

MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT TAILINGS STORAGE FACILITY

<u>UPDATED DESIGN REPORT</u> (REF. NO. 1627/2)

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MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT TAILINGS STORAGE FACILITY

UPDATED DESIGN REPORT (REF. NO. 1627/2)

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Suite 1400
750 West Pender Street
Vancouver, British Columbia
Canada V6C 2T8
Telephone (604) 685-0543
Telefax (604) 685-0147
CIS: 72360,477

Knight Piésold Ltd.
CONSULTING ENGINEERS



MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT TAILINGS STORAGE FACILITY

<u>UPDATED DESIGN REPORT</u> (REF. NO. 1627/2)

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MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

TAILINGS STORAGE FACILITY UPDATED DESIGN REPORT (REF. NO. 1627/2)

SECTION 1.0 - INTRODUCTION

1.1 PROJECT DESCRIPTION

The Mount Polley Project is located in central British Columbia, approximately 56 kilometres north-east of Williams Lake, as shown on Figure 1.1. The nearest settlement is the community of Likely, on the northern tip of Quesnel Lake.

The project derives its name from Mount Polley, a low mountain with a peak elevation of 1260 metres, approximately 300 metres above the surrounding terrain. Mount Polley is situated on a topographic ridge with Polley Lake to the east and Bootjack Lake to the southwest. The site is accessible by paved road from Williams Lake to Morehead Lake and then by gravel forestry road for the final 12 kilometres.

The Mount Polley open pit mine contains an estimated 82.3 million tonnes of copper and gold ore in three ore bodies. After loading in the pit, the ore will be hauled to the crusher where it will be crushed. The ore is then transported to the nearby concentrator where it will be processed by selective flotation to produce a copper-gold concentrate at a rate of approximately 17,808 tonnes per day (6.5 million tonnes per year). Approximately 92.6 million tonnes of waste rock will be stored immediately east of the Millsite.

The mill tailings will be discharged as a slurry into the Tailings Storage Facility which has been designed to provide environmentally secure storage of the solid waste. As the solids settle out of the slurry, process fluids are collected and recycled back to the mill for re-use in the milling process. No surface discharge of any process solution from the tailings facility is required or anticipated.



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1.2 SCOPE OF REPORT

This report is an updated version of the original Tailings Storage Facility design report (Knight Piésold Ref. No. 1625/1). This report has been prepared in response to modifications to the embankment design, review comments and concerns, and the addition of detailed geological and geotechnical information gathered during the Stage Ia/Ib construction program.

This report also contains as-built information for the Stage Ia/Ib tailings embankments, seepage collection ponds, basin liners, sediment ponds and other details.

The updated design is based on the following:

- Results of geotechnical investigations at the Tailings Storage Facility.
- Evaluation of construction materials.
- Revised water balance and initial process water supply requirements.
- Updated requirements for operations, on-going construction and for final reclamation.

The Tailings Storage Facility comprises a valley impoundment that will initially allow storage of water required for mill start-up. Tailings discharge from the embankments will commence in early June, 1997. In its final configuration, the facility will comprise a side-hill impoundment centred 5 km southeast of the ore bodies. The overall site plan is shown on Dwg. No. 1625.200.



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SECTION 2.0 - SITE CHARACTERISTICS

2.1 HYDROMETEOROLOGY

2.1.1 General

Long and short-term climate records are available for a number of locations in the area. Likely (with 6 years of record) and Horsefly (with 11 years) are located in similar terrain within 40 km of the site. The project area is subject to a relatively temperate climate with warm summers and cool winters. The precipitation is well distributed throughout the year.

The mean annual temperature at Likely, the nearest station, is 4.0° C with an extreme maximum of 33.9° C and an extreme minimum of -37° C. At Quesnel, with approximately 70 years of record, extremes are 40.6° C and -46.7° C. Frost free days in the area range from 199 at Horsefly Lake (elevation 788 m) to 244 at Barkerville (elevation 1244 m).

2.1.2 Precipitation and Evaporation

Precipitation data at the site is limited and thus mean precipitation records for climatologically similar stations in the area were used to estimate mean annual site precipitation values. The mean annual precipitation at Likely is 699.7 mm and at Barkerville (with over 70 years of record) is 1043.9 mm. Precipitation for the site is expected to fall within this range. Data for Likely, Barkerville and the site are presented in Table 2.1. A coefficient of variation of 0.16 was determined from regional values which translates to a standard deviation of 121 mm. These conditions were applied to the tailings facility and nearby areas.

A mean annual precipitation of 755 mm was determined for the Tailings Storage Facility based on the above. The waste dumps, pit areas and Millsite, (all at higher elevations) were modelled with a mean precipitation of 810 mm, a coefficient of variation of 0.16 and a standard deviation of



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130 mm. The increased precipitation value was determined by applying an orographic factor of 1.07285 to the values for the Tailings Storage Facility. The orographic factor is consistent with elevation correlations developed in previous studies. This data is summarized on Table 2.2.

An annual evaporation rate of 423 mm at the site has been assumed to be constant for all years of operation and precipitation conditions. The evaporation rate should be monitored and any changes can be incorporated in the design, as required.

2.1.3 Runoff Coefficients

The water balances include runoff coefficients based on average precipitation conditions only. The runoff coefficients are summarized below:

	Runoff Coefficient (%)				
Component Description	$_$ Dry	Average	Wet		
Unprepared Tailings Basin	20	24	29		
Tailings Beach	90	90	90		
Open Pit	45	50	55		
Millsite (Disturbed)	65	70	75		
Waste Rock Dumps	58	60	62		
Undisturbed Catchments	20	24	29		

2.1.4 Storm Events

Intensity-duration-frequency curves have been developed for the site on the basis of data obtained from the Rainfall Frequency Atlas for Canada (RFAC), and these are shown on Figure 2.1. Probable maximum precipitation values for the site have also been estimated, as shown on Table 2.3. As outlined in the RFAC the 1 and 6 hour values are not influenced by orographic factors, while the 24 hour and 10 day values are significantly





affected. A conservative orographic factor of 1.5 was used to evaluate the storm events at the higher elevations.

The 10 day PMP storm event value of 406 mm was estimated by assuming a ratio of 10 day to 1 day PMP of 2.0. This value was used in the evaluation of embankment storage requirements.

2.2 REGIONAL GEOLOGY

The Mount Polley site is located in an alkalic intrusive complex in the Quesnel Trough, a 35 km wide north-west trending volcanic sedimentary belt of regional extent.

The rock units are segmented into blocks by several faults, including an inferred north westerly trending normal fault that extends along Polley Lake. The predominant structure of the region is northwest trending and dipping steeply to the northeast.

The topography is generally subdued and the area has been glaciated. Surficial deposits of well graded dense glacial till are common throughout the region and are typically present in greater thickness in topographic lows. Bedrock exposures are common at higher elevations.

Detailed descriptions of bedrock and overburden geology are presented in Section 5.0.

2.3 SEISMICITY

2.3.1 Regional Seismicity

Mount Polley is situated within the interior of B.C., an area of historically low seismicity. The site is located within the Northern B.C. (NBC) source zone, close to the boundary with the Southeastern B.C. (SBC) source zone, as defined by Basham et al (1982). Basham assigns a maximum earthquake magnitude of 5.0 for the NBC zone, which is one-half magnitude higher than the observed maximum magnitude of 4.5. Similarly, a maximum



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magnitude of 6.5 has been set for the SBC zone, based on historic earthquake data.

There has been much debate in recent years concerning the possibility of a large interplate earthquake of magnitude 8 or 9 along the Cascadia subduction zone. However, such an event would be located at over 400 km west of the project and therefore is unlikely to have a significant impact at the site. Southwest of the site lies the Northern Cascades region where a maximum earthquake magnitude of 7.5 has been estimated, based on historic seismic records and geologic data (Leader Lake Seismic Risk Assessment). This potential source zone lies at a minimum distance of about 200 km and, as above, is therefore unlikely to have a significant impact at the site.

2.3.2 Seismic Design Parameters

A seismic hazard assessment for the site has been completed using both probabilistic and deterministic methods. Seismic ground motion parameters for both the Design Basis Earthquake (DBE) and Maximum Design Earthquake (MDE) have been determined.

The probabilistic analysis was carried out by the Pacific Geoscience Centre based on the method presented by Cornell (1968). The results are:

Return Period (Years)	100	200	475	1000
Maximum Ground Acceleration (g)	0.021	0.028	0.037	0.046
Maximum Ground Velocity (m/sec)	0.043	0.056	0.077	0.094

Four potential source zones were considered for estimation of the maximum ground acceleration at the site for the deterministic analysis. These source zones are the Northern B.C., Southeastern B.C., Northern Cascades and Cascadia Subduction Zones, described in Section 2.3.1. The results are tabulated below together with the maximum magnitude and minimum epicentral distance for each zone:



Source Zone	Maximum Magnitude	Epicentral Distance (km)	Maximum Acceleration (g)
Northern B.C.	5.0	. 0	0.13
Southeastern B.C.	6.5	40	0.13
Northern Cascades	7.5	200	0.04
Cascadia Subduction Zone	9.0	450	0.08

The Northern B.C. magnitude 5.0 earthquake corresponds to a worst case event occurring directly beneath the site with a focal depth of 20 km. Maximum accelerations were calculated using the ground motion attenuation relationship given by Idriss (1993), using the Mean +1 standard error relationship. Based on this, a Maximum Credible Earthquake (MCE) of M=6.5 causing a bedrock acceleration of 0.13 g has been assigned to the site.

Selection of appropriate design earthquakes for the Tailings Storage Facility is based on criteria given by the Canadian Dam Safety Association's "Dam Safety Guidelines for Existing Dams". These criteria are given on Table 2.4. A "LOW" consequence category has been assessed for the Tailings Storage Facility as discussed in Section 6.1.2. For closure and post-closure conditions a conservative "HIGH" consequence category has been adopted for design.

The seismic ground motions adopted and implications for design are summarized below:

The Design Basis Earthquake (DBE) for operations will be taken as the 1 in 475 year return period event. This corresponds to a maximum firm ground acceleration of 0.037 g and maximum ground velocity of 0.077 m/sec. These parameters will be used for the design of all earthwork structures. These values are also recommended for the design of all site buildings and structures,





consistent with the National Building Code of Canada. The above ground motion parameters place the site in seismic zone 0 for acceleration and zone 1 for velocity, $(Z_a < Z_v)$.

• The Maximum Design Earthquake (MDE) for closure of the Tailings Storage Facility shall conservatively be taken as 50% of the MCE. This MDE corresponds to approximately the 1 in 2500 year return period event, based on extrapolation of data from the probabilistic analysis. This event gives a maximum firm ground acceleration of 0.065 g and has been adopted for the design of the embankment for post-closure conditions.

Due to the dense nature of the overconsolidated foundation soils at the site, the amplification of seismic waves as they propagate from bedrock to the ground surface will not be significant. Case studies have shown that ground motion amplification is negligible through dense soil deposits overlying bedrock. Therefore, maximum bedrock ground motion parameters have been used for design.

June 6, 1997



SECTION 3.0 - TAILINGS CHARACTERISTICS

3.1 PROCESS DESCRIPTION

Mount Polley tailings will be produced from conventional milling of copper and gold ore. The anticipated tailings stream from the mill to the tailings storage facility will be as follows:

Solids throughput:

17,808 tpd (6.5 million tonnes per year)

Percent solids:

35 percent

Solids specific gravity:

2.78

The tailings slurry will be deposited from a series of spigots situated along the inside crest of the embankment. During initial discharge from the Stage Ib Main Embankment, tailings will be deposited into stored make-up water. A steep submerged beach with an estimated slope of 15 to 20 percent is expected to develop from the coarser tailings fraction. Finer tailings particles will be transported further before settling. The overall slope of the tailings solids is expected to be about 1.5 percent. (The tailings beach slopes have been estimated from experience at other mines and are based on the results presented in the publication "Tailings Beach Slopes" by B. H. Conlin.) The tailings beach is expected to emerge from the water after about 4 to 5 months of filling at the throughput rate of 17,808 tpd and assuming an initial dry density of 1.1 tonnes/cubic metre.

After the tailings beach has emerged, a sandy beach will develop as the coarser tailings fraction settles more rapidly adjacent to the embankment. The average beach slope above water will be about 1.5 percent. The finer tailings particles will be transported further out into the supernatant pond before settling at a minimum anticipated slope of about 0.25 percent. Overall, the tailings solids are assumed to have an average slope of about 0.5 percent.

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3.2 PHYSICAL CHARACTERISTICS

Coastech Research Inc. conducted preliminary metallurgical testwork on samples of drill core in 1989. A laboratory testing program was conducted on tailings samples obtained from this work. Selected information is summarized below.

- The tailings are comprised predominantly of silt (64 percent) and fine sand (30 percent) with a trace of clay (6 percent). It is non-plastic, yellow grey in colour and has a solids specific gravity of 2.78.
- A series of settling tests were completed at slurry solids contents ranging from 25 to 45 percent. The tailings particles settled rapidly and a pronounced segregation of coarse to fine material was observed. The colloidal clay fraction remained suspended in the supernatant water for several days.
- The tailings initially settled to relatively low dry densities generally in the range of 0.9 to 1.1 tonnes/m³. Consolidation caused by evaporative drying resulted in final dry densities of approximately 1.3 tonnes/m³.
- The initial volume of water recovered from the tailings depends on the initial solids content of the slurry. At 35 percent solids, the initial water recovery was about 64 percent of the total water in the slurry.
- The vertical permeability of the settled tailings varied from 1.0 x 10⁻⁵ and 2.0 x 10⁻⁵ cm/s. The horizontal permeability is expected to be significantly greater due to segregation of the soil particles. In practice, the permeability of deposited tailings will be reduced due to on-going consolidation.
- Particle settling velocities were measured as part of the hydrometer test on the silt and clay sized fraction. The data are used for calculating friction losses in slurry pipelines.

Detailed results of this testwork was presented in the "Tailings Storage Facility Design Report" (Knight Piésold Ref. No. 1625/1). More recently, MET Engineers Ltd. as





part of the overall metallurgical testing for the project conducted additional testwork on the tailings. The results of this testwork are summarized below.

- Tailings will be separated into two streams at the mill; the finer Slime Tails and the coarser Sand Tails.
- The Slime tails will make up about 57 percent of the tailings stream. The Sand Tails will make up the remaining 43 percent.
- The Slime Tails are comprised of about 85 to 90 percent well graded silt. The remaining 10 to 15 percent is comprised of clay sized particles (estimated from gradation limits).
- The Sand Tails are comprised of about 26 percent fine sand and about 70 to
 74 percent coarse silt (estimated from gradation limits).
- The Bulk Tailings, estimated by combining the Slime and Sand Tails, comprise about 13 percent fine sand, 77 to 82 percent silt and 5 to 10 percent clay sized particles (estimated from gradation limits).

The gradation limits of the Slime, Sand and Bulk tailings are shown on Figure 3.1.

3.3 GEOCHEMICAL CHARACTERISTICS

Geochemical testwork on a locked cycle tailings sample was conducted in 1989 by Coastech Research Inc. The testwork included the following:

- Determination of net acid generating potential
- Special Waste Test using acetic acid
- ASTM waste extraction test using carbonic acid

The acid base accounting procedures used were based on recommendations by the U.S. Environmental Protection Agency. The method includes an evaluation of the balance between acid producing components (primarily pyrite) and acid consuming



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components (carbonates and other rock types with neutralizing capabilities). The results of this testwork are as follows:

Sulphur (percent)	Paste pH	Acid Potential (kg CaCO ₃ /t)	Neutralization Potential (kg CaCO ₃ /t)	Net Neutralization Potential (kg CaCO ₃ /t)
0.02	8.22	0.6	24.6	24.0

These results indicate that the tailings are not acid producing and have a significant net neutralization potential.

A special waste classification test was conducted in accordance with the procedure published by the B.C. Ministry of the Environment, entitled "B.C. Special Waste List". The test indicates that the tailings from the locked cycle tests do not exceed the B.C. Waste Management Branch regulations for special wastes.

In addition to the special waste test, an ASTM waste extraction test using carbonic acid at pH 5.5 was carried out. The test uses carbonic acid for leaching of the tailings and is a more realistic indication of actual long term water leachable constituents under slightly acidic rainfall. The test showed very low levels of water leachable constituents in the extract, all at concentrations below the lower range concentration for the pollution control objectives for final effluent discharge.

Detailed results of the geochemical characteristics of the tailings were presented in the "Tailings Storage Facility Design Report" (Knight Piésold Ref. No. 1625/1).

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SECTION 4.0 - SITE SELECTION

A selection and evaluation of alternative tailings disposal sites was carried out in 1989 and included a comparative assessment of the following factors:

- Capacity and filling characteristics.
- Surface hydrology and downstream water usage.
- Hydrogeology and groundwater flows.
- Aesthetics and visual impact.
- Foundation conditions and construction requirements.
- Closure and reclamation requirements.
- Capital and operating costs.

Three possible tailings disposal sites were identified and designated as Areas A, B and C. A preliminary site investigation program was conducted at each site to evaluate the environmental impacts as well as design and construction constraints. Results of initial site investigations for each site were presented in the 1990 Knight Piésold "Report on Geotechnical Investigations and Design of Open Pit, Waste Dumps and Tailings Storage Facility" and are not repeated in this report. The three sites are:

- Area A A cross-valley impoundment located immediately south-east of Bootjack Lake, in the Bootjack Creek Valley.
- Area B A sidehill impoundment located between the south ends of Polley and Bootjack Lakes, within the upper catchment of the Edney Creek Tributary.
- Area C A cross-valley impoundment located between the north ends of Polley and Bootjack Lakes, just west of the Frypan Lake in the 6 k creek swamp.

The evaluation included environmental, design, operational and economic factors. Although Area A is closest to the mill, it was judged the least favourable because it would have required two cross-valley embankments near the outlet of Bootjack Lake and would have significantly impacted the Bootjack Creek catchment. Area C,

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located on a natural divide, would also require a cross-valley impoundment and would have impacted both the Bootjack and Polley Lake catchments. Area B was deemed the preferred site because it will provide secure tailings storage that would meet all environmental and closure requirements at the least cost. Further, it would minimize potential impacts to both Polley and Bootjack Lake catchments and would keep mine tailings and any leachate confined to the Edney Creek Tributary watershed.

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SECTION 5.0 - GEOLOGICAL AND GEOTECHNICAL CONDITIONS

5.1 GENERAL

Numerous site investigations were conducted during Stage Ia/Ib construction in 1996 and early 1997 for various reasons, including:

- Test pit excavations to identify suitable embankment fill materials in potential borrow sources. (The Original Borrow Area is located within the facility, on the east side; the Alternate Borrow Area is located downstream of the Main Embankment, above the topsoil stockpile; the Future Borrow Area is located downstream of the facility, adjacent to the east end of the Main Embankment.)
- Borehole investigations to define maximum borrow area depth in the
 Original and Future Borrow Areas.
- Groundwater Monitoring Well drilling around the perimeter of the Tailings
 Storage Facility for long term groundwater quality monitoring.
- Test pit excavations to delineate the extent of the basin liners.
- Test pit excavations to investigate potential sand deposits for use as filter sand in the embankment drains.
- Borehole investigations to accompany the installation of instrumentation in the Main Embankment foundation soils.
- Cone Penetration Test (CPT) investigations to further investigate the Main Embankment foundation soils.

The tailings area geological and geotechnical conditions have been confirmed using information collected from these investigations and from work conducted in 1989 and 1995. The investigation locations are shown in plan on Dwg. No. 1627.001



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and the interpretations are shown in section on Dwg. Nos. 1627.002 to 1627.012. A summary of the recent geological investigations is presented on Table 5.1 and summary logs are included in Appendix A. A discussion of the site investigations and the geological and geotechnical conditions at the Tailings Storage Facility is presented below.

5.2 <u>SITE INVESTIGATIONS</u>

5.2.1 Previous Investigations

Results from site investigations conducted prior to 1996 have been used in the compilation of the geological and geotechnical data. Site investigations conducted prior to 1996 are summarized below.

1989 Test Pit Investigations

Test pits were excavated throughout the Main and Perimeter Embankment footprints and within the tailings basin during the initial site investigations. Shallow subsurface foundation conditions and potential borrow sources were evaluated. Results from these investigations are presented in the Knight Piésold document "Report on Geotechnical Investigations and Design of Open Pit, Waste Dumps and Tailings Storage Facility, 1990".

1989 Diamond Drillhole Investigations

Seven diamond drillholes were completed at the Tailings Storage Facility as part of the overall site condemnation and exploration program. The investigations included tri-cone drilling of the overburden and diamond drilling in bedrock. Packer permeability testing was also conducted. Results from these investigations are also presented in the Knight Piésold document "Report on Geotechnical Investigations and Design of Open Pit, Waste Dumps and Tailings Storage Facility, 1990".



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1995 Test Pit Investigations

Numerous test pits were excavated at the proposed sites for the Main and Perimeter Embankments, tailings basin, seepage control pond, pipeline road and Millsite as part of the final design investigations. Results from these investigations are presented in the "Report on 1995 Geotechnical Investigations for Mill Site and Tailings Storage Facility" (Knight Piésold Ref. No. 1623/1) and "Response to Review Comments on Tailings Embankment Design" (Knight Piésold Ref. No. 1625/6).

5.2.2 Recent Investigations

Additional geological and geotechnical information was obtained from site investigations conducted during the 1996 and 1997 construction programs. Recent investigations include the following:

1996 Test Pit Investigations

Numerous test pits were excavated during the 1996 construction program with track mounted excavators provided by North American Construction. The test pits were excavated to evaluate shallow subsurface soil conditions for the following reasons:

- Basin Liners Test pits were excavated to determine the limits of the glacial till liner within the tailings basin. As a result of the investigations, the Upper and Lower Basin Liners were delineated immediately upstream of the Main Embankment, as shown on Figure 5.1. As part of the investigations, the thickness of the surficial glacial till was identified and the permeability of the underlying glaciolacustrine / glaciofluvial soil units was evaluated. The basin liners are shown on Sections 1, 3, 4, 6 and 10.
- Borrow Areas Test pits were excavated to evaluate glacial till borrow sources for construction of the Stage I embankments. Most of the test





pits were located in the Original Borrow Area. Additional investigations were conducted at the Alternate and Future Borrow Areas. Soil samples were tested at the on-site laboratory. The results are discussed in Section 6.4.2. The Original Borrow Area is shown on Sections 5, 6 and 7.

• <u>Filter Sand</u> – Test pits were excavated to investigate local sandy deposits. Samples were collected and laboratory tests were conducted to evaluate the suitability of the materials as filter sand for use in the embankment drains. Two areas were investigated, including the small hill at the west edge of the Reclaim Barge Channel and the ridge to the east of the Perimeter Embankment.

1996 Vibrating Wire Piezometer Installations

A total of six vibrating wire piezometers were required in the foundation soils at the Main Embankment. Two piezometers were installed in boreholes at each of the three Main Embankment monitoring planes (boreholes 96-A1, 96-B1 and 96-C1). The boreholes were advanced to a maximum depth of 16.5 metres using solid and hollow stem augers. Standard Penetration Testing (SPT) was conducted during the drilling of the each borehole, including continuous sampling for the first 10 metres. Split spoon samples were analyzed at the on-site laboratory. The borehole locations are shown on Dwg. No. 1627.001 and are included on Section 3. The drill rig was operated by Peace Drilling and Research. R.E. Graham Engineering supervised the drilling and piezometer installations. The R.E. Graham field report and test hole logs are provided in Appendix B1.

1996 CPT Program and Pressure Relief Well Installations

A CPT (cone penetration testing) program was conducted at the Main Embankment to obtain continuous profiles of soil types and information on densities and pore pressures in the foundation soils. The CPT program was also conducted to investigate low SPT N-values encountered in a saturated,



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non-plastic silt layer during the drilling for the 1996 Vibrating Wire Piezometer Installations. The CPT investigations confirmed that the low N-values were due to drilling disturbance and not weak foundation soils. After the CPT holes were completed, a second hole was drilled using a solid stem auger to confirm soil types and permit soil sampling. Four auger holes were drilled to serve as Pressure Relief Wells. The holes were hydraulically connected to the embankment foundation drains by backfilling with filter sand. The drilling was completed by Mud Bay Drilling and the CPT investigations were conducted by Cone Tec, under the direction of Knight Piésold. The results of these investigations are presented in the Cone Tec field report, included in Appendix B2. (In addition to the Pressure Relief Wells, Pressure Relief Trenches were excavated and backfilled with gravelly sand.)

1996 Groundwater Monitoring Well Installations

The 1996 groundwater monitoring well installation program was conducted in November, 1996. The purpose of the program was to drill and install groundwater monitoring wells around the Tailings Storage Facility. Wells were installed in higher permeability zones in both overburden and bedrock. The wells will be used for long-term monitoring of groundwater quality. The program comprised drilling through the overburden and into bedrock at six locations around the perimeter of the Tailings Storage Facility. Three locations around the mine site were also drilled. The drilling was completed using an air rotary rig, operated by Drillwell Enterprises. Standard Penetration Testing (SPT) and sampling was conducted within the overburden at 10 or 20 foot intervals. Bedrock chip samples (for condemnation analysis) were collected at 10 foot intervals. Details of this investigation are presented in the report "1996 Groundwater Monitoring Well Installation Program" (Knight Piésold Ref. No. 1628/4).

1997 Test Pit Investigations

The 1997 test pit investigations were completed in January 1997. The purpose of the investigations was to further delineate near surface glacial till borrow





sources for embankment from the Original and Alternate Borrow Areas and to evaluate the potential of the Future Borrow Area. The test pits were excavated to depths up to 6.7 metres. The test pits investigations were completed using a track mounted excavator owned and operated by North American Construction. Soil samples collected during the investigations were analyzed at the on-site laboratory.

1997 Borehole Investigations

The 1997 borehole investigations were also completed in January, 1997. The purpose of the investigations was to define the maximum borrow depth by identifying the maximum depth of the glacial till in the Original Borrow Area. The investigations included solid stem and hollow stem augering in 28 boreholes. Split spoon sampling was conducted and the boreholes were terminated at a maximum depth of 17 m. The drill rig was owned and operated by Geotech Drilling. Soil samples collected from the auger flights and from the split spoons were analyzed at the on-site laboratory.

The borrow area test pits and boreholes are shown in plan on Dwg. No. 1627.001 and on Sections 5, 6, 7 and 8.

5.3 SUMMARY

A geologic summary has been prepared from the above listed investigations and laboratory test data. The main purpose of the geological summary is to define the surficial overburden conditions, such as the continuity of the surficial glacial till and the location and extent of the underlying sedimentary units. In summary, the geology of the tailings basin is characterized by four units:

Surficial (Ablation) Till Unit

A surficial layer of glacial till underlies all areas of the tailings basin investigated to date. This glacial till is typically comprised of 50 to 65 percent sandy silt (passing the No. 200 sieve). The surficial till unit is believed to be a melt-out or Ablation





Till. It is commonly slightly weathered, firm to stiff, and wet over the top 0.5 to 1.0 metres in the lower areas of the tailings basin. The till is very stiff and is moist to very moist below 1.0 to 2.0 metres depth and at higher elevations. No appreciable fissuring was observed in the surficial till unit, likely due to the shallow groundwater table, which is typically less than 0.3 metres below the ground surface.

The thickness of the surficial till varies, but generally thins from north (4 to 6 metres) to south (2 to 3 metres) along the valley, as shown on Sections 1, 4, 7, 8, and 10. The surficial till unit may pinch out completely downslope (southeast) of the Main Embankment Seepage Collection Pond, as shown on Sections 1, 4, and 10. In-situ field and laboratory permeability testing on the surficial till unit typically yielded results in the order of 10^{-8} to 10^{-9} cm/s. The till thickness appears to exceed 2 metres throughout most of the tailings basin.

Areas where the surficial (Ablation) till is thinner (less than 2 metres) are shown on Figure 5.1. One design criteria for the tailings basin was that a minimum thickness of 2 metres of surficial glacial till must be present or a low permeability soil liner was to be constructed (as discussed in Section 6.2.2). Figure 5.1 indicates that the surficial (Ablation) till unit is less than 2 metres thick at two sections of the Main Embankment, including the right abutment (approx. Ch. 16+00 to Ch. 16+75) and at the bottom of the basin (approx. Ch. 19+50 to Ch. 21+50). As a result, a glacial till basin liner was required in these areas. The locations of the as-built basin liners are also shown on Figure 5.1.

Glaciolacustrine/Glaciofluvial Unit

A glaciolacustrine/glaciofluvial sedimentary unit underlies the surficial glacial till. This unit is primarily comprised of glaciolacustrine layers (silt, some clay), with lesser fine grained glaciofluvial layers (sand). The glaciolacustrine/glaciofluvial sequence thickens from west to east and from north to south and terminates at El. 928 m (approx.). It is not present along the right abutment where the surficial till directly overlies bedrock. The glaciolacustrine/glaciofluvial sequence also appears to transform from a continuous sequence near the Main Embankment into thin (0.5 to 3.0 metre) layers within the glacial till unit to the northwest. The





glaciolacustrine/ glaciofluvial sequence is generally 6 to 8 metres thick at the west and increases to as much as 25 metres thick towards the eastern edge of the tailings basin.

The glaciolacustrine/glaciofluvial sequence consists predominantly of interbedded layers of silt with either clay or fine sand. The glaciolacustrine (silt, clay) sediments are often highly over-consolidated and very stiff to hard. The glaciolacustrine unit has a low permeability as a result of the fine-grained composition. Within the glaciolacustrine sediments, occasional seams of fine sand with only a trace of silt are present. These seams vary in thickness from 0.1 metres to greater than 3 metres.

One continuous sandy unit was identified below the surficial glacial till over the 450 metre stretch (approx. Ch. 16+50 to Ch. 21+00) directly beneath or immediately upstream of the Stage Ib Main Embankment footprint, as shown on Figure 5.2 and Section 3. The sandy unit is characterized as fine-grained, containing 20 to 40 percent coarse silt. The permeability value of this unit is estimated to be 10⁻⁵ to 10⁻⁶ cm/s, based on the results of gradation analyses.

