the slope have been undertaken using the SLOPE/W program, Geostudio 2012 (Ref.4). The Spencer solution method (Ref 5.) has been used for the analyses of circular potential failure surfaces. The Janbu simplified method (Ref.6) has been used for the analyses of potential block/non- circular failure surfaces.

Limit equilibrium analyses have been undertaken to calculate the Factor of Safety (FoS) for static and seismic loading conditions. Where seismic loading results in a FoS less than one it has also been used to determine the yield coefficient  $(k_y)$  to calculate coseismic displacement.

The possibility of failure through the foundation soils has been considered. The seismic stability of the Deepdell East WRS complies with Condition 11.1 of DCC and WDC Consents (No. 201.2016.779, 201.2013.360.1, LUC-2016-234 and LUC-2013-225A (Ref.3). This condition requires that "*the waste rock stacks shall be designed for operating basis earthquake (OBE) with a recurrence interval of 150 years and maximum design earthquake (MDE) with a recurrence interval of 2,500 years and otherwise shall be designed in accordance with sound engineering practice".* The WRS profile maybe subjected to settlement and horizontal displacement in an earthquake, however, the profile is stable post-earthquake.

Stability analyses have been undertaken for four cross sections through the WRS. These are representative of the critical cross sections in terms of loading, topography and rock foliation/discontinuities. The location of the cross-sections is shown in Figure 4 and the cross sections are shown in the stability analyses included in Appendix C.

# **5.2.2. Waste Rock Characteristics**

The waste rock is anticipated to consist of a mixture of psammitic and pelitic schist. It is to be excavated from the new Deepdell North Stage III Pit. The schist rock varies from completely to slightly weathered, depending on the relative depth of excavation.

Physical characteristics of the excavated rockfill were assessed during the design phase for the tailings embankments and were based on tests conducted on samples of rockfill, schist and other various rock types used for similar projects.

The waste rock to be placed in the WRS and Pit Backfill will be end-tipped, so it is assumed to be non-structural fill. The waste rock segregates when end-tipped, such that each lift (approximately 10-20m high) varies from coarse rock at the bottom to silty sandy rockfill at the top. Consequently, the WRS consists of layers of rockfill of varying permeability. Generally, the rockfill could be expected to be free draining, except at the top of each lift where a thin low permeability layer is created by the trafficking of the dump trucks.

The following shear strength function has been adopted for waste rock which is consistent with that previously used for WRS at MGP and Coronation WRS (Ref.3 and Ref 12):

Shear strength ( $\tau$ ) = 1.29  $\sigma'$  <sup>0.91</sup> (kPa), where  $\sigma'$  is the effective overburden pressure.

The design unit weight used is  $21.5 \text{kN/m}^3$ .

# **5.2.3. Foundation Material Characteristics**

The *in-situ* rock beneath the proposed Deepdell East WRS is similar to that beneath the existing Coronation WRS (Ref.3 and Ref.12), so the same foundation shear strength parameters have been adopted for design. They are summarised below:



The above shear strength parameters are based on shearing through relatively intact rock at about right angles to the foliation dip. As discussed in Section 4.2.3, the measured foliation dips between about  $10^{\circ}$  and  $30^{\circ}$  to mine south east around Horse Flat. On the northern end of the existing Deepdell South Pit, the dip of the foliation varies between  $10^{\circ}$  and  $20^{\circ}$  towards mine north. On the southern side of the Deepdell South Pit the foliation dips about  $25^{\circ}$  to mine south (Figure 7).

A second set of shear strength parameters has therefore been adopted for the rock where failure could potentially occur along the rock foliation and any minor faults/shear zones dipping to the north east. The shear strength parameters are the same as those adopted in similar circumstances for the stability analysis of the existing Coronation WRS (Ref.3 and Ref.12) and given below.

Shear along foliations and minor faults/shear zones Effective cohesion  $=$  47 kPa

Effective friction  $=$  23 degrees

# **5.2.4. Ground Water Conditions**

The stability analyses for the WRS assume that the natural ground is saturated and the waste rock is fully drained. The WRS will be comprised of rockfill and the gullies beneath the WRS are to be infilled with coarse rockfill to ensure good drainage. Some localised perched groundwater may occur on the thin low permeability trafficked layers within the WRS (refer Section 5.2.2), but due to the 10 to 20m vertical spacing between these layers is unlikely to significantly affect the overall stability of the WRS.

Similarly, in the existing Deepdell South Pit, the rockfill material in the pit is assumed to be drained and the natural ground is assumed to be saturated.

# **5.2.5. Static Stability**

The results of the static stability analyses are presented in Appendix C and are summarised in Table 5.1 below. Static stability analyses considered potential failure conditions using circular and block type failure surfaces which passed through the rockfill material and/or original ground foundation. Only the critical failure slip surfaces are presented in this report.

Two key variables in the setup of the stability analysis were the inclusion of the overlying loess soil between the rockfill and the foundation rock around Cross Section A-A' and the potential for unfavourable foliations in the schist rock (i.e. in the direction of slope movement). Table 5.1 summarises which cross sections these conditions are applied to.

Cross Sections A-A' and B-B' (refer to Figure 4 for plan and Appendix C for cross sections) check the stability of the Deepdell East WRS on Horse Flat.

Cross Section A-A' (refer to Figure A05 in Appendix C) is through the greatest thickness of rock fill where the deepest overlying soil is found. For stability analysis, the overlying soils in the area of Section A-A' are assumed to be continuous and 4m thick and a foliation dip of zero degrees. In reality, test pitting has demonstrated the thickness of overlying soils is much less than 4m on average and this assumption is just for the stability analyses.

Cross Section B-B' cuts through the steepest topography with a foliation dip direction that is more unfavourable than Cross Section A-A'. A downslope dip of 15 degrees has been applied.

Cross Sections C-C' and D-D' (refer to Figure 4 for plan and Appendix C for cross sections) have been used to check the stability of the Deepdell South Pit area with the addition of the East WRS.

For Cross Section D-D' the dip direction of the foliations is assumed to be at zero degrees. For Cross Section C-C' it is assumed the dip direction of the foliations is at 10 degrees, as the dip of 25 degrees noted in this area is not directly downslope.

In the model, we have applied the reduced strength along the foliations to plus and minus 5 degrees from that indicated in Table 5.1.

<b>Figure</b>	<b>Cross</b> <b>Section</b>	Over lying loess soil	Unfavourable foliation dip considered	<b>Critical Failure Surface</b>	<b>FoS</b>
A <sub>01a</sub>	$A-A'$	Yes	0 degrees	Block slide along schist foliation	2.0
A02a	$B-B'$	N <sub>0</sub>	15 degrees	Block slide along schist foliation	2.1
A03b	$C-C$	N <sub>0</sub>	10 degrees	Block slide along schist foliation	1.7
A04a	$D-D'$	N <sub>0</sub>	0 degrees	Block slide along schist foliation	1.8

**Table 5.1. Summary of Static Slope Stability Analyses**

Based on the above analyses, the performance of the WRS under static loading is satisfactory, as all the calculated FoS are above 1.5, a typical minimum value applied for long-term static stability and considered suitable for the WRS.

# **5.2.6. Seismic Stability**

Seismic stability analyses of the WRS have been undertaken for the following two levels of earthquake shaking;

- Operating Basis Earthquake (OBE) 150 year return period
- Safety Evaluation Earthquake (SEE) 2,500 year return period.

Note that the SEE was previously referred to as the Maximum Design Earthquake (MDE) but has been changed to SEE to follow the terminology used in the latest NZSOLD New Zealand Dam Safety Guidelines (2015). The WRS is not a dam, however, and therefore the NZSOLD definitions are used only as reference. For the WRS the OBE earthquake is a design limit state which aims to have minor damage and the SEE earthquake would look to prevent collapse so not to cause a hazard.

The cross-section geometry used for static stability has been used to assess seismic stability.

Peak horizontal ground accelerations for the OBE and SEE were obtained from acceleration response spectra determined in a site specific seismic hazard study by the Institute of Geological and Nuclear Science (GNS) for the MGP (Ref.7). The spectra were generated from a model which combines earthquakes associated with the three closest faults to the MGP (Billy's Ridge, Taieri Ridge, and Hyde faults).