Additional test pit and borehole investigations have shown that the sandy unit terminates just downstream of the upstream toe of the Main Embankment, except for a 225 metre stretch between Ch. 16+50 and Ch. 18+75. In this area, the sandy unit grades into a fine to medium grained sand and contains localized areas of coarser gravelly sand. The permeability value of this coarser unit is estimated to be 10^{-4} to 10^{-5} cm/s. During investigations, groundwater was seeping into excavations in this material and some of the excavation walls were unstable when exposed, indicating that the unit is likely a confined aquifer. Two foundation drains, one pressure relief trench and one pressure relief well were extended into the sandy unit to help contain groundwater flows in this area. Groundwater Monitoring Well GW96-9 was also advanced into this unit just downstream of the ultimate toe of the Main Embankment.

Test pit investigations showed that the glaciofluvial sand unit extends upstream into the Tailings Storage Facility for approximately 200 metres before it grades into a



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lower permeability silt unit or is overlain by more than 2 metres of surficial glacial till, as shown on Sections 4 and 10. Laboratory testwork on samples of the glaciolacustrine/glaciofluvial materials from further upstream of the Main Embankment (such as the sediment layers exposed in some areas of the Reclaim Barge Channel) have shown that the material is primarily silt with variable clay content and occasional narrow seams of coarse silt with trace to some fine sand.

During drilling of groundwater monitoring well GW96-1 near the Perimeter Embankment Seepage Collection Pond, the surficial glacial till was found to have a minimum thickness of 5 metres, as shown on Section 8. Below this, the glaciolacustrine sequence consists of a 3 metre thick layer of firm to stiff, low permeability silts with variable clay content and thin silty sand laminations. No higher permeability layers were observed in this unit. The glaciolacustrine sequence is in turn underlain by 19 metres of glacial till. Just east of the Perimeter Embankment, in groundwater monitoring well GW96-2, the glaciolacustrine unit is 7.5 metres thick and is overlain by 11 m of glacial till. The glaciolacustrine sequence consists primarily of silt with rare thin (5 to 30 mm) fine to medium grained sand laminations.

A 10 to 13 metre thick sequence of high permeability glaciofluvial sands and gravels was encountered at depths of 27 and 32 metres in groundwater monitoring wells GW96-1 and GW96-2. This unit overlies a Basal Till unit (discussed below). The higher permeability sandy gravel unit is not connected to the tailings impoundment and is therefore not considered to be a significant potential seepage pathway due to the thick layer of low permeability surficial till and glaciolacustrine sediments above this zone.

Basal Till

The glaciolacustrine/glaciofluvial sequence is underlain by a very dense, well graded silt and sand glacial till unit believed to be a basal till. The basal till unit dips and thickens slightly from west to east and north to south, likely following bedrock topography. It is typically 10 to 20 metres thick., massive, highly





consolidated and contains some gravel and trace to some clay. The basal till has a low permeability, estimated to be less than 10^{-6} cm/s.

Bedrock

At the Main Embankment, the bedrock surface dips from west to east across and more gently from north to south, as shown on Sections 1, 3, 4, 5, 6 and 9. Bedrock drops off quickly to the west and is present less than 1 metre below surface on the ridge at the right abutment. The bedrock surface is greater than 30 m deep at the left abutment and bedrock permeabilities will not greatly influence potential seepage from the Tailings Storage Facility because of the thick cover of low permeability overburden soils.

Bedrock at the tailings basin is predominantly a red-brown sedimentary conglomerate unit composed of hematitically altered volcanic tuffs and fragmentals. It is moderately to highly weathered near the surface, but weathering decreases with depth. Results from the 1989 condemnation drilling indicated that rock quality was typically poor to fair, with the top 7 to 15 m often very poor to poor and improving with depth. The unit appeared to be free of large fault fractures and measured permeabilities were typically 10^{-6} cm/s or lower.

A coarse-grained syenite intrusive unit underlies much of the hill up-slope (west) of the Bootjack-Morehead Connector Road, as observed in groundwater monitoring wells GW96-5 and MP89-236. The unit is massive and is generally free of large fractures. Rock quality ranged from fair to good. Smaller isolated units of mudstone, sandstone and basalt were also identified.

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SECTION 6.0 - TAILINGS STORAGE FACILITY DESIGN

6.1 DESIGN BASIS AND CRITERIA

6.1.1 General

The principal objectives for the design of the Tailings Storage Facility are to ensure that regional groundwater and surface water flows are not adversely affected during mining operations and in the long term, and also to permit effective reclamation at mine closure.

The principal requirements of the design are to:

- Provide permanent, secure, and total confinement of all solid waste materials within an engineered storage facility.
- Control, collect and remove free draining liquids from the tailings for recycling as process water to the maximum practical extent.
- Include monitoring features for all aspects of the facility to ensure performance goals are achieved.
- Develop the facility in stages to distribute capital expenditure over the life of the project.

The design basis and operating criteria for the Tailings Storage Facility are based in part on appropriate and conservative design parameters from hazard classification, seismic data, hydrological studies and geotechnical site investigations and on review comments by the Ministry of Employment and Investment (MEI). The design basis and operating criteria for all aspects of the design, construction and operations are presented on Table 6.1. Provisions for water management are also included.



6.1.2 Consequence Classification

A hazard classification for the Tailings Storage Facility has been assessed to establish design flood and seismic criteria. The hazard classification is based on the Canadian Dam Safety Association's (CDSA) "Dam Safety Guidelines for Existing Dams", which states that "Tailings dams and their appurtenant structures must be protected against the same hazards and to the same extent as embankment dams...". Details of each consequence category and the corresponding potential consequences of failure are presented on Table 6.2.

A "LOW" hazard classification or consequence category has been assessed for the Tailings Storage Facility, as discussed in Section 2. This implies that the consequences of failure consist of a low economic loss and low environmental impact. In accordance with the "LOW" hazard classification, a Design Basis Earthquake (DBE) corresponding to the 1 in 475 year return period event (which corresponds to the National Building Code of Canada standard) has been adopted for design of the facility during operations. For closure and post-closure conditions, a conservative "HIGH" consequence category has been assigned.

The embankment has been designed to accommodate a maximum design earthquake (MDE) corresponding to 50% of the maximum credible earthquake (MCE) and the probable maximum flood (PMF) flood event.

6.1.3 Tailings Storage Capacity

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The tailings storage facility depth-area-capacity-filling rate relationships are presented on Figure 6.1. The projected filling rate and rate of rise for the tailings are based on an average annual production rate of 6.5 million tonnes (17,808 tpd). Figure 6.1 indicates that after approximately 3 years of operation the tailings surface area is sufficiently large that the on-going rate of rise is less than 3 metres per year.





The tailings facility has been designed to contain 84.5 million tonnes of tailings solids at an average dry density of 1.28 tonnes/m³ (1.1 tonnes/m³ for Year 1, 1.2 tonnes/m³ for Year 2, and 1.3 tonnes/m³ for Years 3 through 13) with an assumed flat tailings surface.

The filling schedule and anticipated staged construction sequence is shown on Figure 6.2. In addition to the flat tailings surface, provisions for the following have been incorporated into the design:

- Up to 2.5 million cubic metres of process (reclaim) water.
- An emergency storage volume of at least 0.68 million cubic metres for the design storm event.
- An additional one metre of freeboard for wave run-up.

The emergency storage volume of 0.68 million cubic metres will be available on the tailings surface at start-up and during on-going operations. It is required to contain the anticipated runoff from the 24-hour probable maximum precipitation (PMP) storm event centred on the tailings facility and the catchment area immediately above the facility. A minimum of one metre of freeboard will be available above the PMP runoff volume inside the facility for wave run-up and emergency flood storage. The projected runoff from a 10-day PMP event is approximately 1.36 million cubic metres, which can also be completely contained within the basin. As adequate storage capacity will always be available within the tailings basin for complete containment of the PMP events, an emergency spillway will not be required during operations.

6.1.4 Reclaim Water Storage Capacity

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The Tailings Storage Facility includes the provision to store up to 2.5 million cubic metres of reclaim water on top of the tailings surface. This water will be required prior to mill start-up and to supply the milling process during the cold winter months when surface runoff is at a minimum.





An evaluation of the extent and depth of the supernatant pond was previously completed for various tailings slopes during operations. The results indicated that there would be sufficient flexibility within the operating plan to ensure adequate storage volume on the tailings surface.

An average tailings beach slope of 0.5% has been assumed, although slopes adjacent to the upstream face of the embankment are typically in the range of 1 to 2 percent. The steeper beach adjacent to the embankment will provide a sandy zone for construction of on-going raises and for control of the phreatic surface in the embankment.

6.1.5 Staged Development

The tailings embankments have been designed for staged development during operations in order to minimize initial capital expenditures and to maintain an inherent flexibility to allow for variations in operation and production throughout the life of the mine.

The Stage Ia embankment was constructed to El. 927 metres in late 1996. The embankment was raised to Stage Ib El. 934 metres in early 1997. Stage Ib provides adequate water storage for mill start-up, plus tailings storage for approximately the first year of production along with the required freeboard. The final embankment is projected to reach El. 965 metres.

On-going requirements for embankment construction are shown on Figure 6.2 and on Dwg. No. 1625.111. Staged embankment fill quantities are presented on Table 6.3. The Stage II and III embankment raises will each provide incremental storage capacity for approximately one year of production. The Stage IV raise and each of the successive raises have been designed to provide incremental storage capacity for approximately two years of operation. Ongoing evaluation during operations will provide the basis for determining whether annual raises are more expedient for staged expansion of the facility.

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All raises after Stage Ib will require the placement of fill on tailings beaches. A coarse bearing layer of sandy gravel or rockfill will be included, as required, for the initial construction on the tailings beaches. The low rate of rise of the tailings surface and the provision of an upstream drainage system will promote drainage and consolidation in order to enhance fill placement on the tailings beaches.

On-going raises of the tailings embankments may incorporate cycloned sand extracted from the bulk tailings stream. This option will be evaluated during the initial years of operation.

It will be possible to expand the tailings storage facility if the ore reserves are increased above the currently projected 84.5 million tonnes. Embankment raises above the proposed final crest elevation of El. 965 metres would be constructed as required by incorporating a downstream extension of the embankment toe. In addition to an increased storage capacity for the facility this would also ensure that embankment stability is maintained. Detailed stability analyses must be performed in the design of all future embankment raises.

6.2 TAILINGS BASIN PREPARATION

6.2.1 Foundation Preparation

Foundation preparation requirements for the embankment footprints, seepage collection ponds, basin liners, borrow areas, topsoil stockpiles, reclaim barge channel and site roads include clearing, stripping and topsoil stockpiling. Sub-excavation and replacement may be required in soft, saturated areas of the embankment footprints.

Topsoil stripped from the embankment footprint and from the tailings basin, where required, will be stored in the topsoil stockpile, adjacent to the Alternate Borrow Area, as shown on Dwg. No. 1625.201.





6.2.2 Basin Liners

As discussed in Section 5.0, most of the tailings basin is blanketed by naturally occurring low permeability glacial till which functions as an in-situ soil liner. However, an area of thin near surface glacial till with underlying glaciofluvial sediments was identified just upstream of the Main Embankment footprint. This is where the glacial till basin liners were constructed. A basin liner was constructed where the thickness of the surficial glacial till unit was less than 2 metres and the underlying materials were estimated to have permeability values greater than 10⁻⁶ cm/s.

The extent of the basin liners was confirmed during the 1996 site investigations. The Upper and Lower Basin Liners were delineated immediately upstream of the Main Embankment, as shown on Figures 5.1 and 5.2 and on Dwg. No. 1627.201.

The imported glacial till basin liner is connected to the Main Embankment core zone and extends onto existing dense low permeability glacial till that is at least 2 meters thick. The original design requirement specified that the liner was to be placed and compacted in three 150 mm thick lifts and then covered with a 300 mm thick layer of random fill as protection from frost penetration and equipment traffic. The frost protection was later increased to 450 mm for the Lower Basin Liner. In addition, an extra 750 mm of low permeability glacial till was mounded over the backfilled investigation test pits within both basin liner areas. The upstream edge of the Lower Basin Liner was keyed into the existing surficial glacial till materials. The details of the basin liner are shown on Dwg. No. 1625.202.

The basin liner materials were tested as part of the overall QA/QC program. In general, the liner materials comprised well graded silty sand glacial till placed and compacted wet of the Standard Proctor Optimum Moisture Content. The test results on the basin liner materials will be presented in the Stage Ia/Ib construction report.

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6.2.3 Sediment Control Structures

Sediment control measures were included to prevent turbid surface runoff from impacting the environment downstream of the Tailings Storage Facility during basin stripping and early embankment construction. The Main Embankment Seepage Collection Pond was constructed first to provide primary sediment control during basin stripping and early embankment construction. Once the embankment foundation was prepared and embankment fill placement started, the Main Embankment became the primary sediment control feature.

6.3 EMBANKMENT CONSTRUCTION

6.3.1 General

The final configuration of the Tailings Storage Facility will include the Main, Perimeter and South Embankments. The Main and Perimeter Embankments are zoned earthfill structures with low permeability glacial till core zones, chimney drains, upstream drains and a downstream random fill zone. The South Embankment is a zoned earthfill water retaining structure that also includes a Chimney Drain. Future embankment construction may also incorporate waste rock from open pit development.

The tailings embankments have been designed for staged expansion during operations in order to minimize initial capital expenditures and to maintain an inherent flexibility to allow for variations in operation and production throughout the life of the mine. The embankments will be expanded using downstream and centreline construction techniques during the first two years. Further expansion of the Main and Perimeter Embankments will involve modified centreline methods, whereas the South Embankment will be constructed by downstream and centreline methods.



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6.3.2 Foundation Conditions

Geotechnical investigations confirmed that the foundation materials at the embankments generally comprise surficial glacial till overlying a dense glaciofluvial/glaciolacustrine deposit. After foundation preparation was completed (including topsoil removal) most of the foundation soils were found to be dense, low permeability glacial till. However, some localized areas of the Main Embankment footprint contained soft, saturated glacial till which was removed prior to fill placement. At the Perimeter Embankment, a layer of saturated sand and silt was also removed from a small area of the embankment foundation to expose suitable low permeability glacial till.

The prepared embankment foundations were inspected by the Engineer and written foundation approval was required before fill placement was allowed to commence.

A system of foundation drains was installed in the Main Embankment foundation to improve the foundation conditions and enhance the dewatering of near surface soils. This system was installed to permit drainage during the early stages of construction. Pressure relief wells and pressure relief trenches connected to the foundation drains were included to depressurize the underlying glaciofluvial deposits to enhance the stability of the embankment. The foundation drains are discussed in detail in Section 6.3.6. The details of the pressure relief wells and trenches are shown on Dwg. Nos. 1625.210 and 211.

6.3.3 Borrow Areas

The tailings embankments will be constructed predominantly from glacial till borrowed locally at the Tailings Storage Facility. Most of the glacial till used for Stage Ia and Ib embankment construction was taken from the Original Borrow Area which is located within the facility, on the east side. Additional material was taken from the Alternate Borrow Area, located downstream of the Main Embankment, above the topsoil stockpile and from





the Future Borrow Area, located downstream of the facility, adjacent to the east end of the Main Embankment. The borrow areas are shown on Dwg. No. 1625.201.

In some sections of the Original Borrow Area, the glacial till materials were completely removed, exposing higher permeability soils. As a result, a low permeability glacial till basin liner was constructed over the exposed areas.

Glacial till materials present at lower elevations within the Tailings Storage Facility will not be accessible once impoundment filling commences. Therefore, the Alternate and Future Borrow Areas will continue to be developed for future embankment expansions. Waste rock from open pit development may also be utilized in the downstream shell zone (Zone C).

6.3.4 Stage Ia/Ib Embankments

The Stage Ia Main Embankment construction was completed to El. 927 metres in December, 1996. Stage Ia was completed as a water retaining dam to store the 1997 freshet runoff. Approximately 85 percent of the Stage Ib Perimeter Embankment was also completed in 1996.

The Stage Ib Main and Perimeter Embankment construction was completed to El. 934 meters in March, 1997. Stage Ib will provide sufficient storage capacity to contain all mine site runoff, additional make-up water, as required, from the Polley Lake Pumping System, and for approximately one year of tailings production.

The Stage Ia/Ib Main Embankment consists of an upstream core zone (Zone S) and a downstream zone (Zone B) and a Chimney Drain. Zone S is comprised of well graded, low permeability glacial till. The specification for Zone S originally required glacial till placement and compaction in maximum 300 mm lifts to 95 percent of the Modified Proctor maximum dry density. However, after construction started, it became apparent that this compaction standard was unattainable due to high in-situ moisture contents.



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To account for this, a design change was made to the Zone S fill specification. The revised specification required that the fill be compacted to 98 percent of the Standard Proctor maximum dry density, with a moisture content less than 2 percent wet of the optimum moisture content, or as required to meet the compaction specification.

The downstream zone for the embankment (Zone B) is also comprised of well graded glacial till. The specification for Zone B originally required placement and compaction of glacial till (with the inclusion of some coarser material) in maximum 600 mm lifts to 90 percent of the Modified Proctor maximum dry density. However, coarser materials were not located in the borrow areas and the Zone B fill placement specification was subsequently revised to coincide with the revised Zone S specifications. During construction, both Zone S and B fill was placed in thin lifts (typically 100 to 150 mm) using scrapers and record testing was conducted every 300 mm for Zone S and 600 mm for Zone B.

The Chimney Drain is comprised of filter sand (minimum continuous width 1 metre) with Longitudinal and Outlet Drains for conveyance. The Longitudinal and Outlet Drains include filter sand, filter fabric, drain gravel and perforated CPT pipe.

The Stage Ib Perimeter Embankment is a homogeneous fill embankment comprised of Zone S glacial till placed and compacted in maximum 300 mm lifts to 98 percent of the Standard Proctor maximum dry density. The Perimeter Embankment also has a Chimney Drain which will be installed during Stage II construction.

6.3.5 Seepage Collection Ponds

The Main Embankment Seepage Collection Pond, located immediately downstream of the Main Embankment, was completed at the start of the Stage Ia construction program. The Perimeter Embankment Seepage Collection Pond was excavated during Stage Ib construction in 1997.





The seepage collection ponds will collect water from the embankment drain systems and from local runoff. The solutions will be recycled back into the Tailings Storage Facility. Details of the Main and Perimeter Embankment Seepage Collection Ponds are shown on Dwg. Nos. 1625.213 and 214.

6.3.6 Embankment Drainage Provisions

The following embankment drainage provisions have been incorporated to facilitate drainage of the tailings mass and to control the phreatic surface within the embankments:

Foundation Drains - Four foundation drains were installed with individual conveyance pipes for Stage Ia/Ib Main Embankment as shown on Dwg. No. 1625.210. The drains, which run parallel to the embankment axis, consist of perforated CPT tubing placed in a drain gravel surround which is encapsulated by filter fabric. The conveyance pipework for the drains runs into the Drain Monitoring Sump. At least two additional foundation drains will be included in future embankment raises.

<u>Chimney Drain System</u> - A Chimney drain has been included in the Main Embankment and is planned for both the Perimeter Embankment and the South Embankment. The chimney drains will provide a contingency drainage measure for control of the phreatic surface in the embankment and will also function as a crack stopper downstream of the core zone.

As illustrated on Dwg. Nos. 1625.111, 1625.207 and 1625.236, the Chimney Drain System includes the following components:

 Longitudinal Drains - The Longitudinal Drains are located at the bottom of the Chimney Drain. They are the collector drains and consist of perforated CPT tubing placed in a drain gravel surround encapsulated by filter fabric. The drain gravel and filter fabric are





surrounded by filter sand. A proper filter relationship exists between the filter sand and the drain rock.

- Outlet Drains The Outlet Drains consist of perforated CPT tubing placed in a drain gravel surround encapsulated by filter fabric. The drain gravel and filter fabric are surrounded by filter sand. The Main and Perimeter Embankments each include three Outlet Drains, which slope from the Longitudinal Drain at a minimum of one percent. For Stage Ia/Ib, the Outlet Drains were terminated at the downstream face of the Main Embankment, and will ultimately be connected to the Drain Monitoring Sumps.
- Chimney Drains The Chimney Drains extend vertically above the Longitudinal Drains and consist of Type B filter sand.

The Longitudinal Drains were constructed by excavating a trench in placed and compacted embankment glacial till fill or in dense natural ground. The Outlet Drains were constructed by excavating a trench in placed and compacted Zone B glacial till. The Chimney Drain was installed by various methods, including placing and compacting the filter sand in a vertical trench and/or in a mound. The construction techniques were closely monitored to ensure that the filter sand was not contaminated.

The Chimney Drain System at the Main Embankment was installed to El. 920 metres during Stage Ia construction. It was extended to El. 929 meters during Stage Ib construction in 1997 and will be further extended to El. 934 meters during Stage II construction in 1998. Future extensions will be included depending on the operational performance of the embankments, and on the type of construction materials which are incorporated in ongoing stages. The Longitudinal Drain was extended to El. 930.5 meters during the Stage Ib construction in 1997. Filter sand was mounded over the Longitudinal Drain sections above El. 929 metres. The drain gravel and Type B filter sand were processed at the mine site and at the rock borrow area.

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The Perimeter and South Embankment Chimney Drain systems will be identical to the Main Embankment. They will be installed during future construction programs.

<u>Upstream Toe Drain System</u> - Upstream toe drains will be included along the full length of the embankment at selected elevations during future staged expansions. The locations and elevations of the drains will be reviewed after an observation period during operations when parameters such as the tailings characteristics, available borrow materials and the performance of the facility have been established.

It is presently anticipated that the Upstream Toe Drain System will initially include a Longitudinal Drain and will be similar to the Chimney Drain System. Conveyance pipework which penetrates the embankment core zone will comprise solid HDPE pipe with seepage collars bedded in concrete and installed in the abutments in dense natural ground. The details and elevation of future core zone pipe penetrations will be finalized during the detailed design for the staged expansions. The Toe Drain conveyance pipe downstream of the Main Embankment ultimate toe was installed during Stage Ia/Ib construction.

On-going embankment expansions may incorporate cycloned sand for the upstream fill, should an acceptable and economic product be produced. Additional drainage measures will be added as required, based on the behaviour of the materials and the performance of the facility.

Preliminary details of the Upstream Toe Drain System are shown on Dwg. Nos. 1625.111 and 1625.236.

6.3.7 Staged Expansions

The anticipated staged expansions for the embankments are shown on Dwg. No. 1625.111. The staged expansions will incorporate a combination of





centreline and modified centreline construction methods and will utilize glacial till, cycloned sand and random fill for the various embankment zones. The specific requirements for the random fill zones will be determined after construction and operation of the first phase of the project. The on-going embankment raises must be re-evaluated during mine operations to ensure that adequate storage capacity and embankment freeboard are maintained throughout the life of the mine.

The embankment drainage systems will also be extended during on-going embankment expansions as required. Outlet drains and pipework will include suitable levels of redundancy to compensate for minor embankment settlements or earthquake induced deformation.

The design will be reviewed on an on-going basis. Modifications to drainage systems will be incorporated as required based on operating experience, monitoring records and availability of various embankment construction materials.

6.4 CONSTRUCTION MATERIALS

6.4.1 General

The predominant construction material for Stage Ia/Ib construction was a well graded glacial till which was borrowed locally, with the majority of the material coming from within the tailings impoundment. Lesser quantities of sand and gravel were required for the drains. As there are no nearby sand and gravel deposits, the required materials were produced from on-site crushing and screening operations.

During Stage Ia/Ib construction, materials were continually evaluated based on testwork completed in the on-site laboratory, in-situ field tests conducted with a nuclear densometer, and on visual inspections.

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6.4.2 Stage Ia/Ib Laboratory Testwork

Glacial till deposits within the tailings basin were sampled during the site investigation programs prior to the detailed design of the facility. Laboratory testing on selected samples included index testing to characterize the materials, specialized testing to evaluate the compaction characteristics, and permeability and shear strength testing to provide data for embankment design and analyses. Results from the earlier laboratory index testing are summarized on Table 6.4. Results from additional effective strength, compaction and permeability testwork are presented on Table 6.5.

In addition to the earlier laboratory testwork, extensive testing was conducted as part of the overall QA/QC program for Stage Ia/Ib construction. Prior to starting fill placement, numerous control tests were conducted in the on-site laboratory on samples obtained from the borrow areas, after stripping and grubbing was completed. The results of the control tests for glacial till from the Original, Alternate and Future Borrow areas are shown on Tables 6.6, 6.7 and 6.8, respectively. The median values are summarized as follows:

	Borrow Area		
Parameter	Original	Alternate	Future
	11.4	21.1	24.2
Gravel Content (%)	35.7	31.8	35.7
Sand Content (%)	43.4	39.7	36.2
Silt Content (%)	9.5	7.5	4.0
Clay Content (%)	12.5	11.4	11.6
Moisture Content (%)	16.5	15.1	
Plastic Limit, PL (%)		24.0	
Liquid Limit, LL (%)	24.4	8.4	
Plastic Index, PI (%)	8.9	-0.37	
Liquidity Index, LI (%)	-0.11		
Maximum Dry Density (kg/m³)	2057.5	2025.0	
Optimum Moisture Content (%)	9.7	10.5	
Specific Gravity	2.73	2.7	
Lab Permeability (cm/s)	1.8E-09	<u> </u>	<u></u>
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Plots of the median gradation limits for all three borrow areas are shown on Figure 6.3.

After fill placement commenced and it was confirmed that the fill would be placed wet of optimum, additional testwork was conducted to determine the relationship between moisture content and the undrained shear strength of the glacial till. The results of this testwork are shown on Figure 6.4.

Detailed results of all laboratory testwork conducted for the Stage Ia/Ib QA/QC program will be presented in the Stage Ia/Ib Construction Report.

6.4.3 Construction Materials for Future Raises

It is anticipated that either glacial till or non-acid generating rockfill materials from the mining operation will be selectively incorporated in the shell zones of the embankments for future raises. Local glacial till materials will continue to be exploited for the low permeability core zone, and adequate filters and transition zones will also need to be incorporated in future raises.

6.5 QA/QC PROCEDURES

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6.5.1 General

The quality assurance (QA) and quality control (QC) procedures are described in detail in the Site Inspection Manual (Knight Piésold Ref. No. 1625/2). Knight Piésold has directed the QA/QC testing for Stage Ia/Ib construction and will continue to direct the QA/QC testing for the future embankment expansions.

Laboratory testing required for the QA/QC program includes the following:

- Moisture Content (ASTM D2216).
- Particle Size Distribution (ASTM D422).





- Laboratory Compaction (ASTM D1557).
- Specific Gravity (ASTM D854).
- Atterberg Limits (ASTM D4318).
- Field Density (ASTM D2167).
- Laboratory and Field Air Entry Permeameter (LAEP or FAEP).
- Field Density by Nuclear Methods (ASTM D2922).
- Moisture Content by Nuclear Methods (ASTM D3017).

Detailed testing frequencies and schedules are outlined in the Site Inspection Manual.

Technical Specifications developed for the Work are described in the document "Tailings Storage Facility and Ancillary Works" (Knight Piésold, Ref. No. 1625/3), which includes for:

- Basin clearing, stripping and topsoil stockpiling.
- The evaluation of prepared subgrade and removal of unacceptable foundation soils (if any).
- The evaluation of the extent of the basin liner and basin liner fill
 placement inspection. (Basin liner limits and borrow materials were
 based on control tests, including particle size analysis, permeability,
 field density and moisture-density relationship.)
- Embankment fill placement, including setting out and grade control (survey) requirements, control and record testing schedule, fill acceptance criteria (as above), equipment requirements, geotextile selection (as needed), compaction specifications, etc.
- Supply and installation of geotechnical instrumentation, including vibrating wire piezometers, read out equipment and monitoring huts, survey monuments, flow monitoring equipment, etc.





 Installation of groundwater monitoring wells, including materials and procedures.

6.5.2 Supervision, Inspection and Testing

Knight Piésold personnel were on site for full time supervision, inspection and testing duties during Stage Ia/Ib construction. Key items addressed by Knight Piésold included foundation inspection and approval prior to fill placement, assessment of borrow material suitability, inspection of fill placement procedures, in-situ testing of the placed fill for moisture content and density, record and control testing at the required frequencies, and monitoring of all construction instrumentation.

As a result of the prolonged construction season, Stage Ia/Ib construction was completed under winter conditions. Embankment construction restarted in late November, 1996 and Stage Ia was completed by mid-December, 1996. The subsequent Stage Ib embankment raise occurred from mid January to mid March, 1997. During the two cold weather construction periods, additional Knight Piésold personnel were present to ensure that the design objectives were achieved in spite of the freezing conditions. Additional inspections included close monitoring of the borrow areas to ensure that unsuitable or frozen material was not hauled to the embankments. If any frozen material was inadvertently placed on the embankments, it was removed and replaced with non-frozen glacial till. Smaller work areas were developed, both on the embankments and in the borrow areas to reduce the exposure time to freezing conditions.

6.5.3 QA/QC Results

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Results of the Stage Ia/Ib QA/QC program will be presented in detail in the Construction Report. In summary, the QA/QC results have shown that the design objectives were achieved. The fill materials fit within the specified gradation envelopes and density and moisture content values were





acceptable. Median values for the glacial till fill placed at the Main and Perimeter Embankments (taken from the record samples) are as follows:

Parameter	Zone S	Zone B
Gravel Content (%)	17.3	18.7
Sand Content (%)	31.4	34.5
Silt Content (%)	41.3	38.4
Clay Content (%)	10.0	8.5
Moisture Content (%)	11.6	10.7
Plastic Limit, PL (%)	14.9	14.7
Liquid Limit, LL (%)	24.4	22.0
Plastic Index, PI (%)	9.7	8.0
Liquidity Index, LI (%)	-0.30	-0.45
Maximum Dry Density(kg/m³)	2080.0	2085.0
Optimum Moisture Content (%)	9.9	9.7
Specific Gravity	2.74	2.74
Lab Permeability (cm/s)	9.5E-10	1.1E-09

Results of the record tests for Zones S and B are shown on Tables 6.9 and 6.10 and plots of the median results for the particle size distributions are shown on Figure 6.5.