The levels of peak ground acceleration used for analysis of the seismic stability are:

- OBE (150 yr) =  $0.13 \text{ g}$
- SEE (2,500 yr) =  $0.65$  g

The ground motion amplification (ratio of crest acceleration to PGA) relationship given by Harder et al. (Ref.8) has been used to determine the peak motion at the crest of the WRS. This method is based on actual measurements of ground motions recorded at the crests of embankments relative to those recorded near the base. Crest accelerations

using this method are 3.3 and 1.4 times the base ground acceleration for the OBE and SEE. Accelerations will vary throughout the WRS and this needs to be accounted for in the stability analyses. The estimated average horizontal accelerations for both OBE and SEE levels of earthquake shaking applied to each potential failure surface (kmax) are given in Table 5.2.

Stability has been assessed for potential failure surfaces located at 1/3H, 2/3H and 1H below the top of the WRS, where H is the full height of the WRS. Where yielding is predicted, permanent co-seismic (during an earthquake) slope deformations are estimated using the Bray and Travasarou (2007) displacement calculation (Ref.9). Spectral accelerations used for this calculation are reported in Table 5.2. The development of the calculation approach considered the dynamic response of the potential failure sliding mass.

The overlying loess soils around Cross Section A-A' are potentially susceptible to liquefaction if saturated. The test pit logs and the laboratory testing indicate that the loess layers are low-plastic silt with PI less than 12 and therefore are susceptible to liquefaction under the PI criteria of Bray and Sancio (2006) (Ref.13). We have not been able to use the water content over liquid limit criteria as the ground water conditions are currently dry in these layers.

As a perched ground water table beneath the WRS could potentially occur and saturate the loess soils, triggering of liquefaction from shaking equivalent to OBE and SEE, using Idriss and Boulanger (2008) (Ref.14), has been considered. A 35% or greater fines content has been assumed for the loess material. The cyclic stress ratio for the OBE earthquake beneath the WRS indicates that liquefaction is unlikely for the OBE earthquake, however, if the material is susceptible to liquefaction, triggering would have occurred with shaking less than the SEE. The stability analyses have, therefore, used non-liquefied strengths for the loess under the OBE and an undrained liquefied strength, based Olson and Stark (2002) (Ref.15), of 0.06 multiplied by the vertical effective stress for the SEE.

The limit equilibrium stability analyses are presented in Appendix C and summarised in Table 5.2 on the following page.

For the OBE cases, the seismic Factor of Safety is generally 1.0 or greater except for Figure A07 and A19 where small co-seismic displacements occur are estimated. For OBE cases with FoS greater than 1, limited horizontal displacements are expected, however, some vertical consolidation of the WRS, due to shake down, may occur.

With liquefaction of a continuous loess layer beneath the WRS, stability would not be maintained in the post-earthquake (SEE) case as shown in the analysis in Figure A05 in Appendix C with a FoS equal to 0.8. This is predominantly a consideration at the northern toe which the thickest covering of loess. The width of loess that would need to be removed, or waste rock contact directly on the bedrock that would need to be proven, has been determined to be 30m. A 30m wide shear key starting at the WRS toe achieves a post-earthquake FoS of 1.1. This width has been shown as a shear key area in Figure 8. This extends over the Horse Flat area however in many locations to the South East rock is outcropping anyway.

Cross Section A-A' has the greatest co-seismic displacements in the analyses with 12 to 46cm (16%ile to 84%ile) of displacement estimated. Of the other cross-sections, B- B' had the greatest estimated displacements of 11 to 41cm. For these to section the greatest displacement is associated with a shallow slip mechanism along the embankment slope face. With the higher shaking of the SEE, greater vertical shake down (consolidation) settlement of the WRS can be expected. Settlement associated with shake down may be in the order of a tens of centimetres to possibly a meter. In the SEE, these displacements are acceptable as there are no critical elements which would be affected, and the post-earthquake stability has been shown to achieve a FoS of 1.1.

To achieve a stable profile post-earthquake, the extent of loess is to be determined and material removed to match the design assumptions. This will require monitoring at the construction stage.