6.6 TAILINGS CONSOLIDATION ANALYSES

6.6.1 General

On-going consolidation of the tailings deposit is an important consideration for the design and construction of the facility during operations and at closure. Consolidation occurs continuously within the tailings deposit during deposition, and will continue after completion of operations until all excess pore pressures have dissipated. Expulsion of pore fluids during consolidation produces settlement of the tailings surface and a corresponding increase in the average density of the deposit.

A one-dimensional finite element computer model developed by Knight Piésold Ltd. was used to predict the magnitude and rate of tailings settlement



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and the corresponding average tailings density. This model used variable coefficients of consolidation, a void ratio versus effective stress relationship, an actual or predicted tailings deposition rate and large strain consolidation theory.

Analyses have been performed to predict tailings surface settlements and average densities during operations and at closure.

6.6.2 Parameters and Assumptions

The following parameters and assumptions were incorporated into the consolidation model:

- Void radio vs. effective stress and coefficient of consolidation vs. effective stress relationships were based on data for tailings materials which exhibit similar settling characteristics to the Mount Polley tailings. The selected parameters are considered representative of the tailings which comprise approximately 10 to 15% fine sand, 85 to 90% fines and have a specific gravity of about 2.75.
- An initial settled dry density of 1.0 tonnes/m³ was used, based on the results of laboratory settling tests for a Mount Polley tailings slurry with approximately 35% solids content.
- A daily production rate of 17,800 tonnes of dry tailings was assumed until closure. Tailings deposition was assumed to be carried out continuously over a 13 year operating period.
- The tailings are assumed to be deposited in horizontal layers across the entire tailings impoundment.
- An impermeable (no flow) boundary condition was modelled at the base of the tailings due to the presence of the underlying low permeability glacial till foundation and basin liner material.



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• Evaporative losses from the exposed tailings surface were ignored. Surface desiccation of the tailings will further assist in consolidation and densification of the deposit, particularly after closure when tailings deposition has ceased and the supernatant pond has been removed.

6.6.3 Results

Estimates of tailings surface elevation and average dry density have been computed for a 13 year operating period. An average dry density of approximately 1.1 tonnes/m³ was predicted after the first year of operation. The average dry density will likely increase to 1.2 to 1.3 tonnes/m³ and will be maintained until closure. The density of the deposit will increase more rapidly once tailings deposition ceases and self-weight consolidation continues. This would be assisted by surface desiccation due to evaporation.

The tailings deposit will remain partially consolidated during operations and for a period of time after closure until all excess pore water pressures have dissipated. The actual time taken for complete consolidation will be dependent on the in-situ consolidation characteristics of the tailings.

6.7 <u>EMBANKMENT SETTLEMENT</u>

6.7.1 General

Settlement of the embankment fill material occurs progressively as the embankment raises extend over the tailings beaches. Analyses have been carried out to predict the magnitude of these settlements using the one-dimensional finite element computer model described in Section 6.6. Surcharge loading has been applied to represent construction of overlying embankment raises in addition to on-going deposition and consolidation of the tailings. Analyses have been carried out to predict embankment settlements after each on-going staged expansion.





Two tailings columns were evaluated at increasing distance from the Stage Ib embankment crest, as shown on Figure 6.6 and are summarized as follows:

- Column A 6 metres of tailings overlain by Stages IV to VIII.
- Column B 30 metres of tailings overlain by Stages VII and VIII.

Void ratio vs. effective stress and coefficient of consolidation vs. effective stress relationships for the tailings beach materials were based on data for similar coarse tailings from existing tailings facilities. An initial settled dry density of 1.2 tonnes/m³ was adopted for the beach tailings, which will consolidate more rapidly than the tailings slimes within the facility.

Tailings will be discharged into stored make-up water in the initial year of operations. Lateral segregation of tailings and beach development will be limited during this time and tailings deposited adjacent to the embankment will consist of a mixture of bulk tailings material. This material will likely not consolidate as rapidly as the coarser beach tailings. The average degree of consolidation of this initial "bulk" tailings zone has also been predicted. Parameters used for the tailings consolidation analyses described in Section 6.6 were adopted to represent these tailings and are considered conservative.

6.7.2 Results

Estimates of embankment settlements have been made for staged expansions up to the final Stage VIII crest at El. 965 metres. These represent the maximum settlements at the deepest section of the embankment.

The "bulk" tailings adjacent to the Stage Ib embankment crest will be approximately 50% consolidated prior to construction of the Stage II raise. A resulting settlement of about 0.3 metres is expected during construction of Stage II onto these "bulk" tailings. This settlement will occur during initial placement of the coarse bearing layer on the tailings and during placement of the remaining fill. Consolidation will occur rapidly during fill placement and the underlying tailings are predicted to be over 90% consolidated immediately



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des Ingénieurs Conseils after construction. These "bulk" tailings do not have a significant effect on predicted embankment settlements because the compressibility of this material is likely to be higher than the coarser beach tailings only at low effective stresses. Due to confinement from additional tailings, effective stresses will increase in this underlying material. Therefore, the compressibility of these tailings will be similar to the overlying beach tailings by the time on-going embankment raises are constructed.

For the staged expansions, the majority of the settlement for both Columns A and B occurs during placement of embankment fill, as described above. A coarse bearing layer (where necessary) and fill placement during construction routinely compensates for these initial tailings settlements. Excess pore pressures generated in the tailings during fill placement dissipate rapidly and the degree of consolidation is typically 70 to 90% by the end of construction of each raise.

Embankment settlements after construction of each raise in Column A will be negligible (less than 0.1 metres) and the underlying tailings will be consolidated shortly after each raise. Settlements for Column B are also expected to be minor, approximately 0.6 metres and 0.2 metres after construction of Stages VII and VIII, respectively. It was assumed that only one toe drain located at the upstream shoulder of the Stage Ib crest was available to assist in the drainage and dissipation of excess pore pressures in the tailings for these analyses. If required, additional toe drains will be incorporated to allow increased consolidation in the tailings mass. Therefore, settlements are likely to be even smaller than those predicted above.

On-going settlements due to additional embankment raises generally reduce as the tailings become less compressible at the high confining pressures from the overlying fill. Settlements will also vary laterally along the embankment crest due to the variable thickness of the underlying tailings. The minor settlements given above correspond to maximum values in the deepest section of the facility and therefore differential settlements will not be significant.





On-going fill placement during staged expansion of the embankment routinely compensates for settlement of the embankment crest. Sloping internal embankment zones and the chimney drain will deform slightly but will result in only a very slight flattening of the embankment drainage system. This will not reduce the efficiency or integrity of the system.

6.8 <u>STABILITY ANALYSIS</u>

6.8.1 General

Embankment stability analyses were conducted using the limit equilibrium computer program SLOPE/W. This program performs a systematic search to obtain the minimum factor of safety from a number of potential slip surfaces. Factors of safety were computed using Bishop's Simplified Method of Slices.

<u>Downstream Stability</u> - Analyses were performed to investigate the downstream stability of the final Main Embankment for the following conditions:

- Static conditions during operations and post-closure. Minimum acceptable factors of safety of 1.3 (during operations) and 1.5 (post-closure) have been adopted for these cases.
- Earthquake (pseudostatic) loading during operations and post-closure. The stability of the embankment under earthquake loading was analyzed by applying a horizontal seismic coefficient (acceleration) to the potential sliding mass. Factors of safety greater than 1.0 imply that there will be no deformations of the embankment initiated by earthquake loading. For conditions during operations, the Design Basis Earthquake was used (as determined by the hazard classification for the Tailings Storage Facility). The Maximum Design Earthquake was used for post-closure (long-term) conditions.



Post-liquefaction (residual tailings strength) condition. A minimum
acceptable factor of safety of 1.1 is considered appropriate for this
condition.

The tailings were assumed to be partially consolidated during operations (based on the results of the consolidation analysis) and an appropriate undrained shear strength was assigned to the tailings. Tailings effective strength parameters were used for the long-term post-closure condition when complete consolidation has been achieved.

During operations, the tailings beach and upstream embankment drainage systems will ensure that the phreatic surface is kept away from the upstream face of the embankment. However, a worst case condition with an elevated phreatic surface within the embankment core zone was also evaluated.

<u>Upstream Stability</u> - The upstream stability of the Main Embankment has also been evaluated for Stage Ib (during storage of make-up water) and at closure for the final embankment configuration.

The influence of construction pore pressures on embankment stability has also been considered.

6.8.2 Material Parameters and Assumptions

The following parameters and assumptions were used in the stability analyses:

• Bulk unit weights for the embankment and foundation materials are based on testwork conducted on representative samples as part of the 1995 geotechnical investigations. The results were originally presented in the "Report on 1995 Geotechnical Investigations for Mill Site and Tailings Storage Facility" (Knight Piésold Ref. No. 1623/1). An average bulk unit weight for the tailings deposit adjacent to the embankment was estimated from the results of the consolidation





analysis. The cycloned sands (Zone CS) were assigned a typical value for this material.

- Partially consolidated tailings during operations were assigned typical undrained shear strengths ranging from 10 kPa to 55 kPa at depth. For fully consolidated tailings an average effective friction angle of 30° was adopted. These are based on lower bound strengths from insitu Shear Vane and Cone Penetration Testing obtained at other mine sites for similar tailings materials.
- Effective strength parameters for the embankment fill and foundation materials were obtained from consolidated-undrained triaxial testwork performed on representative samples obtained during the 1995 geotechnical investigations and reported in the "Report on 1995 Geotechnical Investigations for Mill Site and Tailings Storage Facility" (Knight Piésold Ref. No. 1623/1). A summary of these results is included on Table 6.11. The downstream shell zone has conservatively been assumed to comprise glacial till materials only, and no additional strength allowance has been included for rockfill materials which will likely be incorporated.
- An undrained shear strength of 85 kPa was adopted to represent the strength of the top two metres of the Stage Ib foundation soils. This value is based on the lower third bound strength obtained from the 1996 cone penetration tests, (Conetec Field Report, July 31, 1996).
- An effective friction angle of 26° was used to represent the strength parameters of the top two metres of the Final Embankment foundation soils. These strength parameters account for long-term consolidation conditions of the foundation soils. This value was based on the consolidated undrained triaxial testwork performed on glacial till samples. These samples were obtained during the 1995 geotechnical investigations and reported in the "Report on 1995 Geotechnical

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Investigations for Mill Site and Tailings Storage Facility" (Knight Piésold Ref. No. 1623/1).

- An effective friction angle of 32° was adopted for the cycloned sands
 (Zone CS), which is considered conservative for this material.
- An undrained shear strength of 10 kPa was conservatively adopted to represent the residual (post-liquefaction) strength of the tailings. This is based on lower bound values obtained for similar tailings and is also consistent with lower bound data presented by Seed (1990) for the residual undrained shear strength of sand.
- An average effective friction angle of 30° was adopted to represent the
 coarse beach tailings beach underlying on-going embankment raises.
 These coarser, more free draining tailings will consolidate rapidly.
 Modelling has shown that these tailings achieve complete consolidation
 shortly after placement of the embankment raise.
- The location of the phreatic surface has been estimated from steadystate seepage analyses. The details are provided in Section 6.9.
- A hydrostatic pore pressure of 1.5 metres above ground was applied to the foundation soils on the downstream side of the embankment. This piezometric condition has been added to simulate baseline artesian pore water pressures within the foundation materials. This value is based on the initial readings from the foundation piezometers, the pore pressure dissipations from the cone penetration tests and the observation of the nearby groundwater well. A sensitivity analysis to check the stability for various hydrostatic pore pressures has been calculated.

The geometry, material parameters and location of the phreatic surface for the stability analyses are illustrated on Figure 6.7.





6.8.3 Results of Analyses

Downstream Stability - For the static case during operations a minimum factor of safety of 1.4 was calculated. This value will increase after closure as tailings consolidation continues with a corresponding gain in strength. Once a minimum factor of safety of 1.5 is obtained it is no longer dependent on tailings strength and the potential slip surface is located within the embankment. The minimum factor of safety for the potential critical slip surface is 1.5. For a deeper embankment slip surface, the factor of safety is 1.6. The location of potential slip surfaces during operations and post-closure are given on Figures 6.8 and 6.9, respectively.

A sensitivity analysis was conducted to evaluate the downstream static stability of the Stage 1b and Final Main Embankments for various hydrostatic pore pressures in the foundations soils. The results, shown on Figure 6.10, indicate that a minimum Factor of Safety of 1.1 is approached as the foundation pore pressures reach a height about 6.0 and 8.5 metres above ground for the Stage 1b and Final Main Embankments, respectively. The pore pressures will be monitored and appropriate actions will be taken to assure embankment stability. These actions may include implementation of depressurization measures and/or buttressing with suitably graded filter materials as discussed in the Operating Maintenance and Surveillance Manual for Stage Ib Embankment El. 934 metres (Knight Piésold Ref. No. 1627/3).

The post-closure factor of safety is dependent on the degree of consolidation and strength of the tailings material adjacent to the embankment. If on-going monitoring records and stability evaluations indicate that the minimum required factors of safety are not achieved, a small stability berm will be constructed at the downstream toe to obtain factors of safety greater than 1.5.

For the extreme case of an elevated phreatic surface within the embankment core zone a factor of safety of 1.4 was calculated for the embankment at closure. If on-going monitoring and piezometric records indicate that an elevated phreatic surface can develop in the embankment, appropriate



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stabilization measures will be incorporated to ensure a minimum factor of safety of 1.5 after closure. The use of rockfill materials in the downstream shell zone will also enhance the embankment stability.

For earthquake (pseudostatic) loading during operations a minimum factor of safety of 1.3 was computed for a seismic coefficient of 0.04, corresponding to the Design Basis Earthquake. For post closure conditions and a seismic coefficient of 0.065, representing the Maximum Design Earthquake, a factor of safety of 1.2 is obtained for the potential critical slip surface. For a deeper slip surface, the factor of safety is 1.3. The location of the potential slip surface for each case is shown on Figures 6.11 and 6.12. It should be noted that even for a seismic coefficient of 0.13, representing the Maximum Credible Earthquake, the minimum factor of safety is greater than 1.0, which implies that the embankment will be stable with no deformation.

The tailings residual strength case represents the steady state strength of the material after degradation by in-situ straining. Such a condition can occur if liquefaction is initiated in the material by rapid static or seismic loading which causes a corresponding increase in pore pressures. A factor of safety of 1.4 was calculated for this case. This indicates that the embankment is not dependent on tailings strength to maintain overall stability. The potential slip surface for this case is shown on Figure 6.13.

<u>Upstream Stability</u> - The minimum upstream static factor of safety for the Stage Ib embankment during water storage is 1.5 and will increase once tailings deposition commences. Under seismic loading conditions for the Design Basis Earthquake, a minimum factor of safety of 1.3 was computed. The probability of occurrence of such an event during Stage Ib construction and water storage is extremely low. However, even for these worst case conditions a factor of safety well above the minimum of 1.0 is maintained.

For the final impoundment configuration, the minimum upstream static factor of safety is in excess of the minimum 1.5. The minimum factor of safety under seismic loading is well above the required minimum of 1.0.





The location of potential slip surfaces and assumed phreatic surfaces for each case are shown on Figure 6.14. The stability analyses show that the Main Embankment has an acceptable factor of safety under the most extreme and seismic conditions.

<u>Construction Pore Pressures</u> - The influence of the construction pore pressures on the stability of the Stage Ib embankment has also been considered. The starter embankment was constructed to a maximum height of 20 metres over a 7 to 8 month period and represents the largest construction lift for the facility. Close control of the fill moisture contents during construction prevented excess pore pressure development in the fill zones.

Pore pressures will be routinely measured during construction of on-going raises using piezometers installed into the embankment fill and foundations. Stability analyses will be required and evaluated for the design of on-going embankment raises.

6.9 **SEEPAGE ANALYSIS**

6.9.1 General

Seepage analyses were performed using the finite element computer program SEEP/W. The purposes of the analyses were:

- To establish the pore water pressures within the embankments for stability analyses.
- To estimate the amount of seepage discharge from the tailings storage facility.

The seepage analyses have been conducted for the Main Embankment only and were based on the cross-section shown on Figure 6.15. The parameters used such as the saturated hydraulic conductivity, are also included on Figure





6.15. For a conservative assumption, the final Main Embankment length of 1,200 metres was used in the seepage analyses.

During the initial year of operations, tailings will be discharged into stored make up water, resulting in limited beach development. As a conservative approximation, fine tailings (zone 3), have been assumed to extend to the upstream face of the embankment up to the maximum stored make up water elevation of 925 m.

6.9.2 Summary of Parameters

Saturated and unsaturated hydraulic conductivities were determined for each material in the embankment and foundation zones. In assigning hydraulic conductivity values for the seepage analysis, typical conductivity functions for similar soil types were used. These functions were adjusted to correspond with the actual saturated conductivities of the material. Hydraulic conductivity values for the tailings mass, embankment and foundation were determined as follows:

- The tailings mass was sub-divided into three zones with decreasing hydraulic conductivity to account for the less permeable consolidated tailings at depth.
- Hydraulic conductivity values for the various zones of the embankment and foundation soils were estimated based upon typical values for similar materials.

6.9.3 Boundary Conditions and Flux Sections

Boundary conditions were imposed on the modelled sections to more accurately represent hydrogeologic conditions in the field. These conditions are summarized as follows:



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- A no-flow boundary condition was assigned along the left side of the model (upstream of the embankment).
- A total head boundary was imposed at the tailings surface to model a supernatant pond.
- The upstream embankment toe drain was modelled by applying a nohead condition at that location.
- Foundation drains were modelled by applying no-head nodes at drain locations.
- A hydrostatic pore pressure profile with the water table 2 metres below the ground surface was assigned to the right boundary of the model (downstream of the embankment).

Flux sections were included in the model to estimate seepage flow across the various geological units, as well as the engineered components. The following locations, in particular, were examined closely:

- Seepage inflow to the upstream toe drain.
- Seepage flow collected by foundation drains.
- Seepage flow which bypasses the seepage collection systems.

The flows collected by the seepage collection systems (i.e. the upstream toe drain and foundation drains) will drain to the Main and Perimeter Embankment Seepage Collection Ponds. These seepage flows will be recycled to the tailings impoundment. The seepage flows which bypass the seepage collection systems are the only component which will be lost to groundwater.

6.9.4 Results

All seepage flow estimates are projected increases over baseline flow rates. In particular, the embankment foundation drains include a baseline



groundwater flow component which is not factored into the following flow projections. Case one modelled the embankment drain as designed and represents the expected final Main Embankment configuration. The total solution flow rate of 1.52 l/s (24 gpm) was calculated from the seepage analysis. Approximately 97 percent of the flow was collected in the upstream toe drain and the embankment foundation drains while the remaining 3 percent of the solution flowed through the foundation. In case one, the solution flow contribution made by each of the components is as follows:

- The upstream toe drain collected 95 percent (1.45 l/s or 23 gpm).
- The embankment foundation drain system collected 2 percent (0.03 l/s or 0.4 gpm).
- Seepage loss through the foundation was 3 percent (0.04 1/s or 0.6 gpm).

Case two modelled the failure of the upstream embankment toe drain. The results provide the maximum seepage pore pressure for use in the stability analysis of the final Main Embankment. The total solution flow of 0.16 l/s (2.5 gpm) was calculated from the seepage analysis. Approximately 88 percent of the flow was collected in the embankment foundation drains, while the remaining 12 percent of the solution flowed through the foundation.

In case two, the solution flow contribution made by each of the components is as follows:

- The embankment foundation drain system collected 88 percent (0.14 l/s or 2.2 gpm).
- Seepage loss through the foundation was 12 percent (0.02 1/s or 0.3 gpm).





The seepage rates presented above are expected maximum incremental values which occur late in the project. However, during the early years of operation, seepage rates are expected to be lower, particularly at the Perimeter and South Embankments where the natural groundwater table provides complete hydraulic confinement during the first year. As the tailings surface rises, the seepage rate is expected to gradually increase to the maximum values presented above.

For both cases, the baseline groundwater flow rates recorded during construction and in the period since are much greater than the flow rates predicted for the foundation drains as a result of tailings impoundment. The total maximum baseline flow rate to date has been measured at approximately 0.9 l/s.

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tributary catchment above the sediment pond. The storage capacity can be increased by selectively routing flows into the tailings system. Erosion control and energy dissipation riprap will be provided as necessary in spillways and discharge channels.

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The need for sediment removal at sediment control ponds should be checked annually and after major storm events and carried out when necessary.

7.3.3 Polley Lake Pumping System

The Polley Lake Pumping System is a contingency option that can provide additional make-up water to the milling process. The system was designed for a maximum water extraction rate of 8,000 US gpm and will operate during the spring freshet only. It includes a submerged intake structure, suction pipeline, pumping station and a discharge pipeline into the Tailings Storage Facility. The general arrangement of the pumping system is shown on Dwg. No. 1628.001. The design was presented in the report "Polley Lake Pumping System" (Knight Piésold Ref. No. 1628/5).

Based on annual water balances for average precipitation conditions, the extraction of up to 1 million cubic metres may be required for the first 3 years of operations, after which the annual make-up water requirements decrease.

The amount of water that can be extracted from Polley Lake is limited by the minimum fish flow requirements for Hazeltine Creek. To minimize the negative potential impacts on Hazeltine Creek, water will only be extracted from Polley Lake during the freshet, from approximately mid March to mid May. Based on annual outflow hydrographs for Polley Lake, approximately 1.4 to 2.0 million cubic metres of water should be available annually for pumping. However, the actual water available for pumping will be determined by monitoring flows in Hazeltine Creek during the pumping period.

7.3.4 Sediment Control Ponds and Discharge Requirements

Sediment control ponds are sized to provide storage (Southeast Sediment Pond) of the 1 in 10 year - 24 hour storm runoff flood flows from the





- Catchment above the tailings access road, between the Millsite and Bootjack Creek - Runoff will be directed to Bootjack Creek with no transfer to the Tailings Storage Facility.
- Catchment areas downstream from the tailings embankments -Runoff for some areas will be directed to the Seepage Collection Ponds and transferred to the Tailings Storage Facility.
- Millsite area Diversion ditches along the lower perimeter of the Millsite area will collect and direct runoff to the Millsite Sump. The water will be pumped to the mill or will flow by gravity to an inlet point (T1) on the tailings pipeline so that it can be returned to the process circuit via the tailings and reclaim systems.
- Waste Rock Storage Area A diversion ditch along the base of the waste rock storage area will collect and direct runoff to the Southeast Sediment Pond. From here, water can be directed into the process circuit via the Reclaim Booster Pump Station or the T2 Dropbox. The design criteria for the diversion ditches are as follows:
 - The ditches are sized to convey the peak flow runoff from the 1 in 10 year precipitation event. This flow was determined on the basis of site IDF values and the rational formula, as outlined in the MOE Manual of Operational Hydrology.
 - Erosion protection and energy dissipation will be provided to minimize erosion under normal operating conditions. Erosion damage requiring repairs to surface water control facilities would be expected during extreme flood events.
 - For design purposes, "extreme" floods are defined as those with an annual probability of occurrence of less than 2%.

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- Minimize the volume of fresh water extracted from Polley Lake.
- Monitor the quality of surface runoff from disturbed areas and groundwater flows within the project site.
- Release only the highest quality water from within the project boundaries in accordance with permitted requirements. The volumes to be released will be as necessary to maintain an overall project water balance under actual hydrometeorological conditions.
- Manage the tailings supernatant pond to optimize the volume of water stored on the tailings surface during operations and provide the final volume of ponded water in the tailings basin at closure. The objective at closure is to provide a walk-away solution.
- Develop and maintain a detailed data base to allow water balances for the site to be as accurate as possible, thereby becoming useful tools for predicting annual make-up water requirements and for scheduling releases of clean surface runoff water as appropriate.

7.3.2 Surface Runoff Diversion Ditches

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Some of the surface runoff diversion ditches will be used to direct runoff to the Tailings Storage Facility from various catchments as follows:

- <u>Catchment above the Tailings Storage Facility</u> Runoff from the area directly above the Tailings Storage Facility will be captured during the early years of operation.
- <u>Catchment above tailings area access road, below Bootjack Creek</u> Runoff in the ditch for the access road will be directed to the
 Tailings Storage Facility. The pipelines are in a separate pipe
 containment channel.





The annual withdrawal of the maximum of 1.0 million cubic metres of water from Polley Lake will likely only be required prior to start-up and for the first three to four years of operations. Withdrawal requirements for subsequent years will decrease progressively with time due to the increased consolidation of tailings in the pond (greater release of pore water) and the progressive development of the open pit and waste dump.

7.2.3 Water Supply at Start-Up

As stated above, the updated water management plan indicates that approximately 1.9 to 2.5 million cubic metres of water should be stored in the Tailings Storage Facility prior to mill start-up. This water volume will be obtained by capturing the 1997 freshet and by pumping up to 1 million cubic metres from Polley Lake.

7.3 WATER MANAGEMENT PLAN

7.3.1 General

The objective of the project Water Management Plan in the early years is to route all project water flows from disturbed areas into the milling process or into associated mine site activities such as dust suppression. In the later years of operation the objective will be to monitor and release selected surface water inflows in order to manage the final volume of ponded water in the tailings impoundment at closure. The following activities will be incorporated in the Water Management Plan:

- Maximize the capture of surface and groundwater flows from within the project area.
- Maximize the use of the uncharacteristic water recovered from within the project area in the milling process.





the analyses are consistent with those used in previous water balances for the tailings impoundment and mine site. For this report, average annual water balances for years 1 and 13 are presented.

7.2.2 Water Balance Results

When conducting the water balance, a fundamental consideration was that an adequate volume of water must be available to the milling circuit during the cold winter months (when precipitation accumulates as snow and surface runoff is at a minimum) or if drier than average conditions occur. This requirement can be met by:

- Providing up to 2.5 million cubic metres of water in the tailings impoundment prior to start-up.
- Ensuring that 1.9 to 2.5 million cubic metres of water is available in the tailings impoundment following the freshet during on-going operations.
- Allowing for contingency water extraction of up to 1.0 million cubic metres annually from Polley Lake during peak flow months.

The water balance components for years 1 and 13 are shown on Figures 7.2 and 7.3, respectively. The results of the water balance for years 1 and 13 are shown on Tables 7.1 and 7.2, respectively, and indicate that water stored in the Tailings Storage Facility will be at a minimum in March of every year, just prior to freshet. The subsequent snowmelt will significantly increase the water storage in the tailings impoundment. However, it is unlikely that the freshet alone will provide the volume required to carry operations through the following winter. Consequently, additional make-up water from Polley Lake (up to about 1 million cubic metres) will be required to make up the difference so that sufficient process water is available for reclaim. A schematic illustration of the water balances for these years is presented on Figure 7.4.



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SECTION 7.0 - WATER BALANCE AND MANAGEMENT

7.1 GENERAL

The water management strategy for the Mount Polley Project was originally presented in the "Report on Project Water Management" (Knight Piésold Ref. No. 1624/1). The current revised water management strategy differs from the concept originally presented as catchment areas A, B and C will no longer be utilized as a source of surface runoff for transfer to the tailings impoundment. Instead, up to one million cubic metres of water will be extracted annually from Polley Lake during the spring freshet high flow period.

The overall project components include disturbed and undisturbed areas at the open pits, waste rock storage area, Millsite, Tailings Storage Facility, the undisturbed catchment area immediately upstream of the Tailings Storage Facility and the diverted areas downstream of the tailings embankments. A schematic of the site water management plan is shown on Figure 7.1.

7.2 WATER BALANCE

7.2.1 General

The overall project water balance was completed in the "Report on Project Water Management" by integrating the water balances for the mine site with those for the Tailings Storage Facility. The analyses included a comprehensive series of water balances to evaluate the volumes of surface runoff water available throughout the life of the mine. A probabilistic water balance analysis using the @RISK Analysis and Modelling program was developed to describe the effects of a statistical range of precipitation conditions over the entire life of the project. Over 1000 different combinations of wet and dry precipitation conditions were considered.

The staged development plans for the various components of the project are included in the water balance analyses. Specific assumptions incorporated in



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SECTION 8.0 - PIPEWORK

8.1 GENERAL

This section describes the pipework and pump systems required for the tailings and reclaim pipelines and for the Tailings Storage Facility seepage recovery system. Design criteria for the various components of the Tailings Storage Facility pipework are provided. The pipeline plan and profiles are shown on Dwg. Nos. 1625.218, 222 and 228, with sections and details shown on Dwg. No. 1625.219, 223, 224, 225 and 226.

8.2 TAILINGS PIPEWORK

8.2.1 General

The tailings pipeline extends approximately 7,000 metres from the Millsite to the southwest corner of the Tailings Storage Facility (the right abutment of the Main Embankment). The tailings system is designed to flow by gravity and to be self-draining.

8.2.2 Design Criteria

The tailings pipeline has been designed to the following criteria:

- Millsite tailings discharge at El. 1,110 metres (approx.).
- Tailings embankment crest elevation at start-up at El. 934 metres.
- Tailings embankment crest elevation at end of mine life at El. 965 metres.
- Continuous downhill grade to ensure pipeline is free draining and to prevent potential sanding and freezing problems.





- Pipe diameter selected for gravity flow (open channel or non-pressure flow) over a range of operating conditions.
- All pipework is butt fusion welded HDPE, with the exception of a short (300 metres) stretch of steel pipe at the reclaim barge.
- A dropbox (T2) provided along the tailings line allows for the addition of waste dump runoff from the Southeast Sediment Pond to the tailings stream and serves as an overflow for the reclaim booster sump.
- Spill containment is provided for the full length of all pipelines.
- The pipelines are buried through the Millsite area and are laid in a pipe containment channel cut in or lined with glacial till from the Millsite to the Tailings Storage Facility.
- The pipelines are sleeved at the Bootjack Creek crossing to provide additional spill containment.

8.2.3 Tailings Delivery Pipework

The tailings delivery pipework consists of High Density Polyethylene (HDPE) pipe of varying diameter. Pipe wall thickness (pressure rating) was selected to accommodate the anticipated operating pressures and vacuum conditions and includes an allowance for internal abrasive wear.

The tailings pipeline has two sections, with different pressure ratings and diameters. The first section extends from the Millsite to the T2 Dropbox and is comprised of 22 inch (556 mm) DR 17 HDPE pipe. The second section extends from the T2 Dropbox to the Tailings Storage Facility and comprises 24 inch (610 mm) DR 15.5 HDPE pipe. Two sections of 30 inch (762 mm) DR 15.5 HDPE pipe are also included at the start of the two pipeline sections (at the Millsite and at the T2 Dropbox) to ensure that flows are not restricted at the inlets.





8.2.4 Tailings Discharge Pipework

At the Tailings Storage Facility, the pipeline will run along the inside crest of the embankment. The pipeline will be provided with a movable discharge section comprising twelve lengths of pipe (199.2 m) with 150 mm offtakes near the end of every second pipe (6 offtakes total). The movable discharge section will allow controlled deposition of tailings over the tailings beach. The tailings pipeline has been installed with a number of flanged connections where the movable discharge section can be installed. For the first year of operations, discharge will be concentrated from the Main Embankment at the deepest part of the impoundment to establish the tailings beach and from the right abutment of the Main Embankment to cover the Upper Basin Liner. Additional discharge will be provided at the M1 dump valves, as required.

The offtakes will be made from one of a number of commercially available options for this purpose. A typical offtake or "spigot" comprises:

- a strap-on tee sleeve (Robar type) with a flanged 150 mm outlet;
- a valve, which is required for regulation of high flows or pressures;
- a length of heavy duty (material handling type) hose or HDPE pipe;
- a pipe anchored to the embankment to direct tailings flow and minimize erosion of the embankment fill.

The tailings pipeline will be secured on the embankment crest by straps and concrete blocks or guide posts to restrict thermally induced movements.

8.3 RECLAIM WATER SYSTEM

8.3.1 General

The reclaim water system returns water from the Tailings Storage Facility to the Millsite for re-use in the process circuit. The reclaim water system





comprises a pump barge, the reclaim pipeline and a reclaim booster pump station.

8.3.2 Design Criteria

The reclaim water system is designed to meet the following criteria:

- To provide adequate pipeline and pumping capacity to meet operational process requirements.
- To provide a Stage Ib drawdown range of the pond using the pump barge channel from El. 932 to El. 917.
- To provide a booster pump station at the midpoint of elevation to reduce pressure rating requirement. Pumps identical to those at the barge will be used in the booster station to reduce spare part requirements and to simplify maintenance.

8.3.3 Reclaim Barge

The reclaim barge will be a prefabricated floating pump station complete with perimeter trash screens, internal wet well(s), pump(s), valving, piping, electrical power, instrumentation and control circuitry. The reclaim barge will be designed by Others.

A hinged walkway/pipe bridge will be provided for access to the barge from the side of the reclaim barge channel. The reclaim barge channel is shown on Dwg. No. 1625.206.

8.3.4 Reclaim Pipeline

The reclaim pipeline will be constructed in two sections, with varying pressure ratings to accommodate anticipated operating pressures and vacuum conditions. The first section will extend from the pump barge to the booster





pump station and will include a stretch (approximately 300 metres) of steel pipe at the reclaim barge. The remainder will consist of HDPE pipe which decreases in thickness (pressure rating) as the booster pump station is approached and the pressure head is decreased. The second pipe section is similar to the first, but does not have any steel pipe sections. Nominal 24 inch (610 mm) HDPE pipe with varying pressure ratings has been selected to provide the required water transfer capacity.

8.3.5 Reclaim Booster Pump Station

The reclaim booster pump station will be constructed using parts identical to those at the pump barge to the greatest degree possible. This will ensure ease of maintenance and will allow parts to be interchanged as needed.

An inter-linked control system, to be designed by Others, will co-ordinate pump operations with process water demand at the millsite. The control system and pipework design will include the necessary provisions for spill prevention.

8.4 <u>SEEPAGE RECOVERY SYSTEM</u>

8.4.1 Monitoring Wells

The monitoring wells around the perimeter of the Tailings Storage Facility will be used to monitor seepage from the tailings area. These wells will be sampled prior to start-up to establish baseline ground water quality and during operations to check for process constituents in the groundwater.

The monitoring wells can not be operated as seepage recovery wells. However, on-going monitoring will indicate if additional seepage recovery wells are required. Any new seepage recovery wells will be suitably sized to permit installation of submersible pumps.





8.4.2 Embankment Foundation Drains

Embankment foundation drains and pressure relief wells and trenches located immediately below the compacted glacial till at the Main Embankment will relieve any groundwater or seepage pressures which may develop in the embankment foundations. The drains will convey the resulting flows to the Drain Monitoring Sump, where the outlets will be monitored for flow quantity and water quality.

In the event that an additional pressure relief trench is required on the right abutment of the Main Embankment, it will be directed to the Drain Monitoring Sump, where a new outlet can be installed and monitored for flow quantity and water quality.

All foundation drain flows will be recycled to the tailings impoundment via the Drain Monitoring Sump.

8.4.3 Toe Drains

The upstream toe drains shown on Dwg. Nos. 1625.111 and 1625.236 are included to dewater the tailings beach. They are an important component for the modified centreline raising of future stages of the embankment. The detailed design of the upstream toe drain system will be completed prior to Stage III construction, after the tailings characteristics, available borrow materials and the performance of the facility during operations can be reviewed.

The toe drain outlets will discharge to the drain monitoring sumps. The flows will be monitored for flow quantity and water quality and the results correlated to other instrumentation and monitoring data.





8.4.4 Drain Monitoring Sump

Drain monitoring sumps are provided upstream of the Main and Perimeter Embankment Seepage Collection Ponds to allow flow measurements and water quality samples to be taken from the individual drains. The sumps are constructed using 1.8 m diameter manholes with low level outlets to the seepage collection ponds as shown on Dwg. No. 1625.202.

8.4.5 <u>Seepage Collection Pond Recycle Pumps</u>

The seepage collection ponds are provided with recycle pump systems to pump the water back to the Tailings Storage Facility. The seepage collection and recycle system is shown schematically on Dwg. No. 1625.213. The pipelines consist of six inch diameter HDPE pipe with suitable pressure ratings and will extend from the pumps to the crest of the tailings embankment to discharge directly onto the tailings beach.

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SECTION 9.0 - INSTRUMENTATION AND MONITORING

9.1 **GENERAL**

Geotechnical and environmental instrumentation and monitoring systems are essential to evaluate the performance of the embankment and associated structures and to detect abnormal conditions relevant to dam safety. Close monitoring of the Tailings Storage Facility is especially important in the first year of operations, before the tailings beaches are established (Stage Ia/Ib will initially function as a water storage dam to store runoff and make-up water for mill start-up.)

9.2 MONITORING PROGRAM

The monitoring program for the Tailings Storage Facility includes the following:

a) Measurement of the rate of filling with water and/or tailings.

The pond elevation, depth, area and volume must be closely monitored:

- To ensure that the required amount of make-up water (maximum 2.5 million cubic metres) is not exceeded so that there is adequate storage capacity for tailings until construction of the next raise.
- To enable monitoring of the supernatant pond depth / area / volume so that tailings characteristics such as dry density can be determined.
- To monitor water recoveries.
- To enable the correlation of the pond level with other data, such as the piezometers and drain flow quantities.

The filling rate can be monitored by measuring the pond level and using a staff gauge together with the depth / area /capacity curve for the facility. The



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d) Monitoring of the vibrating wire piezometers.

For Stage Ia/Ib, vibrating wire piezometers have been installed along three instrumentation planes at the Main Embankment. One piezometer was also The piezometers measure pore installed at the Perimeter Embankment. pressures in the foundation soils, the embankment foundation drains and the embankment fill zones, including the Chimney Drain.

Summary plots for data collected to date at the three Main Embankment piezometer planes are shown on Figures 9.2, 9.3, and 9.4. The results show that:

- Pore pressures in the foundation soils have remained at or near preconstruction levels (Piezometers A2-PE2-01/O2, B2-PE2-01/02, C2-PE2-01/02). Temporary pore pressure increases in the range of one metre of head resulted from fill placement. Foundation pore pressures also increased by approximately 1 m due to impounding water to an elevation of about 926 m.
- Pore pressures in the foundation and chimney drains have remained at zero, indicating that all drains are unimpeded and are functioning well (Piezometers A1-PE1-01/02/03, B1-PE1-01/02/03, C1-PE1-01/02).
- The fill piezometers responded quickly to the placement of additional material and were monitored accordingly. Some high pressures were observed (greater than the load from the new fill). This is attributed to the piezometer installation method, where the saturated tips were immersed in a loose slurry in a small hole and were then quickly Therefore, these pore pressures are not considered to be indicative of general pore pressure conditions in the embankment fill, but only provide an indication of the confined slurry pressure at the The high pressures are slowly dissipating and piezometer tip. illustrate the low permeability nature of the surrounding fill.

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pond level is shown on the summary plots for the foundation drain flows and the vibrating wire piezometer pressure plots, as discussed below.

b) <u>Measurement of the Foundation Drain flow quantities and sampling for water quality analyses.</u>

The flows in the foundation drains at the Main Embankment must be closely monitored to allow for correlations with other data (especially pond level) to be established. This is of particular importance during the initial filling of the Tailings Storage Facility when there are no tailings to help seal the basin.

A plot of the up to date foundation drain flows and pond level is shown on Figure 9.1. The plot shows that the foundation drain flows have continued to remain low, with a total flow below 20 litres/minute, even though the water level in the pond is rising. This indicates that the impounded water has not influenced the underlying soils and that the glacial till liner (natural and constructed basin liner) is working.

In addition to monitoring the foundation drain flows, samples should also be collected for water quality testing. Sampling and testing have already been initiated and should be continued quite frequently prior to mill start-up so that consistent baseline data are established. It is recommended that sampling and testing be continued until at least three consistent sets of results have been collected.

c) Monitoring of the Chimney Drain outlets.

The three outlet drains for the Main Embankment Chimney Drain end at the downstream face of the embankment. They will be extended to the Drain Monitoring Sump as part of Stage II construction in 1998. These drains must also be closely monitored to allow for correlations with other data (especially pond level) to be established. Any outlets installed in the Perimeter Embankment Chimney Drain will be similarly monitored.





h) <u>Sampling of surface water streams down gradient of the facility for water quality analyses.</u>

The surface water streams down gradient of the facility will be sampled for water quality analyses on an on-going basis. As discussed above, sampling and testing should be continued frequently until mill start-up so that consistent baseline data are established.

i) Sampling of process water in the tailings pond and seepage recycle ponds for water quality analyses.

After tailings discharge has commenced, process water in the tailings pond and seepage recycle ponds should be sampled for water quality analyses. The results can be compared to those from groundwater quality monitoring wells.

j) <u>Flow monitoring in diversion ditches, runoff collection ditches, and Polley</u> <u>Lake Pumping System.</u>

This is required for the detailed on-going evaluation of the project water balance and can be used to help predict the amount of make-up water needed from Polley Lake.

k) <u>Meteorological (rain, snow, evaporation) and air quality data collection.</u>

Meteorological data is important for the water balances and can be used to help predict the amount of make-up water needed from Polley Lake. This data is being collected on an on-going basis.

Instrumentation details are shown on Dwg. Nos. 1625.220 and 221 and the frequency of monitoring for the instrumentation is shown on Table 9.1. The instrumentation systems will be upgraded as required during each staged expansion of the tailings embankment. The locations and details of additional piezometers and surface movement monuments will be established during the detailed design of each staged expansion.





Monitoring is being continued on a weekly basis. The piezometer leads will be extended to the instrumentation monitoring huts after the foundation preparation work required for future embankment raises has been completed. All new piezometer installations for future embankment raises will be similarly monitored.

e) Monitoring of Surface Movement Monuments and Survey Control Points.

Surface movement monuments are required on the crests of the embankment each year to monitor vertical and lateral movement of the earthfill dams. End of construction and beginning of construction embankment crest surveys will also be carried out for each staged expansion raise.

Survey control points established for Stage Ia/Ib construction will act as survey control for all features at the Tailings Storage Facility, including the staff gauge for monitoring the pond level and the surface movement monuments on the embankment. New survey control points will be established as required.

f) <u>Monitoring of water levels in groundwater monitoring wells.</u>

A total of 11 monitoring wells were installed at six locations around the perimeter of the Tailings Storage Facility in late 1996, as shown on Figure 9.8. Monitoring of the water levels has been initiated and will be continued on a monthly basis. In addition, well MP89-234 will be monitored.

g) Sampling of groundwater monitoring wells for water quality analyses.

A total of 11 monitoring wells were installed at six locations around the perimeter of the Tailings Storage Facility in late 1996, as shown on Figure 9.5. Sampling for water quality analyses has been initiated and will be continued frequently prior to impounding tailings so that consistent baseline data are established.



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Routine Inspections

Routine inspections provide, to the maximum extent possible, continuous surveillance of the facility and should be done on a weekly basis.

Routine inspections are very important during the initial filling of the Tailings Storage Facility. The downstream slope and toe of the Main Embankment should be inspected in detail and any signs of seepage noted, such as wet areas, seepage boils, flows from the Chimney Drain outlets, etc.

Intermediate Inspections

Intermediate inspections of the civil/structural works and mechanical equipment should be carried out on an annual basis, or as required if abnormal conditions or a particular hazard arises.

Equipment Tests

Intermediate inspections of pumps should be conducted on a regular basis with valve inspections occurring twice a year, or as required, to ensure effective operation.

Special Inspections

Special inspections are required during initial reservoir filling and may be required following a major flood, earthquake or other event which may have affected the operation of the equipment or damaged the facilities.

Special inspections should be conducted by the Design Engineer. Special inspections will include all earthfills and cut slopes, the drain monitoring sumps, an on-site review of all monitoring data and an overall site inspection of all components associated with the Tailings Storage Facility.

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1627/2



An "Operation, Maintenance and Surveillance Manual for Stage Ia Embankment El. 927 metres" was previously issued (Knight Piésold Ref. No. 1627/1), and was a requirement prior to commencing water impoundment in the Tailings Storage Facility. A revised manual "Operation, Maintenance and Surveillance Manual for Stage Ib Embankment El. 934 metres" will be issued (Knight Piésold Ref. No. 1627/3) prior to start-up.

The main objectives of the Operation, Maintenance and Surveillance Manuals are to provide operational procedures and to detail monitoring systems which will be implemented both prior to start-up and during operations. In particular, the various components of the project water management plan will be identified and specific operational and monitoring requirements will be itemized so that the objective of storing sufficient runoff water in the tailings facility for mill operations is maintained, while simultaneously preventing the accumulation of excess water in the tailings impoundment during on-going operations.

The frequency and testing requirements for the water quality monitoring (groundwater monitoring wells, surface waters, tailings water) are being evaluated and will be finalized as required to meet the all guidelines required by government agencies.

9.3 <u>INSPECTIONS</u>

All components of the embankment and associated structures should be regularly inspected to ensure the serviceability and integrity of the Tailings Storage Facility. The type and scope of inspections were originally presented in the "Operation, Maintenance and Surveillance Manual for Stage Ia Embankment El. 927 metres" (Knight Piésold Ref. No. 1627/1) and are summarized below. All inspections are to be carried out by suitably qualified individuals as described in the manual. The inspection requirements will be reviewed and presented in the revised "Operation, Maintenance and Surveillance Manual for Stage Ib Embankment El. 934 metres".





In addition to the special inspection required prior to start-up and after a flood or earthquake, increased site surveillance should also be initiated if the following conditions or circumstances are observed:

- Abnormally high piezometric levels.
- Settlement, cracks and/or slumping of the embankment.
- Sinkholes along slope(s) of the embankment.
- Failure or substantial movement of reservoir slopes.
- Slope failure or seepage flows from the embankment slopes.
- Increased or contaminated flow from foundation drain outlet pipes.
- Damage to any component of the facility.

The minimum frequency of inspections and tests is specified on Table 9.2.

9.4 CONTINGENCY PLANS

In the event that minor repairs are required at the Tailings Storage Facility, construction equipment should be available at the Mine. The minimum equipment requirements include an excavator, a grader and a bulldozer. The equipment will be used to repair any slumped or scoured areas along embankment slopes, or to construct other key cuts or fills etc. Construction materials should be available at the Tailings Storage Facility and at the Mine for use in repairing or remediating any damaged areas. Local stockpiles of filter sand, drain gravel, riprap, and glacial till should be made available for periodic maintenance or for emergency use. Sources of cement, plastic sheeting, filter fabric, miscellaneous pipework and spare parts etc. should also be available at the Mine.

The surficial glacial till on the right abutment (approx. Chainage. 16+40 to 18+70) is underlain by sandy sediments with some elevated pore pressures. There is a possibility that the pore pressures in this unit will increase as the Tailings Storage Facility is filled, especially during the initial filling with water prior to start-up. In the event that on-going monitoring and inspections indicate that uplift pressures are increasing (indicated by elevated pore pressures in the piezometers or evidence of





seepage or piping downstream of the embankment), it may be necessary to install a pressure relief trench in this unit downstream of the embankment.

The pressure relief trench will be designed as required. Materials needed to construct this pressure relief trench would likely include 6 inch corrugated polyethylene tubing (CPT), 8 oz. non-woven filter fabric, drain gravel and Type B filter sand. These materials should be available at the Mine or in the vicinity of the Tailings Storage Facility.

A second option to minimize any effects that increased uplift pressures could have is to construct a buttress on the downstream side of the Stage Ia/Ib Main Embankment. This work is already planned as part of the Stage II expansion in 1998 but could be undertaken earlier, if necessary. The design of the buttress, including material and placement Specifications, would be completed as required.

The Longitudinal Drains have been installed in dense natural ground at the left abutment of the Main Embankment. At the right abutment, the Longitudinal Drain was installed in fill just above the prepared ground surface. As a result, the Longitudinal Drains can function as foundation drains and some seepage should be expected.

In the event that seepage is observed from the Chimney Drain outlets on the downstream face of the Main Embankment, the outlets should be extended to the Seepage Collection Pond or the Drain Monitoring Sump. Extensions to the Drain Monitoring Sump are planned for Stage II construction and will require 6 inch corrugated polyethylene tubing (CPT), 8 oz non-woven filter fabric, drain gravel and Type B filter sand. These materials should be available at the Mine or in the vicinity of the Tailings Storage Facility. A riprap lined trench can be constructed should the flows require temporary redirection to the Seepage Collection Pond. Suitable materials should also be available at the Mine or in the vicinity of the Tailings Storage Facility.

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SECTION 10.0 - RECLAMATION

10.1 GENERAL

In accordance with requirements under the B.C. Mines Act and Health, Safety and Reclamation Code for British Columbia, the primary objective of the proposed Reclamation Plan will be to return the tailings impoundment to an equivalent premining use and capability. This comprises forested wildlife habitat that supports grazing, hunting, guiding, trapping and recreational uses. The following goals are implicit in achieving this primary objective:

- Long-term preservation of water quality within and downstream of decommissioned operations
- Long-term stability of the tailings impoundment
- Re-grading of all access roads, ponds, ditches and borrow areas not required beyond mine closure
- Removal and proper disposal of all pipelines, structures and equipment not required beyond mine closure
- Long-term stabilization of all exposed materials susceptible to erosion
- Natural integration of disturbed lands into surrounding landscape, and restoration of the natural appearance of the area after mining ceases, to the greatest possible extent
- Establishment of a self-sustaining vegetative cover consistent with existing forestry, grazing and wildlife needs

As an overall approach to achieving these objectives, the Reclamation Plan is sufficiently flexible to allow for future changes in the mine plan and to incorporate



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information obtained from ongoing reclamation research programs such as trial tailings re-vegetation plots.

10.2 TOPSOIL STOCKPILING

Soil surveys were completed prior to construction to determine soil characteristics, soil volumes and to select topsoil stockpile areas. Requirements for salvage and stockpiling of surficial materials from the tailings impoundment were subsequently developed. The preferred locations for topsoil stockpiling are shown on the Drawings. All topsoil stockpiles will be provided with runoff collection ditches and sediment control features.

10.3 DECOMMISSIONING AND CLOSURE

Testwork has confirmed that the tailings are non-acid generating. Therefore, the tailings surface will be decommissioned as a mixed forested/wetlands complex with a gradual transition towards a ponded area with an overflow spillway. The downstream face of the tailings embankments will be revegetated progressively during operations to the greatest extent possible, once the final toe position and slope have been established.

At mine closure, surface facilities will be removed in stages, salvaged and sold. The tailings delivery system will be dismantled and removed immediately following cessation of operations. The reclaim barge, pumps and pipeline will be utilized for supplementary flooding of the open pit and will then be dismantled and removed. The seepage collection ponds and recycle pumps will be retained for a period after closure until monitoring results indicate that tailings area seepage is of suitable quality for direct release to the environment. At that time, the seepage collection pond and pumps will be removed. The groundwater monitoring wells and monitoring piezometers in the tailings embankment will be retained for use as long term monitoring devices.

A dual level spillway will be constructed, with the lower level designed to accommodate the 1-in-200 year flood flows and the second capable of accommodating



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the Probable Maximum Flood (PMF) within the tailings basin. The lower spillway, which will include the outflow channel, will be constructed in competent ground adjacent to the South Embankment and will discharge to the Edney Creek north The elevation of this spillway and outflow channel will be tributary drainage. designed to establish a set water elevation over the tailings surface (approximately A secondary, or emergency overflow spillway section will be 15% coverage). designed to accommodate the PMF while maintaining sufficient freeboard within the impoundment. This secondary spillway is also required in the event that beavers, ice or debris block the lower spillway and outflow channel.

Before the final tailings impoundment flooding to the required pond elevation, the area along the final water level will be sculptured using conventional earthmoving equipment to create a series of small bays and channels which will become a margin environment conducive to the creation of waterfowl breeding and staging habitat. The tailings embankments and the upland portions of the exposed tailings beach will be covered with a layer of soil and revegetated with indigenous species of conifer and deciduous trees, and willow and marsh land grasses. The moist transition zone between the topsoiled beach and final pond will be revegetated as an early seral stage meadow, leading to aquatic tolerant, emergent and submerged species of plant. Native vegetation species that are accustomed to swampy areas will be utilized for these transition zones. Where necessary, the final tailings surface will be treated with amendments suitable for sustaining permanent growth. The shoreline will then be planted with native emergent plant species for cover. The expected species will be transplanted from nearby wetlands of a similar aspect and elevation or propagated from root cuttings, turf squares or offsets.

The advice of organizations such as the B.C. Fish and Wildlife Branch, Ducks Unlimited and local trappers/guided outfitters will be sought during final design.

Final seeding of the embankment slopes with grasses and legumes will provide a stable vegetation mat that resists erosion. Once open pit flooding is complete, the surface water diversion system will be dismantled to allow for natural runoff to be routed through the tailings area.

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SECTION 11.0 - REFERENCES

The following Knight Piésold documents provide background information to support this report:

- 1. Imperial Metals Corp. Mt. Polley Project, "Report on Geotechnical Investigations and Design of Open Pit, Waste Dumps and Tailings Storage Facility", February 19, 1990.
- Imperial Metals Corp. Mt. Polley Project, "Report on 1995 Geotechnical Investigations for Mill Site and Tailings Storage Facility, Ref. No. 1623/1, March 14, 1995.
- 3. Imperial Metals Corp. Mt. Polley Project, Report on Project Water Management, Ref. No. 1624/1, February 6, 1995.
- 4. Imperial Metals Corp. Mt. Polley Project, Groundwater Monitoring Program, Ref. No. 1624/2, June 3, 1996.
- 5. Imperial Metals Corp. Mt. Polley Project, Tailings Storage Facility, Design Report, Ref. No. 1625/1, May 26, 1995.
- Imperial Metals Corp. Mt. Polley Project, Tailings Storage Facility, Site Inspection Manual, Ref. No. 1625/2, May 26, 1995.
- 7. Imperial Metals Corp. Mt. Polley Project, Tailings Storage Facility and Ancillary Works, Part 10 Technical Specifications, Ref. No. 1625/3, March 25, 1995.
- Imperial Metals Corp. Mt. Polley Project, Tailings Access Road and Tailings/ Reclaim Pipelines, Part 6 - Technical Specifications, Ref. No. 1625/4, May 17, 1995.



- 9. Imperial Metals Corp. Mt. Polley Project, Manual on Sampling and Handling Guidelines for Determination of Groundwater Quality, Ref. No. 1625/5, May 19, 1995.
- 10. Imperial Metals Corp. Mt. Polley Project, Response to Review Comments on Tailings Embankment Design, Ref. No. 1625/6, January 25, 1996.
- 11. Imperial Metals Corp. Mt. Polley Project, Response to Review Comments on Groundwater Monitoring Program, Ref. No. 1625/7, September 12, 1996.
- 12. Imperial Metals Corp. Mt. Polley Project, Requirements and Specifications for the 1996 Groundwater Monitoring Program, Ref. No. 1625/8, September 12, 1996.
- 13. Mount Polley Mining Corporation, Mount Polley Project, Tailings Storage Facility, Operation, Maintenance and Surveillance Manual for Stage Ia Embankment (El. 927 m), Ref. No. 1627/1, March 11, 1997.
- 14. Imperial Metals Corp. Mt. Polley Project, Report on Geotechnical Investigations and Design of Open Pits and Waste Dumps, Ref. No. 1628/1, July 5, 1996.
- Imperial Metals Corp. Mt. Polley Project, Specification for Drilling, Monitoring Well Installations and Related Services, Ref. No. 1628/3, September 18, 1996.
- Mount Polley Mining Corporation, Mount Polley Project, 1996 Groundwater Monitoring Well Installation Program, Ref. No. 1628/4, February 17, 1997.
- 17. Mount Polley Mining Corporation, Mount Polley Project, Polley Lake Pumping System, Ref. No. 1628/5, February 19, 1997.



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Other references include the following:

- 1. Basham, P.W., Weichert, D.H., Anglin, F.M., and Berry, M.J., 1982. "New Probabilistic Strong Seismic Ground Motion Maps of Canada: A Compilation of Earthquake Source Zones, Methods and Results", Earth Physics Branch Open-File Report 22-33.
- 2. Cornell, C.A., 1968. "Engineering Seismic Risk Analysis", Bulletin of the Seismological Society of America, Vol. 58, No. 5, pp. 1583-1606.
- 3. Idriss, I.M., 1993. "Procedure for Selecting Earthquake Ground Motions at Rock Sites", Report to U.S. Dept. of Commerce, NIST GCR 93-625.
- 4. Seed, R.B. and Harder, L.F., 1990. "SPT-Based Analysis of Cyclic Pore Pressure Generation and Undrained Residual Strength", H. Bolton Seed Memorial Symposium, Vol. 2.
- Zarling, J.P., Nelson, W.G. and Ryan, W.L., November 1990. "Cold Regions Engineering Short Course", University of Alaska.

J. BROUWER

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Ken J. Brouwer, P.Eng.

Director

K. D. K. EMBREE

Ken D. Embree, P.Eng.
Senior Engineer



TABLE 2.1

MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

MEAN MONTHLY AND ANNUAL PRECIPITATION

Location:	Likel	y, B.C.	Mir	ne Site	Bark	erville
Elevation:	72	24 m	10	00 m	1:	265
Location:		36'N		30'N		' 4'N
	121°	' 32'W	121	° 35'W	121°	'31'W
	Mean (mm)	Std. Dev. (mm)	<u>Mean</u> (mm)	Std. Dev. (mm)	<u>Mean</u> (mm)	Std. Dev. (mm)
Jan	74.2	27.0	75.5	27.0	103.0	44.4
Feb	60.2	27.7	58.1	27.7	85.6	42.5
Mar	37.8	13.5	44.5	13.5	85.3	29.1
Apr	42.2	20.9	43.1	20.9	61.8	24.5
May	36.6	15.4	50.6	15.4	65.9	28.9
June	66.3	29.7	81.5	29.7	89.2	28.8
July	47.0	27.4	65.7	27.4	81.7	31.0
Aug	82.0	35.7	83.1	35.7	102.3	53.0
Sept	50.4	27.1	60.4	27.1	85.4	39.9
Oct	61.6	42.3	60.4	42.3	88.4	37.4
Nov	58.4	18.8	57.3	18.8	86.6	28.2
Dec	83.0	36.9	74.8	36.9	108.7	42.5
Annual	699.7	116.4	755	116.4	1043.9	112.7

Source:

Canadian Climate Normals, 1951-1980, Temperature and Precipitation Atmospheric Environment Service, Environment Canada.