# **Table 5.2. Summary of Seismic Slope Stability Analyses and Co-seismic Displacement Estimates**



1. H is the total height of the slip mass. Slide mass fundamental period (T) is estimated using 4H/Vs for deep block failure slips and 2.6H/Vs for shallow circular failure slips.

2. The amplified spectral accelerations at the degraded slide period are applied as pseudostatic horizontal acceleration in seismic stability analysis.

3. Co-seismic displacement calculation is only undertaken when seismic FoS, using full pseudostatic horizontal coefficient equal to SA(1.5T)xAmp, is less than 1. Co-seismic displacements are calculated for a mean magnitude of 7.2.

# **5.3. Surface Drainage**

A perimeter drain or bund will be constructed around the toe of the WRS to collect stormwater runoff and divert it into silt ponds. Runoff flows are expected to be quite small because a high proportion of rainfall is expected to infiltrate the rock fill. This is consistent with what has been observed on site for the existing WRS.

Where necessary, perimeter drains are to be constructed largely by excavating into the natural ground. Some short sections may be located in fill. Drains will be sized to have sufficient capacity to carry the peak runoff from the 10-minute, 5 percent AEP (20 year storm) whilst retaining 0.25m freeboard.

Temporary clean water diversion drains and diversion culvert will be constructed at the early stages of the WRS to reduce the catchments contributing to the silt ponds.

# **5.4. Subsurface Drainage**

Existing ephemeral gullies beneath the WRS footprint are to be filled with coarse free draining waste rock material either through high tip-head segregation or direct placement. This will enable subsurface drainage of gullies which are filled downstream by waste rock.

# **5.5. Silt Control**

Runoff from the WRS during construction will be directed to silt ponds located in the gullies immediately downstream, as discussed in the erosion and sediment control report (Ref.10). Temporary silt ponds may also be constructed upstream of the main silt ponds in the early stages of development of the WRS as required. The WRS will be constructed with the working surface sloping down away from the outside shoulder. The runoff will then infiltrate the rockfill and percolate through the coarse fill subsurface drainage system before discharging as seepage downstream of the WRS. Experience to date indicates that not much silt is generated during waste rock stack construction due to a combination of the material used and the progressive nature of stripping and rehabilitation as the waste rock stack is constructed. Most of the runoff infiltrates the rockfill and the silt is removed before the seepage emerges from the toe of the rock stack.

# **5.6. Rehabilitation**

The final contoured surface of the WRS is to be rehabilitated by spreading 1.65m of weathered rock plus 0.2m of topsoil, excavated from the foundations, and then grassed. Once the grass is established, any runoff from the WRS is generally of good quality.

# **6.0 CONSTRUCTION AND QUALITY CONTROL**

Construction of the WRS will be undertaken by OceanaGold, or in part by contractors under the direct supervision of OceanaGold employees. OceanaGold is responsible for setting out the works, ensuring that the rock stack is constructed to the design profile, that foundation stripping and preparation is properly carried out, subsurface drainage material is suitably placed, surface drainage is properly constructed and maintained, and that rehabilitation (i.e. topsoil and grassing) is to high standards. The proposed construction methods and rehabilitation strategies are similar to those employed on the existing tailing storage facilities and waste rock stacks, and these have been successful during the 25 years of operation at the MGP.

The design requirement for shear keys beneath WRS can be reviewed with additional ground investigation information. Shear keys would be constructed of waste rock and would need to extend through the soils and contact the schist rock over the areas shown in Figure 8.

It is anticipated that similar consent conditions to those for the existing Coronation North WRS will apply to the Deepdell East WRS, such as the specific requirements affecting construction as summarised below:

Conditions 8 and 9 of RM16.138.01, Condition 8 of RM16.138.15 and Condition 12 of RM16.138.20 refer to the requirement for cleaning of construction plant to avoid spreading didymo and the need to minimise work in waterways.