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Knight Piésold Ltd. CONSULTING ENGINEERS MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

PRECIPITATION DETAILS USED IN ANALYSIS

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5/19/97 9:42

:VOB\REPORT\1627\2-TBL2-2.XLS				5/19/97 9:42	
DESCRIPTION		VAL	UE		
Lower Elevations (ie. TSF)					
Mean annual precipitation (mm)		75	5		
"Dry" annual precipitation (mm)		601			
"Wet" annual precipitation (mm)		909			
"Max." annual precipitation (mm)	1050				
"Min." annual precipitation (mm)	450				
Mean annual rainfall (mm)	451				
Mean annual snowfall (mm)	304				
Coefficient of variation	0.16				
Standard deviation (mm)	121				
Higher Elevations (ie. mill site, waste					
dumps, etc.)					
"Elevation" factor	1.07285				
Mean annual precipitation (mm)	810				
"Dry" annual precipitation (mm)	645				
"Wet" annual precipitation (mm)	975				
Coefficient of variation	0.16				
Standard deviation (mm)	130				
Proportions of Total Precipitation:					
Rainfall		0.6	50		
Snowfall	0.40				
Monthly Proportions of Precipitation:					
,	Rainfall	Proportion	Snowfall	Proportion	
	(mm)	as Rainfall	(mm)	as Snowfal	
Oct	48.3	0.11	12.1	0.04	
Nov	17.3	0.04	40.0	0.13	
	7.6	0.02	67.2	0.22	
Dec Jan	6.8	0.02	68.7	0.23	
·	6.0	0.02	52.1	0.17	
Feb	6.0	0.01	38.5	0.17	
Mar	24.2	0.01	18.9	0.06	
Apr	45.3	0.03	5.3	0.02	
May	43.3 81.5	0.18	0.0	0.02	
Jun	81.5 65.7	0.18	0.0	0.00	
Jul	83.1	0.13	0.0	0.00	
Aug Sep	58.9	0.18	1.5	0.00	
		0.10		5,45	
Total (mm)	450.7		304.3		





TABLE 2.3

MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

PROBABLE MAXIMUM PRECIPITATION

1 hour PMP	= 78 mm	= 78 mm/hour
6 hour PMP	= 88 mm	= 14.6 mm/hour
24 hour PMP	= 203 mm	= 8.5 mm/hour
10 day PMP	= 406 mm	= 1.7 mm/hr

Source:

Rainfall Frequency Atlas for Canada, W.D. Hogg, D.A. Carr, Supply and Services Canada 1985.

Note:

- 1. 24 hr. PMP value conservatively assumes an orographic factor of 1.5.
- 2. 10 day PMP value assumes a 10 day to 24 hour PMP ratio of 2.0.

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TABLE 9.1

MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT TAILINGS STORAGE FACILITY

FREQUENCY OF INSTRUMENTATION MONITORING

v. i.o. in concentration (2.1 v. let Gheet)					18/04/97 14:16
INSTRUMENTATION	No.	No.	Frequency	Monitored	Comments
	Installed	Working	of Monitoring	By ⁽³⁾	
STARE GATICE					
	•	_	Weekly	MPMC	
- Basın Filing			,		
PIEZOMETERS:					
Stage Ia/Ib Embankments	_				
Foundation	9	9	Weekly	MPMC	
- Praine	8	8	Weekly	MPMC	One piezometer at Perimeter
Embarkment Fill	6	∞	Weekly	MPMC	Embankment (D2-PE1-01)
DRAIN OUTLETS:					
Stage Ia/Ib Main Embankment					Chimney Drain ouners
Examplation Drains	4	4	Weekly	MPMC	to be extended to Drain
Chimney Drain Ourlets	m	9	Weekly	MPMC	Monitoring in Stage II.
SURFACE MOVEMENT MONUMENTS: (1)					Not yet installed.
Stave Ia/Ib Main Embankment	ю		4 times per year		
CTRUEY CONTROL POINTS: (2)					Survey Hubs established
Ctogo Io Il Main Embankment	2		2 times per year		for Stage Ia/Ib construction.
Stage 14/10 Main Elloainancia					

Notes:

- 1. Surface movement monuments used to monitor lateral and vertical displacements along slopes and crests of embankment.
 - 2. Survey control points to be established for Stage Ia/Ib construction will act as survey control for surface monuments.
- 3. Instrumentation to be monitored during initial construction (by Design Engineer) and during operations (by Dam Co-ordinator).



TABLE 9.2

MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT TAILINGS STORAGE FACILITY

MINIMUM FREQUENCY OF INSPECTIONS AND TESTS

IOB\REPORT\1627\12-TBI 9-2 ylsIShent1

J:\JOB\REPORT\1627\[2-TBL9-2.xls]Sheet1		18/04/97 15:13
	INSPECTION	FREQUENCY
ITEM	Weekly	Annually
1. INSPECTIONS:		
a. Routine	x	
b. Intermediate - earthfills - pipelines - civil - mechanical equipment - electrical equipment	x x x	x x
2. TESTS:		
a. Portable Generators / Pumps - no load - under full load		x x
b. Valves / Flowmeters		x
3. EPP TESTS *:		
a. Communications Tests		X
b. Operational Tests		x

^{*} EPP = Emergency Preparedness Plan

AROJECT MOUNT FOLLEY MI MOUNT POLLE

TAILINGS STORAGE FACILITY
MONTHLY WATER BALANCE - AVERAGE PRECIPITATION CONDITIONS

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Dally Ore and Tailings Throughput (tpd) = 17,808 Subds Content = 35%

Tallings S.G. ≈ 2.78
Water Content of Ore ≈ 4%

Tailings Initial Dry Density (t/m³) = 0.9
Tailings Final Dry Density (t/m³) = 1.1
Minimum Fresh Water Makeup = 2.4%
Open Ptt Groundwater Discharge (m³/mo) = 39.818

63.07 to:14

Catchment Areas: Total Tailings Facility Area (ha) = 233

Unprepared Area (tag) = 134.1 Upstream Undiverted Area (tag) = 61 Total Pit Area (tag) = 17.6 Downstream Area (tag) = 63.1 Pund Area (ha) = 48.6 Beach Area (ha) = 50.3

Milisite Area-disurbed (ha) = 20
Milisite Area-undisurbed (ha) = 38.9
Waste Dump-disurbed (ha) = 17.2
Warte Dump-undisurbed (ha) = 106.8
Area North of Milisite (ha) = 22.6

Tailings Pond = 1.0 Unprepared Basin = 0.24 Tailings Beach = 0.9 Runoff Coefficients;

Undisurhed Catchment Areas = 0.24
Milistic Area-disurhed = 0.70
East Waste Dump-disurhed = 0.66
Beach Evaporation Factor = 0.80
Downstream Area Factor = 0.70 Open Pit Area = 0.5

\$100 feet											Ή.	JOHNDATAN 620 WAS	EXHALXWRAL-9.XLS	.
DESCRIPTION	JUN	JUL	AUG	SEP	ocr	NOV	DEC	JAN	FEB	MAR	APR	MAY	ANNUAL	
A Rainfall (mm/month) B Snowfall (mm/month) C Evaporation (mm/month)	81.5 0.0 112.0	65.7 0.0 107.0	83.1 0.0 92.0	58.9 1.5 50.0	48.3 12.1 15.0	17.3 40.0 0.0	7.6 67.2 0.0	6.8 68.7 0.0	6.0 52.1 0.0	6.0 38.5 0.0	24.2 18.9 0.0	45.3 5.3 47.0	450.7 304.3 423.0	<u>< ≈ ∪</u>
 WATER INTO TAILINGS IMPOUNDMENT> (m²) With Slury Tailings Pond Presipitation Tailings Beach Runoff Undiversed Runoff From Within Tailings Facility Runoff from Upstream Undiversed Area Runoff from Upstream Area Waste Dump Runoff (Disturbed) Water Available From Polley Lake 	39,647 39,647 36,647 26,230 11,932 35,999 36,682 0	1,005,940 31,961 21,145 9,618 29,020 29,570 0	1,005,940 40,425 37,585 26,745 12,166 36,705 37,402 0	1,005,940 29,382 27,318 19,439 8,843 26,679 26,510 0	1,005,940 23,496 21,845 15,445 7,071 21,334 21,739 0	1,005,940 8,416 7,825 5,568 2,533 7,641 7,786 0	1,005,940 3,697 3,697 2,446 1,113 3,357 3,421 0	1,005,940 3,308 3,076 2,189 996 3,004 3,004 0	1,005,940 2,919 2,714 1,931 878 2,650 2,700 0 1,019,733	1,005,940 2,919 2,714 1,931 878 2,650 2,700 0 1,019,733	1,005,940 88,731 82,497 58,704 26,703 80,566 82,095 500,000	1,005,940 92,379 85,889 61,117 27,801 129,740 85,471 500,000	12,071,280 367,278 341,475 242,380 110,532 379,344 339,137 1,000,000 14,852,006	4 4 4 2 2 4 L 8
 < WATER OUT OF TAILINGS IMPOUNDMENT> (m³) Supernalant Recovery (+) Recovery from Tailings (+) Total Net Precipitation and Runoff =(2)+(3)+(4)+(5)+(5a)+(6-(18)-(1a)) (+) Consultation to Final Density (+) Consultation to Final Density (+) Sequage (-) Sequage (-) Sequage 	598,937 87,838 109,426 0 (63,940) 732,262	598,937 55,960 109,426 0 (63,940) 700,384	598,937 109,286 109,426 0 (63,940) 753,710	598,937 93,746 109,426 0 (63,940) 738,169	598,937 97,703 109,426 0 (63,940) 742,127	598,937 39,769 109,426 0 (63,940) 684,192	598,937 17,471 109,426 0 (63,940) 661,894	598,937 15,632 109,426 0 (63,940) 660,055	598,937 13,793 109,426 0 (63,940) 658,216	598,937 13,793 109,426 0 (63,940) 658,216	598,937 419,295 109,426 500,000 (63,940) 1,563,719	598,937 440,638 109,426 500,000 (63,940) 1,585,061	7,187,247 1,404,923 1,313,115 1,000,000 (767,280)	° 5 = 5 5 7
Underdralinage Recovery (+) Underdrainage (5) Sub-Total (total recovered water) = (14)+(15)	58,100	58,100	58,100 811,810	58,100 796,269	58,100 800,227	58,100	58,100	58,100 718,155	58,100	58,100	58,100	58,100	697,200 115	· 2 · 5
Unresovenble Water	297,576 54,484 45,027 5,840 402,928 1,193,290	297,576 52,051 43,017 5,840 398,485 1,156,969	297,576 44,754 36,987 5,840 385,158 1,196,968	297,576 24,323 20,102 5,840 347,841 1,144,110	297,576 7,297 6,030 5,840 316,744 1,116,971	297,576 0 0 5,840 303,416 1,045,709	297,576 0 0 5,840 303,416 1,023,411	297,576 0 0 5,840 303,416 1,021,572	297,876 0 0 5,840 303,416 1,019,733	297,576 0 0 5,840 303,416 1,019,733	297,576 0 0 5,840 303,416 1,925,235	297,576 22,864 18,895 5,840 345,176 1,988,337	3,570,917 205,773 170,059 70,080 4,016,830 14,852,036	25 1 2 2 2 2 2 2 2 3 2 3 2 3 2 3 2 3 2 3 2
Water Required at Milisite 23 Water for slury 24 Water for Dust Control on Roads 25 Wall Water Required 26 (-) Minimum Fresh Water Input to Mill (from open pit groundwater) = 2,4%*(1) 27 (-) Water in Ore Water Required from Additional Sources = (25)-(26)-(27)	1,005,940 25,000 1,030,940 24,143 21,982 984,815	1,005,940 25,000 1,030,940 24,143 21,982 984,815	1,005,940 25,000 1,030,940 24,143 21,982 984,815	1,005,940 25,000 1,030,940 24,143 21,982 984,815	1,005,940 25,000 1,030,940 24,143 21,982 984,815	1,005,940 0 1,005,940 24,143 21,982 959,815	1,005,940 0 1,005,940 24,143 21,982 959,815	1,005,940 0 1,005,940 24,143 21,982 959,815	1,005,940 0 1,005,940 24,143 21,982 959,815	1,005,940 0 1,005,940 24,143 21,982 959,815	1,005,940 0 1,005,940 24,143 21,982 959,815	1,005,940 25,000 1,030,940 24,143 21,982 984,815	12,071,280 150,000 12,221,280 289,711 263,784	24 27 37 72 82
 <water distribution="" in="" system=""> (m²)</water> Open Pi Surface Runuf Open Pi Surface Runuf Open Pi Groundwater (30,818 - fresh water input to mill) Mill Site Runoff (Disturbed and Undisturbed) Cachiment Area North of Milishe Recovered Water from Tallings et Pallity (excluding storage) Recovered Water from Tallings et Troil Wuter Available in the System Water Surplus/(Deficit) = (34)-(28) Cammulative Water Surplus/(Deficit) 	7,694 15,675 20,404 4,743 790,362 838,879 (145,937)	6,203 15,675 16,449 3,823 738,484 800,634 (184,181) (330,118)	7,846 115,675 20,805 4,836 811,810 860,972 (123,844)	5,702 15,675 14,746 3,427 796,269 833,821 (148,994) (602,956)	4,560 15,675 12,092 2,811 800,227 835,365 (149,450)	1,633 15,673 4,331 1,007 742,292 764,339 (194,876)	718 15,675 1,903 442 719,994 738,732 (221,083)	642 15,675 1,702 396 718,155 736,571 (223,245)	566 15,675 1,502 349 716,316 734,409 (225,406) (716,719,11)	566 15,675 1,502 349 716,316 734,409 (225,406) (1,842,423)	17,221 15,675 45,666 10,614 1,621,819 1,710,995 751,179 (1,091,244)	17,929 15,675 47,543 11,051 1,643,161 1,735,359 750,544 (340,700)	71,280 188,105 188,646 43,847 10,835,206 11,327,085 -340,700	88288488

1. Snowfall is provided in equivalent depth of rainfall and is assumed to accumulate on carehment areas until April and May when it melts equally over the two months.

2. Fresh water imput to mill so he supplied from Open Pit dewatering wells.

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Notes:

JRFORATION AOJECT TABI MOUNT POLLEY MII MOUNT POLLA.

TAILINGS STORAGE FACILITY
MONTILLY WATER BALANCE - AVERAGE PRECIPITATION CONDITIONS
YEAR 13

Catchment Areas:

CONSHITINGENGINEERS

Daily Ore and Tailings Throughput (pdd) = 17,808 Solids Content = 35% Tailings S.G. = 2.78

Knight Piésold Ltd

Total Tailings Facility Area (ha) = 233

Pond Area (ha) = 101

Beach Area (ha) = 122

Unprepared Area (ha) = 16

Upstream Undiverted Area (ha) = 64

Downstream Area (ha) = 64.1

(200 lgpm)

Open Pit Groundwater Discharge (m³/mu) = 39,818 Tallings Initial Dry Density $(t/m^3) = 0.9$ Tallings Final Dry Density $(t/m^3) = 1.3$ Minimum Frest Water Makeup = 2.4% Water Content of Ore = 4%

Millsite Area-uistarbei (ha) = 20 Millsite Area-undistarbei (ha) = 39 Waste Dump-distarbei (ha) = 134 Waste Dump-andistarbei (ha) = Area North of Millslte (ha) = 24.3

Runoff Coefficients:

Tailings Porul = 1.0 Unprepared Basin = 0.24 Tailings Beach = 0.9

Millshe Area-disturhed = 0.70 Undisturbed Catchment Areas = 0.24 Open Pit Area = 0.5

Beach Evaporation Factor = 0.80 Downstream Area Factor = 0.70 East Waste Dump = 0.60

65.7 83.1 58.9 48.3 0.0 0.0 1.5 12.1 107.0 92.0 50.0 15.0 1.005,940 1,005,940 1,005,940 1,005,940 1,566 72,316 1,974 1,435 1,148 9,618 12,166 8,843 1,011 22,020 36,703 26,639 41,662 0 0 0 0 1,224,352 1,303,698 1,221,067 1,179,005 1 122,867 115,009 115,807 21,334 223,867 115,009 115,807 143,269 185,183 185,183 185,183 185,183 0 0 0 0 1,224,352 1,303,698 1,221,067 1,179,005 1 221,800 38,100 38,100 58,100 58,100 801,147 893,290 835,987 863,449 14,020 32,000 15,000 15,100 1,005,940 1,005,940 1,005,940 1,005,940 1 22,800 25,000 25,000 25,000 25,000 1,005,940 1,005,9	WANT IN IN											N.F	JAKIBIDATAMBIRIWATIBIRALAWBAL-B.KI S	SHALIWRAL-S.XI.
81.5 65.7 83.1 58.9 48.3	SCRIPTION	JUN	Tor	AUG	SEP	ocr	NOV	DEC	JAN	FEB	MAR	APR	MAY	ANNUAL
1,005,940	fall (mn/month) wfall (mn/month) poration (mn/month)	81.5 0.0 112.0	65.7 0.0 107.0	83,1 0.0 92.0	58.9 1.5 50.0	48.3 12.1 15.0	17.3 40.0 0.0	7.6 67.2 0.0	6.8 68.7 0.0	6.0 52.1 0.0	6.0 38.5 0.0	24.2 18.9 0.0	45.3 5,3 47.0	450.7 304.3 423.0
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88,100	ATER OUT OF TAILINGS IMPOUNDMENT> (m²) sentitud Recovery (+) Recovery from Tailings (+) Total Net Precipitation and Runoff =(2)+(3)+(4)+(5)+(5a)+(6)-(18) (+) Water Available From Polley Lake (+) Serpuge (5) Serpuge (-) Serpuge		598,937 22,867 185,183 0 (63,940) 743,047	598,937 115,009 185,183 0 (63,940) 835,190	598,937 115,807 185,183 0 (63,940) 835,987	598,937 143,269 185,183 0 (63,940) 863,449	598,937 61,988 185,183 0 (63,940) 782,168	598,937 27,232 185,183 0 (63,940) 747,412	598,937 24,365 185,183 0 (63,940) 744,545	598,937 21,499 185,183 0 (63,940) 741,679	598,937 21,499 185,183 0 (63,940) 741,679	598,937 653,563 185,183 0 (63,940) 1,373,743	598,937 632,937 185,183 0 (63,940) 1,353,117	7,187,247 1,909,584 2,222,195 0 (767,280) 10,551,746
121,820 221,820 221,820 221,820 121,820 112,896 107,856 22,736 50,400 15,120 105,840 5,840 5,840 5,840 5,840 5,840 5,840 5,840 5,840 5,840 5,840 24,143 24,	igridrainage Recovery (+) Underdrainage Total (total recovered water) =(14)+(15)	58,100	58,100	58,100	58,100	58,100 921,549	58,100 840,268	58,100 805,512	58,100 802,645	58,100 977,997	58,100 779,779	58,100 1,431,843	58,100	697,200 11,248,946
1,005,940 1,005,	Tailings Supernatum Pond Beach	1	221,820 107,856 104,689 5,840 440,205 1,241,352	221,820 92,736 90,013 5,840 410,409 1,303,698	221,820 50,400 48,920 5,840 326,980 1,221,067	221,820 15,120 14,676 5,840 257,456 1,179,005	221,820 0 0 0 5,840 227,660 1,067,928	221,820 0 0 5,840 227,660 1,033,172	221,820 0 0 5,840 227,660 1,030,305	221,820 0 0 5,840 227,660 1,027,439	221,820 0 0 0 5,840 227,660 1,027,439	221,820 0 0 5,840 227,660 1,659,503	221,820 47,376 45,985 5,840 321,021 1,732,238	2,661,838 426,384 413,863 70,080 3,572,165 14,821,111
FION IN SYSTEM > (m²) 28,286 22,802 28,841 20,963 16,763 15,675 15,675 15,675 15,675 20,404 16,449 20,805 14,746 12,092	er Required at Milisite or for slury or for Dast Control on Roads or for Dast Control on Roads Mil Water Required fininum Fresh Water Input to Mill (from open pit groundwater) = 2.4%*(1) Vater in Ore Water Required from Additional Sources = (25)-(26)-(27)	1,005,940 25,000 1,030,940 24,143 21,982 984,815	1,005,940 25,000 1,030,940. 24,143 21,982 984,815	1,005,940 25,000 1,030,940 24,143 21,982 984,815	1,005,940 25,000 1,030,940 24,143 21,982 984,815	1,005,940 25,000 1,030,940 24,143 21,982 984,815	1,005,940 0 1,005,940 24,143 21,982 959,815	1,005,940 0 1,005,940 24,143 21,982 959,815	1,005,940 0 1,005,940 24,143 21,982 959,815	1,005,940 0 1,005,940 24,143 21,982 959,815	1,005,940 0 1,005,940 24,143 21,982 959,815	1,005,940 0 1,005,940 24,143 21,982 959,815	1,005,940 25,000 1,030,940 24,143 21,982 984,815 '	12,071,280 150,000 12,221,280 289,711 263,784 11,667,785
Carcinment Area North of Milisite Recovered Water from Tallings Facility (excluding storage) Recovered Water from Tallings Facility (excluding storage) Recovered Water Available in the System State Surplus/(Deficity) (67,522) (192,153) (192,153) (21,005) (248,816) (264,528) (264,528)	ATER DISTRIBUTION IN SYSTEM > (m²) PH Surface Runoff Thi Surface Runoff The Connabaster (39,818 - fresh water input to mill) Site Runoff (Disturbed and Undisturbed) Site Runoff (Disturbed) England	28,286 15,675 20,404 5,099 847,829 917,294 (67,522)	22,802 15,675 16,449 4,111 801,147 860,184 (124,631) (192,153)	28,841 15,675 20,805 5,199 893,290 963,811 (21,005)	20,963 15,675 14,746 3,685 894,087 949,157 (35,658) (248,816)	16,763 15,675 12,092 3,022 921,549 969,103 (15,713) (264,528)	6,004 15,675 4,331 1,082 840,268 867,362 (92,454) (356,982)	2,638 15,675 1,903 476 805,512 826,203 (133,612)	2,360 15,675 1,702 425 802,645 822,809 (137,007) (627,601)	2,082 15,675 1,502 375 779,779 819,414 (140,401) (768,002)	2,082 15,675 1,502 375 799,779 819,414 (140,401)	63,305 15,675 45,666 11,413 1,431,843 1,567,902 608,087 (300,316)	65,908 15,675 47,543 11,882 1,411,217 1,552,226 567,410 267,094	262,036 188,105 188,646 47,145 11,248,946 11,934,879 267,094

og it melts equally over the two months. Snowfall is provided in equivalent depth of rainfall and is assumed to accumulate on catchment areas until April and May.
 Fresh water imput to mill to be supplied from Open Pit dewatering wells.

Notes:

MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

RECORD TEST SUMMARY SHEET FOR ZONE B

	T	T	Ī		RBa	LAEP		s/ms	8.9E-10	9,96-10	4.5E-10	. :	3.15-09	3E-09	1	2.615-05		6.2E-10	4.3E-10		.31-0.		1.85-09	1 5	7.8b·10			90	2	-	1	1	i		<u> </u>			36.00	00.35	9 21	2 88.10	
-		r. 13/97		nt - Zone B	R6	S.G.	•		2.74	<u> </u>	 	1 1	f.	-:		<u></u>		اُف	4	i •	<u>-</u> 1	1	<u>- </u>		7	<u>. </u>	1	<u> -</u> 	Ī		<u>ا</u> تر	-	<u>:</u>	1	<u>.</u>	<u> </u>	1	╬	÷	1 2	┿	┪
-	-	28/96 to Ma	: 1627	Main Embankment - Zone			<u>=</u>	,	L	<u> </u> 	<u>i</u> 	! !	9'6	10.0	6'01	6	<u> </u>	10.4	10.0	0.0	14.2	<u> </u>	<u>.</u>	- L	.,	٠		 			 -	<u> </u>			1			╁	+	╁	+	$\left\ \cdot \right\ $
		PERIOD: Aug. 28/96 to Mar. 13/97	ECT NO.:	- 1	R4	Standard Proctor	Pry Opi		ľ	_	<u> </u>	H	4	+	+	<u>.</u>	1	+	<u> </u>	<u> </u>	<u> </u>	+	+	+	1	<u> </u>	\perp	+		+	<u> </u>	Ļ	1	_	1	+	6	╀	5,0	ł	+	-
Carre	14120	PER	PROJECT	AREA:			Ē	y Density	F	1	<u>'</u>	; <u>}</u>	- 1	÷	+		1	***	+	+		+	-	÷	-	+	÷	÷	0.000	÷	÷	<u>+</u>	÷	÷	÷	\pm	2130.0		t	2003.0	+	╣
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						1.19126	0.0469	91 <i>#</i>	72.8	29.6	8.86	2	29.0	79.0	79.5	77.9	9.4	76.9	78.7	10.1	6.23	72.4	6.19	88	7	65.2	71.3	71.4	2.5	20.0	3, 5	8 8	6	73.2	869	3.	88	74.0	73.5	77.0	6.5	4.29
					5	3 2,37998	0.0937	83	75.6	82.4	8 18	78.7	82.2	83.0	-	+	24.9	+	+	+	+	1	+	+	+	+	+	+	+	÷	┿	+	+	+	+	+	+	╬	┿	+	+	29.7
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1	S S.	VT ZON				76.2	-	3	0 001	+	÷	•	100.0	100.0	100.0			0.001	0.001	+	100.0	100.0	+	-+	╍┼	+	+	+	+	+	+	+	1	 	0.00	<u></u>	_	4	+	+	+	0.001
	RECORD LEST - SUMMARICE SHEET	EMBANKMEN'T ZONE B RECORD SAMPLES				H	26	Max	╬╌	Ť	1	1	97.2	<u> </u>			97.9	99.2	9.6	+	101.6	÷	-	-+	÷	$\frac{}{}$	1	\pm	+	÷	1	1	t	+	98.6	17.1	-	╁	┿	t	٦,	
	<u> </u>	IN EMBA			EZ EZ	R7h	Nac	m.c.	١.	2 5	- 2	1 1	9.01	12.2	8.1	2	12.1	11.9	10.3	9.5	12.0	6.6	9,6	9.3	10.4	8.6	8.0	9.5	10.4	9.8	2	8.5	=	9.4	8.6	8.5	10,0	8.6	10.5	10.2	13.8	8.0
		MAI				R73	ρΩ	Density 1007	1034.0	0.5001	0.000	2025.0	2032.0	2044.0	2027 0	2070.0	2026.0	2047.0	2027.0	2028.0	945.0	2110.0	0.7261	2170,0	2126.0	2180.0	2006.0	2050.0	2120.0	2090.0	2120.0	2120.0	2000	2090.0	2110.0	2030.0	2110.0	2120.0	2058,2	2048.5	2180.0	1934.0
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		į	No		22	Field	B,C	Þ¢		2.5	9.71	2 2	10.7	12.4	11.7	1.5	11.7	11.5	13.2	1.1	13.1	8.6	8.6	9.7	11.2	10.2	10.4	9.3	10.5	<u>.</u>	10.6	9.5	8.2	10.2	9.7	9.5	10.0	=	10.9	5.0	13.7	8.3
			NSTRUCT			1	<u>=</u>	1%		1.1	5.0	10.2	8.7	10.8	4.11	10.9	9.3	10.4			8.01	8.3	8.7	8.0	7.2	2.5	4.1	3,4	7.7		8.0	6.3		5.9	6.3		6.8	0.0	8.0	8.0	11.7	2.5
			AGE IS CO		Ī	Anerberg Limits	בו	35		ė.	3, 5	24.5	22.9	25.0	25.6	25.6	24.5	25.1			27.0	21.5	26.7	22.0	22.0	19.4	8.61	19.2	21.7	17.7	22.2	21.2	15.3	30.8	20.8	8,8	21.5	21.0	22.3	22.0	27.0	15.3
			LITY - ST.			ľ	2	84		0.61	14.3	5 	14.2	14.2	5,	14.7	15.2	14.7			16.2	13.2	18.0	14.0	14.8	16.9	15.7	15.8	14.0		14.2	14.9		14.9	14.5	호	14.7	15.0	14.9	7.7	0.81	13.2
			AGE PAC			Neg.	â	Ī	,	915.0	917.0	916.4	917.0	917.5	919.6	920.3	916.6	918.5	929.0	922.0	921.0	924.0	927.0	924.5	923.0	926.0	923.0	925.4	925.7	927.5	921.5	924.0	929.0	931.0	928.0	927.5	929.2	930.0				
			NGS STOR	Tests	3	5	•			Shoulder	5 5	CAL	Shoulder	Shoulder	Shoulder	Shrulder	Shunder	Shoulder	Shloulder	Q/1,	Shoulder	Shoulder	Shoutder	Shoulder	Shoulder	7.D	Q,	Q,	COD	CO	ÇD	Shoulder	Q)	Shoulder	CCD	c _D	C.D	of CL.				
Š	Ltd.	KHS	Y - TAILII	B Record	ACITA OCI	1000	1			Zam N of Das Shoulder	20m S of C/L	20m N of D/S 1 ms	5m N of D/S Shoulder	13m N of D/S Shoulder	27m N of D/S Shoulder	25m N of D/S Shmilder	8m N of D/S Shualder	15 N of D/S Shoulder	Sm N of D/S Shloulder	du N of L/D	13m N of D/S Shoulder	10m N of D/S Shoulder	5m N of D/S Shoulder	3m N of D/S Shoulder	8m N of D/S Shoulder	N of C/D	NufCID	SofC/D	5m D/S uf C/D	4m D/S of C/D	6m D/S of C/D	8m N of D/S Shoulder	3m N of C/D	5m N of D/S Shoulder	4m D/S of C/D	Sm N of C/D	4 m S of C/D	3 m D/S			£	٠
Data Sheet	Knight Piésold Ltd.	CONSULTING ENGINEERS	MOUNT POLLEY - TAILINGS STORAGE PACILITY - STAGE IS CONSTRUCTION	Glacial Till Zone B Record Tests		Cheimagn	A (m)	 [7	Т	- i	21+00	1	Ī	ī	i .		i i	16+60 5	Ė	18+50	18+50	17+75 51	18+50 3	19+90 8	18+75	06+61	19+25	30+0¢	20+20	22+20	 i	21+50	21+10 5	22+25	23+00	23+25	24+50	MEAN	MEDIAN	MAXIMUM (*)	MINIMUM (*)
J-UOBEREPORT 1627/2-f6/610. xlx/Data Sheet	Knigh	(TONS)	MO	5		S GAMPI			╬	4	÷	R/ME/ZB-3	<u> </u>	!	<u>:</u>	_	<u>!</u>	<u>!</u>	<u>!</u>	-	RVME/ZB-13		R/ME/ZB-15			R/ME/ZB-18	R/ME/ZB-19	R/ME/ZB-20	R/ME/ZB/21	R/ME/ZB-22	R/ME/ZB-23		R/ME/ZB-25	R/ME/ZB-26	_	<u> </u>		R/ME/ZB-30			2	
PORT\1627			Ţ	'AL.	-				ᅪ	<u> </u>	+	+	1	_	÷	ı	<u> </u>	<u> </u>	-		-	-		1					_		-	÷				 -		-				
J:VOBNRE			PROJECT	MATERIAL		DATE	CAMPI ED			28-Aug-96	22-Aug-96	01-Sep-96	28-Sen-96	10-00-96	96-12O-11	12-0-1-96	17-0-1-96	19-0-1-96	04-Feh-97	05-Feb-97	06-Feh-97	09-Feb-97	08-Feb-97	09-Feb-97	13-Feh-97	13-Feh-97	13-Feh-97	13-Fch-97	13-Feh-97	13-Feb-97	13-Feh-97	13-Feh-97	13-Feb-97	13-Fcb-97	13-Feh-97	13-Feh-97	14-Mar-97	13-Feb-97				

COMMENTS: These are 100% limits.

R1 R2 R3 R4 R7 R7 R7 R7 R7

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Atterberg Limits (ASTM D4318)
Moisture Content (ASTM D2216)
Particle Size Distribution (ASTM D422)
Laboratory Compaction (ASTM D1557)
Specific Gravity (ASTM D854)
Field Density by Nuclear Methods (ASTM D2922)
Moisture Content by Nuclear Methods (ASTM D3017)
Lab Air Entry Permeameter (LAEP)

MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

SUMMARY OF CONSOLIDATED-UNDRAINED TRIAXIAL TESTS FOR STABILITY ANALYSES

J:VOB\DATA\1627\SLOPEW\APRIL97\TABLE.XLS

5-Jun-97

-			Sample No.	le No.	
Triaxial Testing Stages	Units	Units TP95-27 TP95-37 TP95-37 TP95-37	TP95-37	TP95-27	TP95-37
		(Test 1)	(Test 2)	(Test 1) (Test 2) (Test 3) (Test 4)	(Test 4)
Consolidation Stage					
Final effective consolidation pressure, $p' = \sigma_{3c}'$	kPa	287	498	745	962
Shearing Stage					
Maximum deviator stress, (σ _d ') _{max}	kPa	1416	1268	812	1624
Undrained Shear Strength parameter, $C_{u} = (\sigma_d')_{max}/2$	kPa	708	634	406	812
Undrained Shear Strength Ratio, Cu/p'		2.47	2.47 1.27	0.54	0.84

Note

- 1. For the conservative assumption, Undrained Shear Strength of 0.5 was adopted in the stability analyses.
 - 2. The consolidated-undrained triaxial tests information abstracted from Report on 1995 Geotechnical Investigations for Mill Site and Tailings Storage Facility, (Report 1623/1)

TABLE 6.8

MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

CONTROL TEST SUMMARY SHEET FOR FUTURE BORROW AREA

IRAMAY	l of 1	D: 13-Sep-96 to 24-Oct-96	PROJECT NO.: 1627	: Future Borrow Area (east of PE)	C4 C6 C8a	Standard Proctor (1) S.G. LAEP	ry Opt.	y m/c cm/s														-
	SHEET	PERIOD:	PROJE	AREA:			Max Dry	Density kg/m³												<u> </u>		_
						0.00	quanh (Clay				5.0				3.0			4.0	0.4	5.0	
						0.074	9 0.0029	#200			37.5	42.2		55.0		38.1			43.2	40.2	55.0	
						0.150	5 0.0059	#100			45.8	50.5		67.5	-	46.3			52,5	48.4	67.5	
						0.419	4 0.0165	#40			56.7	61.2		80.3		56.9	_	_	63.8	59.1	80.3	
						0.594	9 0,0234	#30			59.7	63.7		83.3		59.8		_	9.99	61.8	83.3	
	HEET					1.191	7 0.0469	#16			63.8	67.7		87.2		64.2		_	7.07	66,0	87.2	
	CONTROL TEST - SUMMARY SHEET	REA		,	ខ	2.380	0.0937	8#			67,1	71.9		8.		8.89			74.6	70.4	93.4	
	SUMM	RROW A				4.750	0.187	##			79.1	76.7		93.4		75.0			78.8	75.9	93.4	
	rest.	FUTURE BORROW AREA				9.525	0.375	3/8			75.7	81.3	-	96.2		79.5		_	83.2	80.4	96.2	
	rrol.	FUT		PE)		19.05	0.75	3/4			81.9	88.0		98.6		83.6			88.0	85.8	98.6	
	NO3			ASTOF		25.4	-				84.6	91.3		0.00		86.7			7.06	89.0	100.0	
				A AREA E		38.1	1.5	711	-	ĺ	87.3	8		100.0		88.7		_	77.7	91.8	100.0	
			UCTION	IGATION		76.2	3	<u></u>			100.0	100.0		100.0		100.0			100.0	100.0	100.0	
		•	CONSTR	NVEST				86			-											
			STAGE IS	FER SAN	ខ	Field	ш/с	₽¢	14.7	15.4	10.2	0.0	12.6	10.6	13.0	9.7	14.9	9.2	12.0	11.6	15.4	
			HLITY .	EA (FIL		Ē.	ā	88				.										
			AGE FAC	RROW A	J	Atterberg Limits	13	26														
			GS STOR	TURE BO		And	E	路														
	#4.	,	-TAILIN	ROM FU		Depth	£		(0.3-0.5)	(0.8-1.0)	(2.0-4.0)	(5.0-5.4)	(0.7-1.0)	(3.0-3.5)	(1.0-1.5)	(2.5-3.0)	(0.7-1.2)	(2.5-3.0)	MEAN	MEDIAN	MAXIMUM (*)	
heet	Knight Piésold Ltd.	CONSULTING ENGINEERS	MOUNT POLLEY - TAILINGS STORAGE FACILITY - STAGE IS CONSTRUCTION	GLACIAL TILL FROM FUTURE BORROW AREA (FILTER SAND INVESTIGATION AREA EAST OF PE)		Ē	SAMPLE	No.	ET-96/10/24-1	E		•	ET-96/10/24-2 (0.7-1.0)	•	ET-96/10/24-3 (1.0-1.5)	•	ET-96/10/24-4 (0.7-1.2)	•			MAXII	
J:VOBNREPORT\1627\[2-tb16-8.xis]Data Sheet	Knig	CON				CONTROL	SAMPLE	No.	C/ME(FB)/(ZS,ZB)-1A	C/ME(FB)/(ZS,ZB)-1B	C/ME(FB)/(ZS,ZB)-1C	C/ME(FB)/(ZS,ZB)-1D	C/ME(FB)/(ZS,ZB)-2A	C/ME(FB)/(ZS,ZB)-2B	C/ME(FD)/(ZS,ZB)-3A	C/ME(FB)/(ZS,ZB)-3B	C/ME(FB)/(ZS,ZB)-4A	C/ME(FB)/(ZS,ZB)-4B				
FLIOBUREPO			PROJECT :	MATERIAL		DATE	SAMPLED		24-Oct-96	•		•				•						

These are 100% limits. ទិដិដី៥៥៥៥៤៤€

Atterberg Limits (ASTM D4318) Moisture Content (ASTM D2216)

Particle Size Distribution (ASTM D422)

Laboratory Compaction (ASTM D1557) Specific Gravity (ASTM D854)

Field Density by Nuclear Methods (ASTM D2922)

Moisture Content by Nuclear Methods (ASTM D3017) Lab Air Entry Permeameter (LAEP)

TABLE 6.9

MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

RECORD TEST SUMMARY SHEET FOR ZONE S

	ì
Sheet	
mrQlS1	
X-6-97	
742-TB	
KTA162	
REPO	ŀ

Part	31,16	JOBNEPORTMESTUS-TRL6-9-XLSIDau Sheet	LSjbata Sjeed							ı								,										18/04/97
Part		Kni	ight Piésa	old Led.									RECC	ORD TES	ST-SUN	MMAR	Y SHEE	_ ــا							SHEET		1 of 1	
NATIONALIS NAT		5	ONSHITTING ENG	GINEERS							•	MAIN	'EMBA	NKMEN	TZON	ESREC	CORD S.	AMPLE	Š						PERIO	: 22-Aug-96	22-Aug-96 to 19-Oct-96	96-13O-61 o
Charge C			MOUNT P.	OLLEY - TAILINGS	STORAGE	FACILIT	Y - STAG	E Ib CON	STRUCTIC	N															PROJE	T NO.:	1627	
Column C			Claciat Till	Zone S Record Tests										i											AREA:	! !	nkment - Zon	c S
Chinge Original				LOCATION			RI		R2			R7							R3							8.4	R6	R8:1
Heat		SAMPLE	Chainage		Elevation		terberg Li	mits	Field		Н	R7h	76	H	Н	Н	Н	Н	2.37998	1.19126		Н			Ц	ard Proetur	S.G.	LAEP
This can be not		No.	Ē		(w)	됩	17	ы	m/c	=		_		\vdash	-	0.75	Н	Н	0.0937	_	\rightarrow	\rightarrow			_			
			:			PG	88	£€	26							3/4	3/8	44	84	#16	#30							s/mɔ
		R/ME/ZS-1	21+84	15th N of C/L	0.916	15.4	28.1	12.7	4.4	⊩	┡	╟	╬┈	╟	╟	┢	┢	85.3	81.4	78.2	┢	╟┈	╟┈┆	H	Ļ	10.5	2.74	2.4E-09
D1-50 D1-50 D1	1	R/ME/25-2	19+30	10m N of C/L	915.0	15.8	27.8	12.0	16.0	0.03	L	\vdash	-	\vdash	H	\vdash	\vdash	89.0	9.98	83.6	Н		┝╌╢	Н	냄		2.74	
2.549 Diam S LOS Securide 1913 11.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2 12.2		R/ME/ZS-3	20+50	20m S U/S Shoulden		15.0	27.0	12.0	12.4	-0.22	Ξ	Н	-	-		H	Н	89.4	86,9	84.0	-	Н		-			2.74	7.9E-10
13.00.00 Most Mos	- 1	R/ME/ZS-4	20+50	20m S U/S Shoulder		13.9	26.	12.2	15.1	-	\dashv	\dashv	믁	\dashv	\dashv	-	\dashv	84.2	79.0	75.5		\dashv	\dashv	4	+	4	2.75	9.8E-10
One of the billing o	-1	R/ME/ZS-5	21+90	20m S U/S Shoulder	_	12.1	27.3	12.2	13.5	4	4	-			-	-	\dashv	89.6	84.5	80.7		⊣	-	┥	-	_	2.72	1.4E-10
13-14-05 10.00 M. S. W. S. Bandler 1912 15.2 13.2 13.2 13.2 13.2 13.2 13.2 13.2 13	-1	R/ME/ZS-6	20+00	20m S U/S Shoulder	_!	14.2	24.7	10.5	12.4	_	\dashv	\dashv		-	-{	-	\dashv	85.5	61.7	78.0	-	\dashv	┪	\dashv	+		2.73	4.7E-08
20.54 May No. Shanker 91.04 51.2 51.4 6	-1	R/ME/ZS-7	21+00	10m N at C/L	916.5	<u>2</u>	24.6	5.0		ᆛ	\dashv	\dashv	ᆛ		\dashv	-	-	78.8	75.6	72.5		\dashv	┪	+	+	-	2.72	4.86-10
20-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50 10-4-50	-1	R/ME/ZS-8	2 7 7	5m S U/S Shoulder	0.710	7	26.8	12.1	<u>=</u>		-{	-	╡		-1	╌	┪	82.3	79.4	76.7	┥	⊣	┪	\dashv	┽	-	2.75	
20+400 Disciplinarie Probability Probability Probability Probability Probability Probability Probability Probability Probability Probability Probability Probability Probability Probability Probability Probability Probability Probability Probability Probability	1	R/ME/ZS-9	20+20	S of U/S Shoulder	917.8	15.2	23.4	8.2	8.0	ᅷ	┈╢	4	╡	-	┪	┪	-	808	2.77	73.9	-	┥	+	\dashv	ᆉ	-		5.3E.09
11-50 Diale State Standard 20.0 15. 25. 10.1 20.0 20.1 20.1 20.1 20.0 20.1 20.1 20.0 20.1 20.0 20.1 20.1 20.1 20.1 20.1 20.0 20.1	-	VME/ZS-10	╡	10m N L/D	919.0	14.3	22.6	8.3	11.2		-	-	╡		-	-	-	76.7	72.1	9.89	┥	1	-	-	┪	4		
19+90 (ass NG Shanide 2010) 15, 14, 14, 15, 14, 14, 15, 14, 14, 15, 14, 14, 14, 14, 14, 14, 14, 14, 14, 14	-	V/ME/ZS-11	20+80	13m S U/S Shoulder		12.1	25.2	10.1	12.0	-	-	-	=		-			84.5	81.4	78.4		┪	-	-	┽			6.3E-10
	-	ME/ZS-12	19+00	Ditch crossing	917.6	14.6	24.1	9.5	12.3	4		Η	_	-	-	-	-	82.4	78.1	74.5	-	\dashv	-	-	ᆛ	4		
Hearth H	4	VME/ZS-13	21+50	6m S U/S Shoulder	920.9	14.5	24.8	10.3	11.4			4	╡	\dashv	-	╣	ᅱ	80.9	77.9	74.4	\dashv	-	-	\dashv		-		2.95-09
National N		VMB/2S-14		18m S U/S Shoutder	917.6	14.9	24.9	10.0	12.4	ᆛ	\dashv	\dashv	╡	\dashv		┪	-	83.4	79.5	76.2	┥	-	+	┥	ᅷ	4		1.215-09
Hy-15 George US Shaulder Paris		UME/ZS-15	[Sm N of C/L	918.5	14.7	24.9	10.2	12.2	ᆛ		+	┪	\dashv	-	\dashv	-	84.5	80.1	76.6	\dashv	\dashv	-	+	ᅷ	4		
14-15 6m S UMS Shaulder 9226 15-7 21-5	-41	(/ME/ZS-16		5m S U/S Shoulder	921.0	15.3	24.3	9.0	10.3		-	\dashv	+	\dashv	-	┥	-	84.3	80.7	7.6	\dashv	\dashv	-{	\dashv	-	\downarrow		
11+10 6 mS USS Shaulder 92.53 15.9 15.0 1.0 1.0 0.0 0.0 11.0 0.0 0.0 15.0 0		IME/ZS-17	18+25	6m S U/S Shoulder	922.0	15.7	23.5	7.8	10.1	+	-	\dashv	╡			┪	-	82.7	79.4	76.3	\dashv	┪	┪	+	믁	4		
21+20 20m S US Shaulder 215.2 14.2 214.4 9.0 11.9 -0.39 2082 11.3 9.1 100.0 87.9 82.3 80.4 77.2 77.2 77.5		VME/ZS-18		6m S U/S Shoulder	923.5	15.9	21.0	3.1	8.2			\dashv	╡		-	-	-	78.7	74.9	71.5	-1	╣	+	\dashv	-+	_		9.3E-10
21+50 [OIII-S UIS Simulde] 202.5 [14] 21-50 [OIII-S UIS Simulde] 202.5 11.0 6.7 10.0 -0.64 20.0 10.0 <td></td> <td>/ME/25-19</td> <td>21+20</td> <td>20m S U/S Shoutder</td> <td>923.0</td> <td>15.4</td> <td>24.4</td> <td>9.6</td> <td>6:11</td> <td>-</td> <td></td> <td>-</td> <td>+</td> <td>-</td> <td></td> <td>-</td> <td>\dashv</td> <td>72.8</td> <td>70.4</td> <td>67.4</td> <td>-</td> <td>\dashv</td> <td>-+</td> <td>\dashv</td> <td>┪</td> <td>4</td> <td></td> <td></td>		/ME/25-19	21+20	20m S U/S Shoutder	923.0	15.4	24.4	9.6	6:11	-		-	+	-		-	\dashv	72.8	70.4	67.4	-	\dashv	-+	\dashv	┪	4		
13+40 Sin Nolls Shaulder 9270 145 22.0 27.1 10.1 -0.066 21.14 8.1 102.9 100.0 95.3 97.7 91.3 81.8 71.4 71.7 71.5 71.4 71.7 71.5	-	ME/ZS-20	23+50	10m S U/S Struulder	925.5	14.3	21.0	6.7	10.0	+	-	-	-	-+	+		+	80.9	75.3	71.9	-+	+	+	+	$^{+}$	4		:
18+155 Col. 13.5 2.5.5 1.5.5 2.5.5 1.5.5 2.5.5		VME/ZS-21	23+00	5m N D/S Shoulder	927.0	14.9	22.0	7.1	<u>.</u>	+	-	+		+	+	+	+	87.8	81.8	78.	+	+	+	+	÷	4		
1940 21.0 21.0 22.0	ا رجد	L/ME/ZS-22	18+25	C/L	926.0	2.5	23.0	9.5	20.	+	+	+	+	+	+		+	89.0	9. F	47.7		+	╁	╅	+	+	1	
19+50 7m N of CD 9250 14.2 24.5 10.1 10.2 90.5 100.0 100.0 93.7 90.1 83.8 76.1 70.0 65.2 60.7 75.8 66.2 57.8 46.5 57.6 88.8 2150.0 88.8 20+60 4m N US Shaulder 92.5 1.7 1.6 2.7 1.6 0.3 10.0 10.0 96.0 97.0 98.6 97.0 14.3 69.4 66.2 54.6 45.4 4.0 17.0 17.2 66.7 17.0 98.7 98.7 98.7 98.7 18.0 17.2 17.0 17.0 17.0 98.0 18.0 98.0 18.0 17.0 17.0 98.0 18.0 98.0 18.0 98.0 18.0 98.0 18.0 98.0 18.0 98.0 18.0 98.0 18.0 98.0 18.0 98.0 18.0 98.0 18.0 98.0 18.0 98.0 18.0 18.0 18.0 18.0 </td <td>~1 @</td> <td>ME/75.24</td> <td>8</td> <td>Sm N of C/D</td> <td>977.5</td> <td>2 =</td> <td>22.8</td> <td><u> </u></td> <td>9 %</td> <td>┿</td> <td>╀</td> <td>+</td> <td>+</td> <td>┿</td> <td>+</td> <td>+-</td> <td>+</td> <td>11.7</td> <td>100</td> <td>S</td> <td>┿</td> <td>┿</td> <td>╬</td> <td>╁╌</td> <td>┿</td> <td>╀</td> <td></td> <td></td>	~1 @	ME/75.24	8	Sm N of C/D	977.5	2 =	22.8	<u> </u>	9 %	┿	╀	+	+	┿	+	+-	+	11.7	100	S	┿	┿	╬	╁╌	┿	╀		
25+85 Jin N Dis Shuulder 92.5 17.0 92.6 </td <td>ı≃</td> <td>/ME/25-25</td> <td>19+50</td> <td>7m N of C/D</td> <td>929.0</td> <td>14.2</td> <td>24.5</td> <td>5.0</td> <td>6.4</td> <td> </td> <td>┞</td> <td>┞</td> <td>┢</td> <td>┞</td> <td>\vdash</td> <td>┝╾</td> <td>⊢</td> <td>76.1</td> <td>70.0</td> <td>65.2</td> <td> -</td> <td>-</td> <td>⊢</td> <td>-</td> <td>L!</td> <td>Н</td> <td></td> <td>2,8E-10</td>	ı≃	/ME/25-25	19+50	7m N of C/D	929.0	14.2	24.5	5.0	6.4	 	┞	┞	┢	┞	\vdash	┝╾	⊢	76.1	70.0	65.2	 -	-	⊢	-	L!	Н		2,8E-10
25+86 Jian N Dis Simulute 93.0 19.6 27.0 10.6<	1 =	JME/ZS-26	20+05	4m S U/S Shoulder	927.5		17.0			-	H	10.7	Ξ	H	Н		92.6	86.7	7.67	74.3	Н	H	Н	H				
21+00 6man of Crib 96.5 16.3 26.5 10.5 10.5 96.7 90.0 82.4 76.1 71.3 69.9 66.2 61.3 72.5 11.0 12.0 96.7 10.0 97.5 91.5 91.5 66.9 66.3 66.9 66.9 66.9 67.1 66.9 67.2 120.0 91.0 91.2 96.0 91.0 92.5 11.0 90.0 88.4 88.5 87.5 71.8 67.9 66.9 48.1 47.0 90.0 88.5	, ≠	/ME/2S-27	25+85	3m N D/S Shoulder	933.0	19.6	27.0	7.4	9.81		Щ		_	Н	Н	Η		8.08	89.2	87.5	\dashv	-	-	-	ᅱ	4	•	3.4E-08
21+18 DVS Shwilder 91.0 15.2 21.7 6.5 10.6 0.1 96.0 92.6 88.6 88.5 88.7 11.8 67.9 68.9 68.9 11.8 67.9 68.9 68.9 11.8 67.9 68.1 88.7 88.2 11.8 67.9 68.1 88.7 88.2 11.9 11.0 96.0 96.0 97.0 94.9 97.0 88.7 88.5 66.9 68.1 87.0 88.2 88.7 88.5 66.0 88.1 37.0 88.2 37.0 88.2 77.0 87.0 88.2 37.0 87.0 88.2 77.0 87.0 88.2 77.0 87.0 88.2 77.0 87.0 88.2 77.0 87.0 88.2 77.0 87.0 88.2 77.0 87.0 87.0 97.0 97.0 97.0 97.0 97.0 97.0 97.0 97.0 97.0 97.0 97.0 97.0 97.0 97.0 97.0 97.0	į iž	/ME/2S-28	22+00	6m N of C/D	926.5	16.3	26.2	9.6	12.8						-	-	_	1.97	73.2	6.69	\dashv	Н	-	\vdash	+			
24+50 Smy US Shwalder 911-0 I 13.7 24.6 I 10.9 I I 13 - 0.22 2040 I 12.1 95.3 I 100.0 94.9 94.9 1 10.0 NG 94.9	+4	/ME/ZS-29	⊢	D/S Shoutder	932.0	15.2	21.7	6.5	9'01	_	Ļ	H	H	⊢	⊢	۳	-	75.5	71.8	67.9				_	=			
24+50 8m.5 US Shwalder 92.0 10 NP 10.0 NP 10.0 9.8 10.3 9.6 9.7	1 -	VME/ZS-30	⊢	5m S U/S Shoulder	9,11.0	13.7	24.6	6.01	1.3	 	-	H	⊨	-	-		Н	76.6	72.1	67.9		\vdash		Н		-		
24+50 4m UNS a/CL 9320 1.69 2.05 1.69	1 44	:/ME/ZS-31	⊢	8m S U/S Shuatder	928.0	ĝ	18.3	ž	11.0	-	Н	Н	<u> </u>	┝	H	-	Н	0.00	88.5	87.3	-	\dashv	\dashv	\dashv	-			
25+75 US sh. 934.0 I.5.5 21.7 6.2 9.9 4.09 2100 9.7 100.0 9.7 100.0 9.7	1 15	ME/ZS-32	H	4 m U/S of CL	932.0	6.91	20.5	3.6	8.6		H	H		Н	Н	Н	Н	77.6	75.2	72.5		\exists	-	-	닉	4		
14.5 22.9 9.0 11.7 0.43 2055.8 11.3 99.0 10.0 96.5 94.2 91.8 86.9 82.4 75.6 71.5 69.1 59.5 10.5 10.5 2075.8 10.1 10.0 11.3 98.6 100.0 97.5 94.5 91.7 86.2 82.7 79.4 75.5 72.1 68.7 88.8 85.2 85.7 88.4 75.9 87.5 87.5 88.7 88.4 75.9 88.5 88.7 88.4 75.9 88.5 88.7 88.4 75.9 88.7 88.4 75.9 88.7 88.4 75.9 88.7 88.4 75.9 88.7 88.4 75.9 88.7 88.4 75.9 88.7 88.4 75.9 88.7 88.4 75.9 75.0 75.	1 15	:/ME/ZS-33	╄	U/S sh.	934.0	15.5	21.7	6.2	6.6	느	H	H	H	Н	Н	Н	Н	82.7	79.5	76.2	Н	Н	Н	\dashv	-{			
44.9 24.4 9.7 11.6 0.30 2070.0 11.3 98.6 100.0 97.3 91.7 86.2 82.7 79.4 75.5 72.1 68.7 78.8 51.2 70.0 70.3 90.0 11.2 98.6 100.0 99.3 99.7 99.3 90.3	II		MEAN			14.5	23.9	0.6	11.7		ш	Н	Η	Н	Н	Н	Н	82.4	78.4	75.0	H	Н	Н	Н	뭐	Ц	2.74	7.5E-09
19.6 28.1 12.7 16.0 0.10 2150.0 14.3 103.5 109.0 109.0 99.3 97.7 99.3 90.8 89.2 87.5 87.5 87.5 87.7 87.5 87.7 87.8 87	; 1		MEDIA	z		14.9	24.4	9.7	11.6	╗	_	1	+	-	-	-	ᅥ	82.7	79.4	75.5	+	-	+	+	┿	4	2.74	9.8E-10
0.0 17.0 0.0 8.2 -1.07 1870.0 7.8 95.3 100.0 81.9 82.3 80.4 77.2 72.8 70.0 65.2 60.7 57.8 46.5 37.4 4.0 1910.0 8.7	- 1		MAXIMUN	M (•)		19.6	28.1	12.7	0.9	=		-	=	-+	-	ᆛ		8.	89.2	87.5	+	╛	+	┪	÷	4	2.75	4.7E-08
	1		MINIMUN	4(*)		0.0	17.0	0.0	8.2	=	_	-	_		-	-	ᅱ	72.8	70.0	65.2	60.7	┥	\dashv	-	╣	-1	2.72	1.4E-10

COMMENTS: These are 100% limits.