Condition 14a of RM16.138.01, condition 15a of RM16.138.15, condition 16a of RM16.138.20 and condition 14.6a of 201.2016.779, 201.2013.360.1, LUC-2016-234 and LUC-2013-225A requires if there is discovery of koiwi tangata (human skeletal remains) or Maori artefact material that the Consent Holder shall without delay:

- i. Notify the Consent Authority, Tangata whenua and New Zealand Historical Society, in the case of skeletal remains, the New Zealand Police and
- ii. Stop work within the immediate vicinity
- iii. Any koiwi tangata discovered shall be handled and removed by tribal elders responsible for the tikanga (custom) appropriate to its removal or preservation

Condition 14b of RM16.138.01, Condition 15b of RM16.138.15, Condition 16b of RM16.138.20 and Condition 14.6b of 201.2016.779, 201.2013.360.1, LUC-2016-234 and LUC-2013-225A requires if there is discovery of any features or archaeological features that pre-dates 1900, or heritage material, or disturbs a previously unidentified archaeological or history site that the Permit Holder shall without delay:

- i. Stop work within the immediate vicinity
- ii. Advise the New Zealand Historical Places Trust, and in the case of Maori features or materials, the Tangata whenua, and if required, shall make an application for an Archaeological Authority pursuant to the Historic Places Act 1993
- iii. Arrange for a suitably qualified archaeologist to undertake a survey of the site.

# **7.0 CONCLUSIONS**

The Deepdell East WRS is designed in accordance with accepted engineering practices. Existing WRS have been designed to similar standards and their performance to date has been satisfactory. Construction procedures, including supervision and quality control practices for the Deepdell WRS will meet accepted engineering standards.

All final slopes of the WRS have been designed for a long term static factor of safety against instability exceeding 1.5 for the expected water levels.

The WRS has been designed for an operating basis earthquake (OBE) with a recurrence interval of 150 years and maximum design earthquake (SEE) with a recurrence interval of 2,500 years. Minor deformation is expected from the OBE and satisfactory performance is shown for the post-earthquake case and the SEE and settlement and slope deformation will not affect any critical elements of the WRS.

Based on the results of the static and seismic stability analyses conducted it is concluded that the WRS is sufficiently stable for the intended long-term use of pastoral farming postrehabilitation and mine closure.

OceanaGold will ensure short and long term stability of the WRS, associated works, and surrounds at all time during the operational life of the structure. This will be achieved through construction, rehabilitation and ongoing monitoring in accordance with the controlling documents:

Deepdell East WRS Design Report (*i.e.* this report); Deepdell North Stage III Erosion and Sediment Control Report (Ref 10); Macraes Water Quality Management Plan (Ref.11)

**ENGINEERING GEOLOGY LTD** Report prepared by

4. Torvelan

E P Torvelainen Senior Geotechnical Engineer BE (Hons) Civil MEngNZ

Reviewed by: **IX** 

Principal Geotechnical Engineer CPEng



N Tan Geotechnical Engineer BE (Hons) Civil

**FIGURES 1 TO 8**

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Source: NZMS Sheet 15 Waitaki.



**Engineering Geology Ltd**<br> **2** Esmonde Rd, PO Box 33-426, Takapuna<br>
Ph (09)486-2546 Fax (09)486-2556

**OCEANAGOLDLTD**OCEANA GOLD (NEW ZEALAND) LTD**craesocality Locality Plan** Macraes Gold Project

**Fig ure 1**

Ref. No.: Date: 26June2002Drawn: File: 1410 SPlo c al.grf











# Figure 5





Original Ground Surface

Original Ground Surface





*Fieldwork Location Plan*







# **APPENDIX A**

# **GROUND INVESTIGATIONS**



# **TEST PIT LOG**

# **TESTPIT No.: TP1**

**Job No.:** 8528  $\vert$ SHEET 1 OF 1

**PROJECT:** Deepdell East WRS

**LOCATION:** Deepdell East WRS, Oceana Gold Macraes Mine, Macraes Flat, Otago









# **TEST PIT LOG**

# **TESTPIT No.: TP2**

**Job No.:** 8528  $\vert$ SHEET 1 OF 1

**PROJECT:** Deepdell East WRS

**LOCATION:** Deepdell East WRS, Oceana Gold Macraes Mine, Macraes Flat, Otago



EOH: 0.80 m







**TESTPIT No.: TP3 TEST PIT ENGINEERING GEOLOGY LTD** SHEET 1 OF 1 **LOG Job No.:** 8528 **PROJECT:** Deepdell East WRS **LOCATION:** Deepdell East WRS, Oceana Gold Macraes Mine, Macraes Flat, Otago **COORDINATES:** East 71003.1 North 17547.6 **RL GROUND:** 473.2m **DATE:** 27/06/2019 **GRID:** MMG **DATUM:** Mine **TESTPIT DEPTH:** 2.0m **WATER CONTENT CORRECTED VANE SHEAR STRENGTH** GEOLOGICAL UNIT **GEOLOGICAL UNIT CONSISTENCY /** WATER LEVEL **WATER LEVEL GRAPHIC LOG (kPa) / RL DEPTH (m) MOISTURE CONDITION DENSITY SOIL MATERIAL DESCRIPTION FIELD TESTS SAMPLES (%)** Field Vane (BS 1377) **DEPTH**  Remoulded Field Vane 50 100 150  $473.2$ Organic SILT, minor clay; grey brown. Very stiff, moist, low plasticity TS Š, *0.2* VSt 473.0 SILT, some clay, gravelly (f-c); orange brown, light grey. Hard, mist, low plasticity *0.4* -<br>clasts of schist up to 50mm, Bag sample 1, 0.45m 472.8 SV: 0.5m, 200+ kPa Colluvium/Slopewash Colluvium/Slopewash *0.7* SV: 0.7m, 472.5 Bag sample 2, 0.7m 200+ kPa ò M H 1 *1.3* Completely weathered, orange brown, SCHIST, extremely  $471.9$  $S$ V: 1.4m, weak; weathered to a gravelly SILT, hard; Bag sample 3, 1.4m 2.00m, 27/06/2019 Terrane TZII<br>Schist 2.00m, 27/06/2019 200+ kPa *1.5* Moderately weathered, grey, orange brown, SCHIST; very weak 471.7 VW *1.8* j \_<br>「sl. weathered, weak, Foliation 144°/23° NE  $471.4$ S W Joints 260°/80° S, 131°/80° NE, tight, smooth, discontinuous, saturated, seepage at 2.0m *2.0* 2 EOH: 2.00 m







 $\overline{\phantom{a}}$ 



**TESTPIT No.: TP5 TEST PIT ENGINEERING GEOLOGY LTD** SHEET 1 OF 1 **LOG Job No.:** 8528 **PROJECT:** Deepdell East WRS **LOCATION:** Deepdell East WRS, Oceana Gold Macraes Mine, Macraes Flat, Otago **COORDINATES:** East 70989.3 North 17232.4 **RL GROUND:** 492.9m **DATE:** 28/06/2019 **GRID:** MMG **DATUM:** Mine **TESTPIT DEPTH:** 1.2m **WATER CONTENT CORRECTED VANE SHEAR STRENGTH** GEOLOGICAL UNIT **GEOLOGICAL UNIT CONSISTENCY / WATER LEVEL WATER LEVEL GRAPHIC LOG (kPa) / RL DEPTH (m) MOISTURE<br>CONDITION<br>CONSISTEN<br>CONSITY<br>SAMPLES<br>SAMATER CO<br>V<sup>(%)</sup> SOIL MATERIAL DESCRIPTION FIELD TESTS** Field Vane (BS 1377) **DEPTH**  ORemoulded Field Vane 50 100 150 TS Organic SILT, tr. clay; grey brown. V. stiff, moist, low plasticity 492.9 *0.1* SILT, minor clay; light grey orange brown, light grey, orange. Very stiff, moist, low plasticity 492.8 VSt ۰ SV: 0.2m, 145 / - kPa (N/A) j M Loess *0.3* Groundwater Not Encountered Groundwater Not Encountered  $\mathsf{I}_{\mathsf{SV: 0.3m,}}$ -<br>trace gravel (f), light grey, orange brown, hard, Bag sample 1, 492.6 0.3m H 200+ kPa *0.5* SV: 0.5m, UTP  $\frac{1}{492}$ Completely to highly weathered, light grey, orange brown, SCHIST; extremely to very weak W-V Rakaia Terrane TZIII<br>Schist *0.7* .<br>highly weathered, very weak 492. VW D-M *0.9* 492.0 moderately to slightly weathered, weak Foliation 135°/25° NE 1 W Joints 285°/70° S, 242°/80° SSE, tight, smooth, discontinuous *1.2*