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Atterberg Limits (ASTM D4318)
Moisture Content (ASTM D2216)
Particle Size Distribution (ASTM D422)
Laboratory Compaction (ASTM D1557)

Specific Gravity (ASTM D854)
Field Density by Nuclear Methods (ASTM D2922)
Moisture Content by Nuclear Methods (ASTM D3017)
Lab Air Entry Permeameter (LAEP)

R6 R7a R7b R8a

Knight Piésold Ltd. consulting engineers

TABLE 6.7

MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

CONTROL TEST SUMMARY SHEET FOR ALTERNATE BORROW AREA

J:\JOB\REPORT\1627\[2-tb\6-7.x\s\Data Sheet

LUODINELL	J. DOBINE FOR LINE A 14-1010-1, XISTORIA SHEET	Steel								-0.0		1000									200		,	
	Knight Piësold Ltd. consulting engineeus	d <i>Ltd.</i> verhs								3	ALTE	RNATE B	CONTROL 1531 - SCHWANT SHEET	IREA	199						PERIOD		13-Sep-96 to 18-Oct-96	18-Oct-96
PROJECT		MOUNT POLLEY - TAILINGS STORAGE FACILITY - STAGE IS CONSTRU	Y TAILIN	GS STOR	AGE FACI	ILITY - ST	TAGE Ib (ONSTRU	CTION												PROJECT NO.		1627	
MATERIAL		GLACIAL TILL FROM ALTERNATE BORROW AREA (NEAR TOPSOIL ST	FROM ALT	ERNATE	BORROW	' AREA ()	VEAR TO	SOIL ST	OCKPILE)												AREA:	Alternate Borrow Area	пом Алеа	
					ວ		ខ							ຮ								ಶ	ర	85
DATE	CONTROL	ā	Depth		Atterberg Limits	sits	Field		63.5	38.1	25.4	19.05	9.525 4.750	50 2.380	80 1.191	91 0.594	4 0.419	9 0.150		0.002	Standard	Standard Proctor (1)	S.G.	LAEP
SAMPLED	SAMPLE	SAMPLE	E	F.	7	Ы	m/c	3	2.5	1.5	-	0.75 0.	0.375 0.187	87 0.0937	337 0.0469	169 0.0234	34 0.0165	5 0.0059	9 0.0029	WKID'D	Max Dry	o ji		
	No.	No.	,	88	₽₽	36	ж	86	3	1 1/2	_	3/4	3/8 #4		#18	#16 #30	#40	001 <i>#</i>	#200	Clay	Density kn/m³	л/с %		сш/s
13. Con. QK	C/ME/AR//(7S 7B)-1A	FT06/09/11.1	(7.4-2.7)	14.7	20.9	6.2	6.8	9,0	91.2	82.8	72.5	67.5 59	59.1 53.9	9 50.5	3 47.6	6 44.5	42.5	34.1	27.3	2.5	e e		2.74	
	C/ME(AB)/(ZS,ZB)-1B		(3.6-3.9)	<u>'</u>			12.1		· _	,	! <u> </u>	: :	71.9 67.3		8 62.1	_	1	T	41.9	8.0			2.73	
13-Sep-96	C/ME(AB)/(ZS,ZB)-2A	ET96/09/13-4	(0.3-0.6)				12.8		0.001	94.5	93.6	92.4	88.1 84.3	3 79.1	15.1	1 70.8	67.9	56.7	- 8	0.6	2025.0	10.2	2.7	
	C/ME(AB)/(ZS,ZB)-2B	•	(2.0-3.0)	14.8	25.4	10.6	11.2	0.34	100.0	94.3	8 2.06	89.2	84,1 78.5	.5 73.1	1 69.3	3 66.1	2	56.4	49.5	13.0	2083.0	10.5	2.73	
	C/ME(AB)/(ZS,ZB)-2C	,	(2.5-3.1)				13.0																	
	C/ME(AB)/(ZS,ZB)-2D	•	(3.6-4.1)				12.1								<u> </u>	<u> </u>	1	i	-					4.43
	C/ME(AB)/(ZS,ZB)-2E	•	(5.0-5.3)				10.9						1	1	<u> </u> 				-					
13-Sep-96	C/ME(AB)/(ZS,ZB)-3A	ET96/09/13-5	(1.0-2.0)				11.9		100.0	8.96	94.5	91.3	84.0 78.3	.3 72.9	8.89	64.7	62.0	51.6	43.3	7.5			2.74	
16-Sep-96	C/ME(AB)/(ZS,ZB)-4A	ET96/09/16-1	(0.0-0.7)	ļ			17.0				1			-	1	<u> </u>		_						
	C/ME(AB)/(ZS,ZB)-4B	•	(1,0-2.0)				10.3									-	<u> </u>		1					
	C/ME(AB)/(ZS,ZB)-4C	•	(2.0-3.0				10.5		100.0	7.86	95.5	93.9	89.6 85.2	2 80.9	9.77.0	73.1	70.5	29.8	20.1	0.0	2070,0	10.5	2.73	1
	C/ME(AB)/(ZS,ZB)-4D	•	(4.0-5.0)				10.1	1		1													1	
16-Sep-96	C/ME(AB)/(ZS,ZB)-5A	ET96/09/16-2	(2.0-3.0)	14.8	24.3	9.5	11.0	-0.40	100.0	97.2	92.2	8 9.68	83.8 77.3	.3 72.0	8.79	8.	61.7	53.3	46.4	7.5	2083.0	9.5	2.73	3
	C/ME(AB)/(ZS,ZB)-5B		(3.3-3.6)				10.4							1	1	+	+	+	+					
96-daS-91	C/ME(AB)/(ZS,ZB)-6A	ET96/09/16-3	(0.5-1.5)				11.8		100.0	0.66	98.0	96.1	92.1 88.5	83.	9 80.0	76.4	*	9.6	26.4	0.9	2000.0	4:-	2.74	
	C/ME(AB)/(ZS,ZB)-6B		(2.2-2.7)				11.4							+	1	-	<u> </u>	1						
	C/ME(AB)/(ZS,ZB)-6C	•	(4.4-4.7)				10.2				$^{+}$	-	+	+			+	-	Ė				1	
16-Sep-96	C/ME(AB)/(ZS,ZB)-7A	ET96/09/16-4	(3.5-4.5)				9.3		0.00	95.4	94.1	91.6	85.9 81.1	.1 77.3	.3	-1 68.5	65.5	-52.7	42.3	7.5	2168.0	200	2.72	
	C/ME(AB)/(ZS,ZB)-7B	•	(4.5-5.0)				12.6				\dagger		-	_	+	1	<u> </u>	+	1					İ
19-Sep-96	C/ME(AB)/(ZS,ZB)-8A	ET96/09/19-1	(0.3-1.0)				11.9			+	$\frac{+}{\parallel}$	1	<u> </u>	$\frac{1}{1}$	<u> </u>	1	<u> </u>	1	<u> </u>	1				
	C/ME(AB)/(ZS,ZB)-8B	*	(1.0-2.0)	<u> </u>			6.6		1	Ì		1		<u> </u>	1	<u> </u>			1					
	C/ME(AB)/(ZS,ZB)-8C	•	(2.5-3.0)	1			10.7							1	1		1	1		-				
19-Sep-96		ET96/09/19-2	(1.0-2.0)				11.6		İ		-		1	1	<u> </u>	<u> </u> 	<u> </u>	<u> </u>		_				
	C/ME(AB)/(ZS,ZB)-9B	•	(3.0-4.0)				12.5		Ì			+		1	1	1		1	+	_				·
	C/ME(AB)/(ZS,ZB)-9C		(4.5-4.9)				12.1				1	+				1		<u> </u>	-					
	C/ME(AB)/(ZS,ZB)-9D	•	(6.0-6.3)	-			10.5							1		<u> </u>	1	<u> </u>	-	1				
19-Sep-96	C/ME(AB)/(ZS,ZB)-10A	LET96/09/19-3	(0.2-0.5)				8.41							1	$\frac{\perp}{\uparrow}$	1	<u> </u> 	1		_				
	_		(0.5-1.1)				13.4							-	-	-	-	_					-	
					:																			

CONSULTING ENGINEERS

TABLE 6.7

MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

CONTROL TEST SUMMARY SHEET FOR ALTERNATE BORROW AREA

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J:VOB\REP	J:UOB\REPORT\1627\ 2-ib 6-7.xls Data Sheet	a Sheet																						19/04/97
	Knight Plesold Ltd.	ld Ltd.								iO3	VTROL	CONTROL TEST - SUMMARY SHEET	SUMM	ARY SH	EET						SHEET		of I	
	CONFULTING ENGINEERS	INERIO									· ALTE	ALTERNATE BORROW AREA	ORROW	AREA							PERIOD:		13-Sep-96 to 18-Oct-96	18-Oct-96
PROJECT		MOUNT POLLEY - TAILINGS STORAGE FACILITY - STAGE IB CONST	Y - TAILIN	SS STOR	AGE FAC	ILITY - S	rAGE 16 C	ONSTRU	RUCTION												PROJECT NO.:		1627	
MATERIAL	T: 1	GLACIAL TILL FROM ALTERNATE BORROW AREA (NEAR TOPSOIL	FROM ALT	ERNATE	BORROV	/ AREA ()	VEAR TOP		STOCKPILE)												AREA:	Alternate Borrow Area	row Area	
					ວ		ខ							c3							S	C4	C6	C8a
DATE	CONTROL	뮵	Depth	γıγ	Atterberg Limits	aits	Field		63.5	38.1	25.4	6 50.61	9.525 4.	4.750 2.380	1.191	91 0.594	4 0.419	0.150	0.074	0.002	Standard Proctor (1)	Proctor (1)	S.G.	LAEP
	C/ME(AB)/(ZS,ZB)-10C		(2.0-2.4)	15.4	23.6	8.2	13.2	-0.27	0.001	95.9	94.6	93.2	88.8 84.1	.1 78.7	7 74.1	1 69.7	6.99	57.5	49.7	6,3	2025.0	11.3	2.75	
-777	C/ME(AB)/(ZS,ZB)-10D	•	(3.4-3.6)				6.6								<u> </u>					-				
19-Sep-96	C/ME(AB)/(ZS,ZB)-11A	ET96/09/19-4	(0.1-0.4)				12.2						_ <u> </u>		_									
	C/ME(AB)/(ZS,ZB)-11B		(0.4-0.85)				14.5	-																
	C/ME(AB)/(ZS,ZB)-11C	•	(0.85-2.0)				10.8																	
	C/ME(AB)/(ZS,ZB)-11D	•	(2.0-3.0)				0.01																	
	C/ME(AB)/(ZS,ZB)-11E	•	(3.5-3.7)				9.3				.			<u> </u>					1					
:	C/ME(AB)/(ZS,ZB)-11F	•	(4.0-4.2)				11.2				!						-			į				
	C/ME(AB)/(ZS,ZB)-11G	•	(5.0-5.4)				9.5							-	<u> </u> 			-						
13-Oct-96	C/ME(AB)/(ZS,ZB)-12	Hand Dug	(0.0-0.4)	17.2	25.8	8.6	16.9	-0.03	100.0	97.9	89.4	86.6	79.5 75.1	.1 70.6	6 66.2	2 61.9	59.2	50.4	43,8	10.0	1970.0	12,4		
13-Oct-96	C/ME(AB)/(ZS,ZB)-13	Hand Dug	(0.0-0.3)	16.4	21.6	5.2	13.9	-0.48	_	96.5	93.6	8 6.06	84.7 79	79.4 75.8	_	4 67.1	64.6	56.0	48.0	7.0	1975.0	12.0		
			MEAN	15.6	23.6	8.1	11.7	-0.41	99.3	94.6	-	88.4 8	82.6 77		3 69.4	4 65.5	63.0	53.5	45.6	7.7	2044.3	10.7	2.73	
			MEDIAN	15.1	24.0	8.4	11.4	-0.37	100.0	96.2	93.6	91.1	84.4 79	79.0 74.5	\dashv	\dashv	-	54.7	47.2	7.5	2025.0	10,5	2,73	
		MA	MAXIMUM (*)	17,2	25.8	10,6	17.0	-0.03	100.0	0.66	98.0	96.1	92.1 88	88.5 83.9	9 80.0	76.4	74.1	64.6	56.4	13.0	2168.0	12.4	2.75	
		MIN	MINIMUM (*)	14.7	20.9	5.2	6.8	-0.94	91.2	82.8	72.5	67.5 5	59.1 53	53.9 50.5	5 47.6	6 44.5	42.5	34.1	27.3	2.5	1970.0	8.3	2.72	
€	These are 100% limits.																						/	
ប	Atterberg Limits (ASTM D4318)	TM D4318)																						ند
ខ	Moisture Content (ASTM D2216)	TM D2216)																						
ខ	Particle Size Distribution (ASTM D422)	ion (ASTM D4	22)																					
2	Laboratory Compaction (ASTM D1557)	m (ASTM D15:	53																					
జ	Specific Gravity (ASTM D854)	M D854)																						
C7a	Field Density by Nuclear Methods (ASTM D2922)	ear Methods (A	STM D29	a a																				
දි දූ	Moisture Content by Nuclear Methods (ASTM D3017) Lab Air Entry Permeameter (LAEP)	vuclear Method imeter (LAEP)	s (ASTM I)3017)																				





MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

CONTROL TEST SUMMARY SHEET FOR ORIGINAL BORROW AREA

JAJOBAREPORTA 1627/2-TBL6-6.XLS

	Knieht	Piesold Ltd.	-							S	CONTROL TEST - SUMMARY SHEET	EST - S	UMMAR	Y SHEE	L						SHEET		1 of 2	
	CHINNEA)	CONSULTING ENGINEERS						•			ORIGI	NAL BOR	ORIGINAL BORROW AREA	V.							PERIOD	ۃ	23/6/ to 5/10/96	96/0
PROJECT:		MOUNT POLLEY - TAILINGS STORAGE FACILITY - STAGE IS CONSTRUCTION	INGS STOR	GE FA	CILITY - 1	STAGE 1b	CONSTRU	CTION													PROJECT	CT NO.:	1627	
MATERIAL		GLACIAL TILL • FROM ORIGINAL BORROW AREA (OB)	ORIGINAL	BORRO	WAREA	(go)															AREA:	Original	Borrow Area	
					ວ		z						i				ıı			!!			ප	83 2
DATE	SAMPLE	Lecation	Depth	Ĭ	Alterberg Limits	imits	Field		76.2	38.1	25.4	╁┼	-			-	\rightarrow		-	_	÷		S.G.	LVEP
SAMPLED	No.		Œ	7	=	교	m/c	3	~	-	-	┥	-		-	_	╗	_	-	_	$\overline{}$			
				P6	P6	88	88	<i>P</i> 5	m	<i>2</i> 11 I		3/4	3/8	#4	***	#16	#30 ##	#40 #10	1100	#200 Clay	y Density	7 m²c	<u> </u>	cm/s
	C/ME/ZS-29	OB : FT-96/08/15-10	0.5-1.2	Į,			14.4	ŀ					-	-	<u> </u>			L				L		ŀ
	C/ME/ZS-30A	OB : ET-96/08/15-11	10-3.0	16.2	22.2	6.0	10.2	91.0	100.0	97.5	96.3	-	87.9	82.7	78.6 70	70.4 65.9	.9 62.7	7 49.6	40.0	0.8	2090.0	9.5	2.73	
	C/ME/ZS-30B		4.0-6.0			•	10.9	•			<u> </u>	<u> </u> 	 	+-	ļ		1	<u> </u>	1	<u> </u>	1	<u> </u>		
	C/ME/2S-31A	OB : ET-96/08/15-12	0.5-2.0	13.6	20.1	6.5	12.7	-0.14	100.0	100.0	99.1	97.5	95.4	91.6	86.4 81	81.1	74.6 69.6	51.5	5 40.6	9.0	2015.0	0.11	2.75	•
•	C/ME/ZS-31B		4,0-6.0		,		10.9					<u> </u>	<u> </u>	+	<u> </u>	<u> </u>	╁╾	+-	1		<u> </u>	<u> </u>		•
16-Aug-96	C/ME/ZS-32A	OB : ET-96/08/16-1	9.0-1.0	٠			14.7									•		<u> </u>	<u>'</u> .	<u>.</u>		٠		•
	CIME/ZS-32B	•	1.5-2.5	13.3	24.4	1.11	13.1	-0.02	100.0	100.0	100.0	100.0	91.4	86.9	81.8	77.6	73.6 71.0	0.60.9	.9 52.9	9.11.0	2005.0	10.9	1.71	'
	C/ME/ZS-32C	•	4.0-5.8				8.6														•	•		
	C/ME/ZS-33A	OB : ET-96/08/16-2	0.6-1.5	'			14.0		•		•	,						<u>'</u>		<u>'</u>	•	•		•
	C/ME/ZS-33B	•	1.5-2.0		٠		9,1	٠	•		•	•	۱,	<u> </u> -	 						•	-		
•	C/ME/ZS-33D	•	4.5-6.0	•	١.	Ŀ	13.8	•													•	٠	•	•
23-Scp-96	C/ME/ZS-34A	ET96/09/23-1	(0-0.25)	<u> </u>			12.0		,											•	•	•		
	C/ME/ZS-34B	,	(0,25-0,55)	Ŀ	•		15.2				•	•				•					•		•	•
	C/ME/ZS-34C	•	(0.55-2.0)	•		,	9.01		٠	,	•			•						•	_	•	•	•
•	C/ME/ZS-34E	•	(4.8-5.1)			•	12.7	•										'	_	•	•	٠	•	
	C/ME/2S-34F	•	(5.5-5.7)	•	٠	,,	13.0	•	•						•						•	٠	•	٠
•	C/ME/ZS-35A	ET96/09/23-2	(0-0.0)	•	•	-	11.9	٠	•	•	•										•	٠		
	C/ME/ZS-35B	•	(0.6-2.0)	٠	•	,	9.3	·	•	•		,					_			<u>' </u> .]	-	•		'
	C/ME/ZS-35C	•	(3.0)		'	•	8.6		100.0	93.3	92.3	91.3	85.9	79.8	74.7	70.7	67.0 64.4	8.48	.8 46.4	4. 7.5			2.72	1
	C/ME/ZS-35D	•	(4.5)	٠	'	,	10.4	•	-			•	•		<u> </u> .					· .	-	•		
24-Sep-96	C/ME/ZS-36A	ET96/09/24-1	(0.0.25)	٠	٠		10.6		•				-	-		•						•	1	'
	C/ME/ZS-36B		(0.25-0.6)	1	•		14.9	,	•	•	•		1		<u> </u> -	<u> </u> -	1		<u>]</u>]	<u> </u> -	1		-	•
-	C/ME/ZS-36C	•	(0.6-2.0)	١	•		9,01		•		-	<u> </u>	+		<u></u>	-1	-+	+	-		•	·		•
	C/ME/ZS-36D	•	(3.1)	•	<u> </u>		13.0		000	0.00	93.6	8.3	0.20	76.0	71.1	67.6	64.6	62.6 54.0	96.5	2. 0.6			2.75	
•	C/ME/ZS-36E	•	(4.5)		•	•	10.4		'	•	•	•	-	+		1	<u> </u> -	1	_	1	<u> </u>		•	
•	C/ME/ZS-36F	•	(6.0)	1	•		6.0	1	•			-	-	-	<u> </u> -	1	<u> </u> -	1		<u> </u>	<u> </u>	·	-	
•	C/ME/ZS-36G	•	(6.3)		•	•	9.5		-	•		•	<u>.</u>	-		1	<u> </u>	1	1	<u> </u>	1	•	•	
•	C/ME/ZS-37A	ET96/09/24-2	(0-0.4)	•	<u>·</u>	•	12.6	•								-		t	· -	<u>. </u>	-	•	,	
•	C/ME/ZS-37B	•	(0.4-1.0)	•	•		8, 1	•	. 6		. 8		+	$^{+}$, ,				. -	ا'	•	. 2	•
	CIMEIZS-37C	r resource	(1.0-2.5)	1	-		2	_	3.0	*.C	,0,0	q: ,		,0,	+	+	+	+-	+	$^{+}$	 -	ŀ		,
	C/ME/23-36A	E13000/154:3	(1.1-0.)		1		2		8	4 10	7 70	Į-	653	78.5	2000	9 63	919	805	0.02	7.0	 		2.75	
	C/ME/25-38B	•	0.1-2.0	<u>. </u>	•		2 2		P. I			╬	<u> </u>	+	1	+	+	+-	<u> </u>	 	1	.		
	C/ME/25-38C	•	(5.0.5 8)			<u> </u>	9				1	-		-			<u> </u>			<u> </u>	:		ļ.	
-	C/ME/2S-38E		(5.9)	.	-	.	10.4					,		<u> </u>		<u> </u>	<u> </u> -							,
•	C/ME/ZS-39A	ET96/09/24-4	(0.2.5-1.1)			.	12.5									,	<u>'</u>				•	•		!
	C/ME/ZS-39B	•	(1.1-2.5)			,	10.8	•			,	•	•					<u> </u>		1	<u> </u>	:	•	,
•	C/ME/ZS-40A	ET96/09/24-5	(05-1.2)	٠	•		13,5		١		•	•	-						1		•	•		•
•	C/ME/ZS-40B	•	(1:2-2.5)	-		1	10.4	1					+	1			<u> </u>	<u> </u>	1	<u> </u>	$\frac{1}{1}$	'		
•	C/ME/ZS-41A	ET96/09/24-6	(0.0.6)	٠	'	,	12.5	•	,		-		-			1	1	<u>' </u> .]			•	<u>. </u>	•	
•	C/ME/ZS-41B		(0.6-2.0)	٠		-	11.2			-			-	-					$\ $		·			

MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

CONTROL TEST SUMMARY SHEET FOR ORIGINAL BORROW AREA

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CONKELORI	J:UOB\REPORT\1627\2-TBL6-6.XLS	XIX																						5/19/97
	Knight	Knight Piesold Ltd.								CON	ROL TE	ST - SUN	CONTROL TEST - SUMMARY SHEET	HEET						S	SHEET:		of 2	
	CONNILL	CONSULTING ENGINEERS						٠			ORIGIN	ORIGINAL BORROW AREA	W AREA							<u>.</u>	PERIOD:	23	23/6/ to 5/10/96	96/
PROJECT:	<i>-</i>	MOUNT POLLEY - TAILINGS STORAGE FACILITY - STAGE IS CONSTRUCTION	LINGS STOR.	AGE FAC	ILITY - ST	TAGE IS CO	NSTRUCT	NOI												-	PROJECT NO.:		1627	
MATERIAL:		GLACIAL TILL • FROM ORIGINAL BORROW AREA (OB)	M ORIGINAL	BORROW	AREA (C)B)														V	AREA: Original Borrow Area	iginal Borre	w Area	
					CI		7.5	H			1			ខ									93	ర
DATE	SAMPLE	Location	Depth	Att	Atterberg Limits	vits	Pield		76.2	38.1	25.4 19	19.05 9.525	5 4.7498	8 2.37998	1 '	1,19126 0,59436	0.4191	0.14986	0.07366	0.002	Standard Proctor ⁽¹⁾	charth	S.G.	LAEP
SAMPLED	No.		(ii)	P.L.	77	Ы	m/c	 _	3	1.5	1	0.75 0.375	15 0.187	0.0937	0.0469	0.0234	0.0165	6500.0	0.0029	4 strates	Max Dry	Opt.		==
				96	96	88	89	24	3	1 1/2	1	3/4 3/8	8 #4	8#	91#	05#	#40	001#	ff200	Clay	Density	υγс		cm/s
												_						·			kg/m³	28		
о •	C/ME/ZS-41C		(2.5)	·		F	9.4	 -	-	-	<u> </u>					Ŀ			-	-	-			
	C/ME/2S-42A	ET96/09/24-7	(0.5-1.2)	-		•	12.4						1				,	•						
	C/ME/ZS-42B		(1.2-2.5)		•		6.1								٠									
	C/ME/ZS-42C	•	(2.5)			•	,	_					-	_		٠		,			•	•		
5-Oct-96	C/ME/ZS-43	OB		13.9	22.5	9.8	14.4	90.0	100.0	98.4	95.2 92	92.6 88.1	93.6	80.8	77.6	73.6	70.8	58.8	48,5	8.1	2062.0	10.7		
5-Oct-96 (C/ME/ZS-44	0.0		13.9	20.2	6.3	12.4	-0.24		97.0	94.0	91.6 86.7	7 81.4	77.9	73.5	69.2	66.4	55.1	42.7	5.9	2095.0	9.7		1.0E-09
15-Aug-96 C.	C/ME/ZS-26A	OB : ET-96/08/15-7	2.0-3.5			7	15.5			-	100.0		3 96.2		93.7	92.4	91.3	85.7	81.0		-	•		•
16-Aug-96 C.	C/ME/ZS-33C	•	3.4-4.5	•		•		\equiv		-					95.5	93.7	92.3	87.6	83.4	13.0				,
	_	MEAN		16.0	24.6	8.6	12.7	98.0	0.001	99.2	97.8 95	95.4 91.5	0.78 3	83.4	78.5	74.7	71.5	8.65	50.4	10.5	2005.2	11.6	2.73	3.5E-09
	W	MEDIAN		16.5	24.4	8.9	12.5	-0.11	100.0	100.0	98.6 96	96.7 92.5	5 88.6	85.0	81.5	17.1	74.3	62.1	52.9	9.5	2002.5	11.1	2.73	1.8E-09
	MAX	MAXIMUM (*)		18.2	30.1	14.8	19.4	0.31	100.0	10001	100.0	6.76 0.001	9 65.6	93.3	7.06	88.0	86.0	77.2	69.2	18.0	2095.0	13.9	2.75	1.4E-08
	MIN	MINIMUM (*)		13.3	18.5	1.3	1.9	2.79	100.0	93.3	88.8 84	84.0 73.1	1 66.1	60.2	54.6	51.2	47.9	28.3	13.8	3.0	1920.0	9.5	2.70	5.0E-10

Bold and Italicized no suitable for fill material. Not Used for Mean, Median, Maximum, Minlmum calculations.

These are 100% limits.

Atterberg Limits (ASTM D4318)

Moisture Content (ASTM D2216) Particle Size Distribution (ASTM D422) Laboratory Compaction (ASTM D1557)

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Specific Gravity (ASTM D854)

Field Density by Nuclear Methods (ASTM D2922) Moisture Content by Nuclear Methods (ASTM D3017) Lab Air Entry Permeameter (LAEP)

TABLE 6.5

MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

SUMMARY OF LABORATORY TESTS

EFFECTIVE STRENGTH PARAMETERS, COMPACTION AND PERMEABILITY TEST RESULTS

Test Pit Sample Location No. TP95-7 Mill Site TP95-18 Tailings/Reclaim Pipeline Route TP05-7 Parimeter Foundation	tion	BPFECTIVE STRENGTH PARAMETERS	TERNOTH		10000			
	tion	PARAM	THE COLUMN)	COMPACTION	7	PERMEABILITY	
	tion		STERS	Natural	Optimum	Maximum	Permeameter	Soil
		Friction Angle,	Cohesion,	Moisture	Moisture	Dry	Falling	Description
		ø,	- 0	Content	Content	Density	Head Test	
		(degrees)	(kPa)	(%)	(%)	(kg/m³)	(cm/sec)	
		•	ŀ	10.9	8,4	2192	,	Silty SAND, some gravel and clay
	peline Route	·		13.8	10.1	2130	•	GRAVEL and SAND, some silt, trace clay
	Perimeter Embankment Foundation	35	0	11.1	8.0	2200	4×10^{-8}	SAND and SILT, some gravel and clay
TP95-31 East Ridge Borrow Area	Area	ı	1	11.0	7.6	2200	6×10^{-8}	Silty, sandy GRAVEL, trace clay
TP95-35 South Basin		,	•	,	,	•	7×10^{-7}	Sandy SILT, some clay, trace gravel
TP95-37 South Basin		35	0	•	•	,	•	SAND and SILT, some gravel and clay
TP95-38 Main Embankment Foundation	Foundation	•		1	,	,	3×10^{-7}	SILT, some clay, trace sand and gravel
TP95-39 Main Embankment Foundation	Foundation	33	0	,	ı	•	2×10^{-6}	SILT and fine SAND, some clay
TPB-13.14.16 Embankment & Pond Foundations	rd Foundations		ı	25.1	13.3	1935	2 x 10 ⁻⁸	Clayey SILT, some sand, trace gravel

A PART

Notes

1. Triaxial tests results from samples TP95-27 and 37 were combined to determine average strength pearameters for the glacial till material.

2. Compaction tests performed as per ASTM D1557 Modified Proctor tests.

3. Permeability tests carried out on samples compacted with standard proctor energy and at natural moisture content.

4. Sample TPB-13,14,16 was selected for laboratory testwork in 1989 and has been reported for comparison.

MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

CONTROL TEST SUMMARY SHEET FOR ORIGINAL BORROW AREA

JAJOBAREPORTA 627/2-TBL6-6.XLS

	T.tusno)	CONSULTING ENGINEERS									10100	OPICINAL BORDOW ABEA	a v AxO a	7.5							PERIOD	. 65	20/01/2 or /2/10	70/07/
PROJECT :		MOUNT POLLEY - TAILINGS STORAGE FACILITY - STAGE IS CONSTRUCTION	LINGS STOR	AGE FAC	LITY - ST	AGE IS C	ONSTRUC	NOL			5	100									PROJ	PROJECT NO.:	1627	
MATERIAL:		GLACIAL TILL - FROM ORIGINAL BORROW AREA (OB)	M ORIGINAL	BORROV	/ ARBA (C)B)															AREA:		Original Borrow Area	
\vdash					ວ		22								ຍ								C6	CB
	SAMPLE	Lycation	Depth	VII.	Atterberg Limits	siis	Field		76.2	Н	25.4	Н	Н		\vdash	\vdash	┝╼┿	╼┼	-		+	<u>8</u>	S.G.	LAEP
SAMPLED	Ë		Ē	7 <u>.</u>	73 %	E 8	m/c %	1 %	т п	1.5	- -	3/4	3/8	0.187 #4	0.0937 0.0 #8 #	#16 #	#30 #	#40 #	0.0059 0.0 #100 #2	#200 Clay	y Density	ory Opt.		сш/х
23-Jun-96	C/ME/ZS-1	OB : Test Pit	ļ.	14.4	28.3	13.9	16.6	0.16	100.0	0.001	100.0	100.0	97.9	94.4	91.4 87	87.7 69	69.0	54.4 2	28.3 13	13.8 3.7	-	<u> </u>	·	3.9E-09
23-Jun-96	C/ME/ZS-2	OB : Test Pit		16.4	23.0	9.9	15.6	-0.12		1	-	<u> </u>	<u> </u>	91.0	88.4 8:	85.3 82	82.0	79.6	68.7 58	58.9 9.0		٠	2,75	6.7E-09
23-Jun-96	C/ME/ZS-3	OB : Test Pit	,	16.7	22.9	6.2	13.9	-0.45		•	•		<u> </u>	,							•	•		'
	C/ME/ZS-4	OB : Test Pit		16.5	23.6	7.1	13.1	-0.48	100.0	100.0	10001	100.0	93.9	89.3	86.3	82.7 78	78.4 75	75.5 6	62.4 51.7	7. 8.0		•	2.73	'
	C/ME/ZS-5	OB : Test Pit	•	18.2	21.6	3.4	16.4	-0.53	<u> </u>		 	<u> </u>				82.0	78.4	75.9 6.	65.5 56.5	.5 15.0		-	2.74	1.56-09
23-Jun-96	C/ME/ZS-6	OB : Test Pit	•	17.5	18.8	1.3	14.9	-2.00	100.0	0.001	0.001		94.5	91.0	86.7 84	84.0 81	81.3	79.5 6	8.65 7.69	8.		-	2.75	ا.
23-Jun-96	C/ME/ZS-7	OB : Test Pit	•	16.9	30.1	13.2	17.8	0.07		0.001	<u> </u>	100.0	97.8	94.1	90.4	87.0 83	83.6 81	81.4	71.4 62	62.2 17.0	ا ٔ	<u> </u>	2	2.5E-09
17-Jul-96	C/ME/ZS-8	OB : Test Pit		17.1	28.0	10.9	17.6	0.05	100.0	0.001	98.1	6.90	8 6.19	88.6	85.7 81	81.8	76.7	73.4 6	61.9 53.1	13.0	1920.0	13.9	2.74	5.0E-10
17-Jul-96	C/ME/ZS-9	OB : Test Pit	,	16.6	25.7	9.1	19.4	0.31	100.0		98.0	96.7		90.6	85.0 81	81.1		74.3 6	63.6 55	55.2 12.0	0 1935.0	13.1	2.72	5.7E-10
17-Jul-96	C/ME/ZS-10	OB : Test Pit	•	15.8	26.4	10.6	17.0	0.11	100.0	0.001	6.96	95.6	92.5	89.6	87.8	85.0 81	81.2	78.3 6	66.2 57.1	.1 17.0	0 1940.0	.0 13.4	2.72	1.8E-09
2-Aug-96	C/ME/ZS-11	OB : Test Pit		15.8	26.4	10.6	15.7	-0.01	100.0	0.001	0.001	98.8	97.6	94.8	92.8	88.9 82	83.6 86	80.3	68.3 58.1	.1 10.0		1	2.70	1,6E-09
96-8nV-6	C/ME/ZS-12	OB : Test Pit	,	_	•	'	18,4	,	,	•		•		•		٠,				· 	<u>'</u>	•	٠	<u>' </u>
9-8nV-6	C/ME/ZS-13	OB: Test Pit					13.8	,			·			,				,		<u> </u>	1	,	,	•
9-4nV-6	C/ME/ZS-14	OB : Test Pit	١	17.5	29.4	11.9	15.8	-0.14	100.0	0.001	98.6	97.4	95.8	92.7	89.6	86.9	83.6	81.4	72.2 64	64.2 18.0	<u> </u>	-	2.72	<u>'</u>
9-Aux-96	C/ME/ZS-15	OB : Test Pit	•	17.8	27.3	9.5	19.4	0.17	100.0	100.0	0.001	98.6	97.6	95.6	93.3	90.7	88.0 86	86.0	77.2 69	69.2 18.0	0	•	2.73	1
96-8nV-6	C/ME/2S-16	OB : Test Pit	·	<u> </u>	·		12.0	,	•	-	 -	-		1	_	,		-		<u>'</u>	<u> </u>	1		•
9-Aug-96	C/ME/ZS-17	OB: Test Pit			•	•	18.5	•	•		•				<u> </u> ,	,		<u> </u> 			1	•]	•	'
12-Aug-96	C/ME/ZS-18	OB : Test Pit		17.1	18.5	4.1	13.2	-2.79	100.0	97.6	1.96	94.0	89.7	85.2	79.9	75.5	71.2 68	68.7	58.6 50.3	.3 7.0		•	2.71	4.1E-09
12-Aug-96	C/ME/ZS-19	OB : Test Pit		14.4	29.2	14.8	14.0	-0.03	100.0	0.001	6.86	91.7	88.5	86.3	83.1 80	80.1	77.1	75.1 6	66.1 58.2	.2 13.0	0 1980.0	.0 12.5	2.73	1.4E-08
10-Aug-96	C/ME/ZS-20	OB : ET-96/08/10-1		·	•	•	14.0	•	100.0	100.00	100.0	97.9	94.8	91.8	88.7 8	85.7 82	82.8	80.8	71.6	63.2 15.0		•	•	
10-Aug-96	C/ME/ZS-21	OB: ET-96/08/10-2		·	·	•	13.4	•	•	•	•	-		•		<u> </u> -	-					٠	٠	
- 1	C/ME/ZS-22A	OB : ET-96/03/15-1	0,5-1,5		•	•	13.4	•	•	-	<u> </u> 	-		-		-			<u> </u> -		-	•	1	<u>. </u>
	C/ME/ZS-22B	•	2.5-3.0				16.1	•	-		-	-	_	,	<u> </u> ,			_		<u> </u>	1	·	•	<u> </u>
ان ا	C/ME/ZS-23A	OB : ET-96/08/15-2	0.5-1.5	14.9	28.1	13.2	13.6	-0.10	-	-	- !	-	-	98.6	84.7 8	<u>i</u>	+	- 1	+	59.5 15.0	0.0001	.0 12.3	2.75	<u>'</u>
7	C/ME/ZS-23B		2.5-3.5		,		12.9	•	0.001	0.001	0.001	95.6	91.6		+	81.5	78.1	75.5	63.2	53.7		<u> </u>	•	1
1	C/ME/ZS-24	OB: ET-96/08/15-4	0.5-1.5		1	-	30	1	 -	-	<u> </u>	$^{+}$	÷	+	+	+	+	+	+	_	<u> </u>	· 	<u> </u>	-
-1,	C/ME/ZS-25	OB : ET-96/08/15-5	4,0-6.0		•	-	8 : 0	·	000	0.00	5.86	97.0	0.0	82.4	89.1	74.6	77.9	727	39.7	5.50		1	-	'
`° 	CIMEICS-2018	0 2000000000000000000000000000000000000	5.7-7.5				6	<u> </u>	7	1	+	╬	÷	$^{+}$	+	╁	÷	÷	1	<u>;</u>	<u> </u>		,	•
١,	Chetta-410	0-C1/90/05-13-00	3030	. :	•	<u>. </u>	2		5	2	8 60	÷		-	87.8	87.8	78.3	74.8	62 1 53	52.4	2000.0	0 11.2	2.73	
ع ر	C/ME/24-21D		10.40	•			8.7	: .	0.00		1	77.1	•	!	<u>:</u>		-	•					•	.1
10	C/ME/ZS-27D	•	4.0-6.0				8.5		100.0	100.0	1	1 1	1		79.5	71.2 6	62.6 5	58.2 4	! !	1 /		.		•
٥	C/ME/ZS-27E		6.0-7.0	,			10.1	•	1	-		i	- 1	:		÷	_	+	1	- !	'	- !	-	•
٥	C/ME/ZS-28A	OB : ET-96/08/15-9	1.0-3.0	17.0	24.4	7.4	11.8	0.70	0.001	100.0	98.9	97.6	8.46	-	-	83.6	80.6	+	68.6	51.5 12.0	1	0.	2.70	
	200 000 0000	•											-	-		_	_	2				_	_	