EOH: 1.20 m







2

D-M

VW

 W

EOH: 2.00 m

₹

*1.8* 504.6 *2.0*

moderately to slightly weathered, weak, Foliation 166°/23° E Joints 259°/80° S, 239°/85° SSE, 126°/80° NE, tight, smooth, discontinuous














### **TESTPIT No.: TP8**

**Job No.:** 8528  $\vert$ SHEET 1 OF 1

**PROJECT:** Deepdell East WRS

**LOCATION:** Deepdell East WRS, Oceana Gold Macraes Mine, Macraes Flat, Otago



EOH: 0.90 m







### **TESTPIT No.: TP9**

**Job No.:** 8528  $\vert$ SHEET 1 OF 1

**PROJECT:** Deepdell East WRS









#### **TESTPIT No.: TP10**

**Job No.:** 8528  $\vert$ SHEET 1 OF 1

**PROJECT:** Deepdell East WRS









#### **TESTPIT No.: TP11**

**Job No.:** 8528  $\vert$ SHEET 1 OF 1

**PROJECT:** Deepdell East WRS









#### **TESTPIT No.: TP12**

**Job No.:** 8528  $\vert$ SHEET 1 OF 1

**PROJECT:** Deepdell East WRS







#### **APPENDIX B**

### **LABORATORY TESTING**



# **Central Testing Services**

18 Ngapara St, P.O. Box 397, Alexandra 9340, Central Otago, New Zealand P: 03 4487644, W: www.centraltesting.co.nz, E: info@centraltesting.co.nz

Page 1 of 3 Pages Reference No: 19/2051 Date: 15 July 2019

# **TEST REPORT - OCEANA GOLD INVESTIGATIONS**



**Checked By:** 

emplus

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Date: 15 July 2019

## **TEST REPORT-OCEANA GOLD INVESTIGATIONS**





<u>WATER CONTENT &amp; FLASTICITY INDEX RESULTS - NZS 4402.1900, Test 2.1, 2.2, 2.3 &amp; 2.4</u>	
Water Content: ("All In" As Received)	$20.1\%$
Liquid Limit: (LL)	45
Plastic Limit: (PL)	
<b>Plasticity Index: (PI)</b>	
Note: The sample was received in a natural state. The plasticity index material tested was the fraction passing the 125 um test sieve	

Note:

Information contained in this report which is Not IANZ Accredited relates to the sample descriptions based on NZ Geotechnical Society Guidelines 2005, the sample method \* and sampling.  $\bullet$ 

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**Tested By:** 

L.T. Smith

Date: 10 to 12-Jul-19

**Checked By:** 

emplus

**Approved Signatory** 

a

A.P. Julius **Laboratory Manager** 

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accreditation

## **APPENDIX C**

## **STABILITY ANALYSES**

**Static Stability Analyses**



















Name: Rockfill Model: Shear/Normal Fn. Unit Weight: 21.5 kN/m<sup>3</sup> Strength Function: Waste rock stack (Peak) Phi-B: 0 ° Piezometric Line: 1

Method: Janbu Slip Surface Option: Block

1.813



Name: Rockfill Model: Shear/Normal Fn. Unit Weight: 21.5 kN/m<sup>3</sup> Strength Function: Waste rock stack (Peak) Phi-B: 0 ° Piezometric Line: 1

Method: Spencer Slip Surface Option: Entry and Exit

2.165



**Seismic Stability Analyses**









Name: Rockfill Model: Shear/Normal Fn. Unit Weight: 21.5 kN/m<sup>3</sup> Strength Function: Waste rock stack (Peak) Piezometric Line: 1

Horz Seismic Coef.: 0.068 Ignore seismic load in strength: No

Method: Janbu Slip Surface Option: Block Name: Loess Model: Mohr-Coulomb Unit Weight: 18 kN/m<sup>3</sup> Cohesion': 0 kPa Phi': 30 ° Phi-B: 0 ° Piezometric Line: 1

1.708






Name: Rockfill Model: Shear/Normal Fn. Unit Weight: 21.5 kN/m<sup>3</sup> Strength Function: Waste rock stack (Peak) Piezometric Line: 1

Horz Seismic Coef.: 0.12 Ignore seismic load in strength: No

Method: Janbu Slip Surface Option: Block Name: Loess Liquefied Model: S=f(overburden) Unit Weight: 18 kN/m<sup>3</sup> Tau/Sigma Ratio: 0.06 Minimum Strength: 0 Piezometric Line: 1

0.999













Name: Rockfill Model: Shear/Normal Fn. Unit Weight: 21.5 kN/m<sup>3</sup> Strength Function: Waste rock stack (Peak) Phi-B:  $0^{\circ}$ Piezometric Line: 1

Horz Seismic Coef : 0.14 Ignore seismic load in strength: No

0.998 Method: Janbu Slip Surface Option: Block



Deepdell East WRS Section B-B' Name: Figure A18 SLOPE/W Analysis - SEE 1H Block Slide

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Name: Schist (Foliation 15 deg. dip) Model: Anisotropic Fn. Unit Weight: 23.5 kN/m<sup>3</sup> Cohesion': 50 kPa Phi': 40 ° Phi-Anisotropic Strength Fn.: Schist Foliation Phi Func. C-Anisotropic Strength Fn.: Schist Foliation C Func. Phi-B: 0 ° Piezometric Line: 1





1.411 Deepdell East WRS Section C-C' Name: Figure A21 SLOPE/W Analysis - OBE 1H Block Slide Engineering Geology Limited Horz Seismic Coef : 0.054 Ignore seismic load in strength: No Method: Janbu Slip Surface Option: Block Name: Rockfill Model: Shear/Normal Fn. Unit Weight: 21.5 kN/m<sup>3</sup> Strength Function: Waste Rock Stack (Peak) Piezometric Line: 1 Name: Schist (Foliation 10 deg. dip) Model: Anisotropic Fn. Unit Weight: 23.5 kN/m<sup>3</sup> Cohesion': 50 kPa Phi': 40 ° Phi-Anisotropic Strength Fn.: Schist Foliation Phi Func. C-Anisotropic Strength Fn.: Schist Foliation C Func. Piezometric Line: 1 Distance (m) 200 300 400 500 600 700 800 900 1,000 1,100 1,200 1,300 1,400 1,500 Elevation (m)  $\frac{200}{200}$ 300 400 500 600





0.997 Deepdell East WRS Section C-C' Name: Figure A24 SLOPE/W Analysis - SEE 1H Block Slide Engineering Geology Limited Horz Seismic Coef : 0.17 Ignore seismic load in strength: No Method: Janbu Slip Surface Option: Block Name: Rockfill Model: Shear/Normal Fn. Unit Weight: 21.5 kN/m<sup>3</sup> Strength Function: Waste Rock Stack (Peak) Piezometric Line: 1 Name: Schist (Foliation 10 deg. dip) Model: Anisotropic Fn. Unit Weight: 23.5 kN/m<sup>3</sup> Cohesion': 50 kPa Phi': 40 ° Phi-Anisotropic Strength Fn.: Schist Foliation Phi Func. C-Anisotropic Strength Fn.: Schist Foliation C Func. Piezometric Line: 1 Distance (m)  $\begin{array}{|c|c|c|c|c|}\hline \text{C}} \end{array}$ <br>
200 300 300 400 500 600 700 800 900 1,100 1,100 1,200 1,300 1,400 1,500<br>
200 300 300 400 500 600 700 800 900 1,000 1,100 1,200 1,300 1,400 1,500  $\frac{200}{200}$ 300 400 500 600





Name: Rockfill Model: Shear/Normal Fn. Unit Weight: 21.5 kN/m<sup>3</sup> Strength Function: Waste rock stack (Peak) Phi-B: 0 ° Piezometric Line: 1

Horz Seismic Coef.: 0.043 Ignore seismic load in strength: No

Method: Janbu Slip Surface Option: Block

1.463







Name: Rockfill Model: Shear/Normal Fn. Unit Weight: 21.5 kN/m<sup>3</sup> Strength Function: Waste rock stack (Peak) Phi-B: 0 ° Piezometric Line: 1

Horz Seismic Coef.: 0.15 Ignore seismic load in strength: No

Method: Janbu Slip Surface Option: Block

1.001