CONSULTING ENGINEERS

TABLE 6.3

MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

STAGED EMBANKMENT FILL QUANTITIES

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FOTAL 1,400,500 3,349,400 6,189,800 110,700 403,400 36,900 60,000 828,900 (E) 129,900 213,600 535,200 152,500 23,800 15,400 2008 965 ∞ 1,794,900 ,403,700 173,400 28,900 164,700 14,700 9,500 2006 196 537,800 168,800 160,900 169,800 25,300 13,000 2004 926 FILL QUANTITY BY STAGE (m³ 216,300 155,900 180,700 561,900 9,000 2002 5 951 1,256,100 202,800 844,700 208,600 2000 946 240,500 75,800 40,700 98,000 18,200 1999 7,800 940 412,200 610,800 21,000 60,200 42,700 59,800 14,900 1998 937 652,600 22,000 386,300 242,000 1996/97 2,300 934 Stage El. (m) Year Chimney Drain: clean sand and gravel C: Random Fill, downstream shell oe Drain: clean sand and gravel B: Glacial Till, downstream shell 3. Cyclone Sand, upstream shell CBL: free-draining random fill **ZONE AND MATERIAL** S: Glacial Till, core zone OTALS

Notes:

) All quantities listed above are neat line + 10% contingency.

2) No allowance has been added for cut to fill shrinkage

3) Coarse Bearing Layer material type dependent on tailings beach development; to be determined prior to construction.

TABLE 6.4

MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

SUMMARY OF LABORATORY TESTS INDEX TEST RESULTS

Specific Natural Atterborg Limits Fight Fight<	J:\JOB\REPORT\162;	J:JOBNREPORT/1627/2-TBL6-4,XLS/Table 3,1					٠,						Apr 18 97 6:00 am
Gravity Moisture (%) +#4 #4 + 200 2000-0.002nm -0.002nm -0.002nm - 10.4 21 11 PL PL 1.1 6 27 6 - 10.4 21 13 8 -0.3 21 46 27 6 - 10.9 24 14 11 -0.2 16 39 35 10 - 12.6 25 17 9 -0.4 38 36 19 7 - 13.8 27 13 14 0.0 38 36 19 7 - 14.5 26 13 13 0.1 10 39 25 26 - 17.1 24 12 13 0.4 15 37 33 15 - 11.0 22 14 8 -0.4 41 27 25 14 14 2 22 <td< td=""><td>Test Pit</td><td></td><td>Specific</td><td>Natural</td><td>٧</td><td>tterberg</td><td>Limits</td><td><u> </u></td><td></td><td>Grain Size</td><td>Distribution</td><td></td><td></td></td<>	Test Pit		Specific	Natural	٧	tterberg	Limits	<u> </u>		Grain Size	Distribution		
Content (%) LL PL PI LI % Gravel % Sand % Silt % Clay - 10.4 21 13 8 -0.3 21 46 27 6 2.78 10.9 24 14 11 -0.2 16 39 35 10 - 12.6 25 17 9 -0.4 38 36 19 7 - 13.8 27 13 14 0.0 38 36 19 7 - 14.5 26 13 13 0.1 10 39 25 26 - 17.1 24 12 13 0.4 15 37 33 11 - 11.0 22 14 8 -0.4 41 27 27 25 - 11.0 22 14 8 0.0.4 41 27 27 25 - 18.8 27 16 11 0.2 14 40 33 11 2.79 28.4 33 19 14 0.7 3 6 73 18 - 13.7 29 19 10 -0.5 3 14 67 67 16	Sample	Location	Gravity	Moisture		%)	<u></u>		+#4	#4 - #200	#200 - 0.002mm	-0.002mm	Soil Description
- 10.4 21 13 8 -0.3 21 46 27 6 2.78 10.9 24 14 11 -0.2 16 39 35 10 - 12.6 25 17 9 -0.4 38 36 19 7 - 13.8 27 13 14 0.0 38 36 19 7 - 14.5 26 13 13 0.1 10 39 25 26 - 17.1 24 12 13 0.4 15 37 33 15 - 11.0 22 14 8 -0.4 41 27 25 7 - 11.0 22 14 7 0.4 41 27 25 7 - 18.8 27 16 11 0.2 14 40 35 11 - 18.4 33 19 14 0.7 3 6 73 18 -	No.			Content (%)		PL	II.	П	% Gravel	% Sand	% Silt	%Clay	
2.78 10,9 24 14 11 -0.2 16 39 35 10 - 12.6 25 17 9 -0.4 38 36 19 7 - 13.8 27 13 14 0.0 38 36 19 7 - 14.5 26 13 14 0.0 38 36 19 7 - 17.1 24 12 13 0.4 15 37 33 15 - 11.0 22 14 9 -0.3 19 37 33 15 - 11.0 22 14 8 -0.4 41 27 25 7 - 11.0 22 14 4 2 25 65 11 - 18.8 27 16 11 0.2 14 40 35 11 - 18.8 27 16 11 0.7 3 6 73 11 - 13.7	TP95-1	Mill Site	,	10.4	21	13	∞	-0.3	21	46	27	9	Silty, gravelly SAND, trace clay
- 12.6 25 17 9 -0.4 38 36 19 7 - 13.8 27 13 14 0.0 38 36 19 7 - 14.5 26 13 13 0.1 10 39 25 26 - 17.1 22 14 9 -0.3 19 37 33 11 - 11.0 22 14 8 -0.4 41 27 25 - 18.8 27 16 11 0.2 14 40 35 11 - 18.8 27 16 11 0.2 14 40 35 11 - 18.8 27 16 11 0.2 14 40 35 11 - 18.8 27 16 11 0.2 14 40 35 11 - 18.8 27 16 11 0.2 14 60 16 17 - 18.8 27 16 11 0.2 14 60 16 17 - 18.8 27 16 11 0.2 14 60 16 17 - 18.8 27 16 11 0.2 14 60 16 17 - 18.8 27 16 11 0.2 14 60 16 17 - 18.8 27 16 11 0.2 14 60 17 - 18.8 27 16 11 0.2 14 60 16 17 - 18.8 27 10 0.5 3 14 07 16 17	TP95-7	Mill Site	2.78	10.9	24	14	11	-0.2	16	39	35	.01	Silty SAND, some gravel and clay
- 13.8 27 13 14 0.0 38 36 19 7 - 14.5 26 13 13 0.1 10 39 25 26 - 17.1 24 12 13 0.4 15 37 33 15 - 11.0 22 14 8 0.4 41 27 27 2.78 16.5 21 14 7 0.4 2 22 65 11 - 18.8 27 16 11 0.2 14 40 35 11 2.79 28.4 33 19 14 0.7 3 6 73 18 2.76 28.5 0 40 40 16 2.77 2.86 15 10 0.5 3 14 67 16	TP95-10	Bootjack Lake Road	•	12.6	25	17	6	÷.0-	. 38	36	19	7	GRAVEL and SAND, some silt, trace clay
n 2.73 11.1 22 14 9 -0.3 19 37 33 15 15 15 2.78 11.0 22 14 8 -0.4 41 27 27 25 7 15 2.78 16.5 21 14 7 0.4 2 2 22 65 11 1	TP95-18	Tailings/Reclaim Pipeline Route	•	13.8	27	13	7	0.0	38	36	19	7	GRAVEL and SAND, some silt, trace clay
n 2.73 11.1 22 14 9 -0.3 19 37 33 15 15 - 11.0 22 14 8 -0.4 41 27 25 7 25 7 2 2 16 11 0.2 14 40 35 11 11 0.2 14 8 0.7 3 3 11 11 0.2 18.8 27 16 11 0.2 14 40 35 11 18 2.79 28.4 33 19 14 0.7 3 6 73 18 18 2.76 28.5 0 0 40 46 14 14 15 17 2.76 28.5 0 40 40 46 11 16 11 0.5 11 17 17 18 18 18 19 10 -0.5 3 14 67 16 11	TP95-20	Polley Lake	1	14.5	56	13	13	0.1	10	39	25	92	Silty, clayey SAND, trace to some gravel
2.73 11.1 22 14 9 0.3 19 37 33 11 2.78 16.5 21 14 7 0.4 2 22 65 11 2.79 28.4 33 19 14 0.7 3 6 73 18 2.76 28.5 0 40 46 14 2.77 29 19 10 0.5 3 14 67 15 2.77 29 15 17 0.5 17 0.5 1 14 67 15 15 15 15 15 15 15 15 15 15 15 15 15	TP95-25	Polley Lake	ı	17.1	24	12	13	4.0	15	37	33	15	SAND and SILT, some gravel and clay
- 11.0 22 14 8 -0.4 41 27 25 7 2.78 16.5 21 14 7 0.4 2 22 65 11 - 18.8 27 16 11 0.2 14 40 35 11 2.79 28.4 33 19 14 0.7 3 6 73 18 2.76 28.5 - - - - - 0 40 46 14 - 13.7 29 19 10 -0.5 3 14 67 16 - 14 0.6 1 17 61 11	TP95-27	Perimeter Embankment Foundation	2.73	1:1	22	4	6	-0.3	19	37	33	=	SAND and SILT, some gravel and clay
2.78 .16.5 21 14 7 0.4 2 22 65 11 - 18.8 27 16 11 0.2 14 40 35 11 2.79 28.4 33 19 14 0.7 3 6 73 18 2.76 28.5 - - - - - 0 40 46 14 - 13.7 29 19 10 -0.5 3 14 67 16 - 14 07 16 17 61 17 61 21	TP95-31	East Ridge Borrow Area	ı	11.0	22	14	~	÷.ö.	41	27	25	7	Silty, sandy GRAVEL, trace clay
2.79 28.4 33 19 14 0.7 3 6 73 18 27 6 2.76 28.5 0 40 46 14 67 16 17 3 75 17 18 18 18 18 18 19 10 0.5 3 14 67 16 17 16 17 15 17 17 15 17 15 17 15 17 15 17 15 17 15 17 15 17 15 17 15 17 15 17 15 17 17 17 17 17 17 17 17 17 17 17 17 17	TP95-35	South Basin	2.78	.16.5	21	14	7	0.4	7	22	65	=======================================	Sandy SILT, some clay, trace gravel
2.79 28.4 33 19 14 0.7 3 6 73 18 2.76 28.5 - - - - - 0 40 46 14 - 13.7 29 19 10 -0.5 3 14 67 16 - 15 10 16 17 61 16	TP95-37	South Basin	•	18.8	27	16	=======================================	0.2	14	40	32	=	SAND and SILT, some gravel and clay
2.76 28.5 - - - 0 40 46 14 - 13.7 29 19 10 -0.5 3 14 67 16 - 15 16 16 16 17 61 16	TP95-38	Main Embankment Foundation	2.79	28.4	33	19	14	0.7	е	9	73	18	SILT, some clay, trace sand and gravel
27. 29 19 10 -0.5 3 14 67 16	TP95-39	Main Embankment Foundation	2.76	28.5		,	•		0	94	46	14	SILT and fine SAND, some clay
- 13.7 29 19 10 -0.5 3 14 67 16 16				,							•		نسم
2 2 25 1 20 15 14 06 1 15 22		Main Embankment Foundation	,	13.7	59	19		-0.5	ю	14	29	16	SILT, some clay and sand, trace gravel
17 10 11 1 10 10 11 10 10 11 10 10	TPB-13,14,16	TPB-13,14,16 Embankment & Pond Foundations	2.76	25.1	20	16	14	9.0	-	17	19	21	Clayey SILT, some sand, trace gravel

1. Samples TPB-1 and TPB-13,14,16 were selected for laboratory testwork in 1989 and have been reported for comparison.

Note:

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CONSULTING ENGINEER MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

DESIGN BASIS AND OPERATING CRITERIA

6.4 Water Quality Monitoring	 To ensure environmental compliance. Water quality samples taken at regular intervals from sediment ponds, drains (at drain monitor sump), groundwater monitoring wells, seepage
	ponds and tailings pond. • Upstream and downstream samples for impact analysis.
6.5 Hydrometeorology	 Operator weather station for input to water balance calculations. Precipitation (rain and snow). Evaporation. Air quality monitoring (dust, etc.).
6.6 Operational Monitoring	 Quantify operation of tailings storage facility. Rate of tailings accumulation in terms of mass and volume. Tailings characteristics and water recovery. Supernatant pond (depth, area and volume).
7.0 CLOSURE REQUIREMENTS	
7.1 General	Return impoundment to equivalent pre-mining use and productivity by establishing a wetland area adjacent to a final spillway and re-vegetating remainder of tailings surface with indigenous species of trees, shrubs and grasses adjacent to embankment grading to aquatic species along and adjacent to final pond.
7.2 Spillway	Two stage spillway with lower channel outlet designed to pass 1 in 200 yr. 24 hour flood event and upper wider outlet section designed to pass Probable Maximum Flood without overtopping embankments.

Notes:

1. The closure plan will remain flexible during operations to allow for future changes in the mine plan and to incorporate information from on-going reclamation programs.





MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

CONSEQUENCE CLASSIFICATION OF DAMS

J:\JOB\REPORT\1627\[2-TBL6-2.XLS]dam-class

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	[2-1 DEO-2.7CO]detti-Class	
	Potential Incremental	
Consequence	Conse	equences of Failure ^[a]
Category	Loss of Life	Economic, Social, Environmental
Very High	Large increase expected [b]	Excessive increase in social, economic and/or environmental losses.
High	Some increase expected [b]	Substantial increase in social, economic and/or environmental losses.
Low	No increase expected	Low social, economic and/or environmental losses.
Very Low	No increase	Small dams with minimal social, economic and/or environmental losses. Losses generally limited to the owner's property; damages to other property are acceptable to society.

- [a] Incremental to the impacts which would occur under the same natural conditions (flood, earthquake event) but without failure of the dam.

 The type of consequences (e.g. loss of life, or economic losses) with the highest rating determines which category is assigned to the structure.
- [b] The loss-of-life criteria which separates the High and Very High categories may be based on risks which are acceptable to society, taken to be 0.001 lives per year for each dam. Consistent with this tolerable societal risk the minimum criteria for a Very High Consequence dam (PMF and MCE) should result in an annual probability of failure of less than 1/100,000.

Knight Piésold Ltd. CONSULTING ENGINEERS MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

DESIGN BASIS AND OPERATING CRITERIA

<u> </u>	SASIS AND OF ERATING CIGIERIA
	Butt fusion welded HDPE with 30" DR15.5, 22" DR17 and 24" DR15.5.
Spigots	Movable discharge section placed on tailings embankment crest.
Flow Rate	Design throughput 900 tonnes/hr dry solids.
·	Slurry solids content 35%.
	• Design flow 19.6 cfs (0.55m ³ /s). Increases to 23.8cfs (0.67m ³ /s) at
	30% solids content with addition of 4.2cfs storm water runoff
	Waste dump and Millsite runoff will be added to tailings stream,
	increasing flow and decreasing solids content.
Spill Containment:	_
 Mill site to Bootjack Creek 	Pipeline laid in pipe containment channel. There is an overflow pond
	for the T2 dropbox.
 Bootjack Creek Crossing 	Pipeline sleeved in pipe containment channel.
- Bootjack Creek to TSF	Pipeline laid in pipe containment channel.
4.2 Reclaim Water System	
Function	Primary source of water for milling process. (Pump and Barege System Designed by Others.)
Reclaim Barge	Prefabricated pump station on barge in excavated channel in TSF.
	Local and remote control from Millsite.
Reclaim Pipeline	24" pipeline with a steel section at the reclaim barge and HDPE with
-	varying pressure ratings along length.
Reclaim Booster Pump Station	Prefabricated pump station located between TSF and Millsite.
	• Identical pumps, sensors and controls as reclaim barge for ease of
·	maintenance.
Spill Containment	See Item 4.1 above, all same for pipelines.
	Booster pump station has closed sump.
4.3 Seepage Recycle System	
Function	Return seepage and foundation drain flows to TSF.
Drain Monitoring Sumps	Flow quantity and water quality measurements on individual drains.
Seepage Collection Ponds	Sized to hold 10 times max. weekly seepage flow quantity.
	• Excavated in low permeability natural soil liner, operated as
	groundwater sink.
Seepage Recycle Pumps	Set in vertical pump sumps.
	Submersible pumps, system by Others.
	Pumps discharge back to TSF via 150 mm HDPE pipes.
5.0 MAKE-UP WATER SUPPLY	
5.1 General	
Function	To direct runoff from the Millsite and Southeast Sediment pond to the TSF,
	providing additional water for recycle to the mill. Also, to implement the
	Polley Lake Pump Station when and as required to meet the project Water
	Management Plan objectives.
5.2 Millsite Sump	
Catchment Area	Approx. 20 ha direct catchment, plus pit dewatering.
Design Storm	1.5 x 1 in 10 yr. 24 hour event runoff (6,000 m ³)
Sump Cross-Section	3:1 inside slope, 2:1 outside slope, 4m crest width.
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CONSULTING ENGINEER MOUNT POLLEY MINING CORPORATION **MOUNT POLLEY PROJECT**

DESIGN BASIS AND OPERATING CRITERIA

	ADIO AND OTERATINO CRITERIA
Normal Operating Level	1102.7 m
Maximum Operating Level	1106.2 m
Flow Control Structures	See Dwg. No. 1625.232 for layout details.
Discharge Pipe	300 mm HDPE DR 21 to plant or tailings line.
Flow Monitoring	None.
5.3 Southeast Sediment Pond	
Catchment Area	Approx. 150 ha direct catchment.
Design Storm	1 in 10 yr. 24 hour event runoff (25,000 m ³)
Sump Cross-Section	3:1 inside slope, 2:1 outside slope, 4m crest width.
Normal Operating Level	1054.5 m
Maximum Operating Level	1057.4 m
Flow Control Structures	See Dwg. No. 1625.232 for layout details.
Discharge Pipe	250mm HDPE DR 21 to Reclaim sump or T2 dropbox
Flow Monitoring	None.
5.4 Polley Lake Pump Station	Report and Drawings soon to be issued.
Max. Volume to be extracted	1,000,000 m ³ annually
Period for water extraction	Freshet
Max. Intake Velocity	0.11 m/s
Intake Screen Opening	0.1 inch (No. 8 Mesh wire cloth)
Spill Containment at Pump	Collection into a Holding Basin
Discharge Pipe	22 ½ inch ID, 350 ft of 19 ½ inch ID and 5200 ft of 17 ½ inch ID pipe.
Max. Flow	5,500 US GPM
Flow Monitoring	Flows in Hazeltine Creek, water level on Polley Lake, pumping hours times
	measured flow rate.
Security and Access	Signs for buried or submerged components, buoys attached to intake in Polley Lake.
6.0 INSTRUMENTATION AND MONITO	RING
6.1 General	
Function	To quantify environmental conditions and performance characteristics of the TSF to ensure compliance with design objectives.
6.2 Geotechnical Instrumentation and	
Monitoring	
Piezometers	 Measure pore pressures in drains, foundations, fill materials and tailings. Vibrating wire piezometers. Installed by qualified technical personnel.
	Three instrumentation planes for Main Embankment and one for Perimeter Embankment.
Survey Monuments	Deformation and settlement monitoring of embankments.
6.3 Flow Monitoring	 To provide data for on-going water balance calculations. Drain flows regularly monitored.
	Reclaim and seepage pump systems flow meters.
	Tailings output monitored at millsite.
	Streamflow monitoring.
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Association des Ingénieurs-Conseils du Canada

Knight Piésold Ltd. CONSULTING ENGINEERS MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

DESIGN BASIS AND OPERATING CRITERIA

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ITEM	DESIGN CRITERIA
1.0 GENERAL DESIGN CRITERIA	
Scope	Generally applicable to all components and structures.
Regulations	MEI
	MELP (Water Management Branch)
Codes and Standards	NBC and related codes
	CAN/CSA
	HSRC (Health, Safety and Reclamation Code for Mines in B.C.)
	ASTM
,	ACI
	ANSI
Design Life	14 Years
Operational Criteria:	
General	NBC where relevant
Rainfall/Precipitation:	Section 2.1 (Ref. No. 1625/1) and (Ref. No. 1624/1)
Seismic:	
DBE (operations)	M = 6.5, A max. = 0.037 g
MDE (closure)	M = 6.5, A max. = 0.065 g
2.0 TAILINGS BASIN	
Site Selection	Section 4.0, (Ref. No. 1625/1) and (Ref. No. 1621/1)
	Capacity and filling characteristics.
	Hydrology and downstream water usage.
	Hydrogeology and groundwater regime.
	 Aesthetics and visual impact. Foundation conditions.
	Construction requirement.
	Closure and reclamation requirements.
	Capital and operating costs.
a 1 1 1 C huisel Conditions	Section 5.0. (Ref. No. 1625/1) and (Ref. No. 1623/1).
Geological and Geotechnical Conditions	Compacted glacial till with frost protection layer required in areas with
Basin Liner	<2 m in-situ glacial till.
	Liner placed in 3 - 150 mm lifts.
•	• Liner compacted to 95% Std. Proctor max. dry density (ASTM D698)
	at optimum moisture content minus 1% to plus 2%.
Embankment Foundation Drains	Installed in Main Embankment Foundation.
Embankment Foundation Drums	Geotextile wrapped 1000 mm x 800 mm gravel/drain with 100 mm
	perforated CPT drain pipe.
	Drain conveyance nines are solid HDPE.
	Discharge to Main Embankment Seepage Collection Pond via Drain
	Monitoring Sump.
Stripping	Required at areas directly affected by construction (embankments,
ourkhang.	basin liners, seepage collection ponds, reclaim barge channel,
	stockniles, roads etc.).
	Remove organic soil to topsoil stockpiles.

Association des Ingénieurs-Conseils du Canada

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CONSULTING ENGINEERS MOUNT POLLEY MINING CORPORATION **MOUNT POLLEY PROJECT**

DESIGN BASIS AND OPERATING CRITERIA

2.0 TAILINGS TREE LANGE		ADIO INTO OTENATINO CATENIA
3.0 TAILINGS EMBANK	MENT	
Function		Storage of tailings and process water for design life.
		Provide storage for 24 hour PMP storm.
Embankment Crest Width		Provision for routing PMF at closure.
<u> </u>	<u> </u>	8 m starter dam and 8 m final dam.
Embankment Height (Max):	Starter	15 m (Crest El. 927 m)
To be desired to the second se	Final	53 m (Crest El. 965 m)
Embankment Crest Length:	Starter	1000 m
D	Final	4500 m
Design Tonnage		6,500,000 tpy (17,808) tpd
Solids Content of Tailings Str		35% (before Millsite and waste dump runoff added to tailings stream)
Freeboard:	Operations	24 hour PMP event (679,000 m ³) plus 1.0m wave runup on 2.5 million m ³ operational storage pond.
	Closure	Sufficient to provide routing of PMF plus wave run-up.
Storage Capacity		84.5 million tonnes.
Tailings Density:	Year 1	1.1 t/m³
	Year 2	1.2 t/m³
	Year 3-13	1.3 t/m ³
Tailings Specific Gravity		2.78
Borrow Material Properties		Section 6.3.3, 1995 Site Investigation Report (Ref No 1623/1), and (Ref No
		1625/1).
Construction Diversion		Not required.
Emergency Spillway Flows:	Operations	Not required.
	Closure	Design flow for routing PMF event.
Filling Rate		Figure 6.1 and 6.2, Figure 6.3 (Ref. No. 1625/1).
Fill Material Properties		Drg. No. 1625.212
Compaction Requirements		Drg. No. 1625.211
Geotechnical Data		Sections 6.3, 6.4 and 6.5, 1995 Site Investigation Report (Ref. No. 1623/1),
		and Section 5.1 (Ref. No. 1625/1).
Stability Analysis		Section 6.8 and (Ref. No. 1625/1).
Seepage Analysis		Section 6.9 and (Ref. No. 1625/1).
Sediment Control		Primary control from Main Embankment. Main Embankment Seepage Collection Pond provides secondary sediment control.
Seepage Control		Seepage collection ponds and pumpback well systems.
Seismic Parameters		Section 2.3, and (Ref. No. 1625/1).
Spillway Discharge Capacity		Not required during operations.
Settlement		Section 6.6 and (Ref. No. 1625/1).
Surface Erosion Protection		Revegetation with grasses on final embankment slope.
4.0 PIPEWORKS		
4.1 Tailings Delivery and D	ischarge	Section 8.0 and (Ref. No. 1625/1).
Pipework		
Function		Transport tailings slurry and mill site and waste dump runoff to Tailings
		Storage Facility (TSF).
Tailings Pipeline		Free draining, gravity flow pipeline.

Association des Ingénieurs-Conseils du Canada

TABLE 5.1

MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

SUMMARY OF RECENT GEOLOGICAL INVESTIGATIONS

J:\JOB\DATA\1627\MISC\INVSUM.XLS

lentification		l		
Label	Field Name	Date	Location Description	Type of Investigation
5H	5H	23-Nov-96	Original Borrow Area	Original Borrow Exploration Trench
ВН5Н	BH5H	13-Jan-97	Original Borrow Area	Original Borrow Drilthole
51	51	24-Nov-96	Original Borrow Area	Original Borrow Exploration Trench
BH5I	BH5I	13-Jan-97	Original Borrow Area	Original Borrow Drillhole
6E	6E	22-Nov-96	Original Borrow Area	Original Borrow Exploration Trench
6F	6F	22-Nov-96	Original Borrow Area	Original Borrow Exploration Trench
6G	,6G	23-Nov-96	Original Borrow Area	Original Borrow Exploration Trench
6G-A	6G-A		Original Borrow Area	Original Borrow Exploration Trench
BH6G	BH6G	5-Jan-97	Original Borrow Area	Original Borrow Drillhole
6H	6H	23-Nov-96	Original Borrow Area	Original Borrow Exploration Trench
61	61	24-Nov-96	Original Borrow Area	Original Borrow Exploration Trench
6I-A	6I-A		Original Borrow Area	Original Borrow Exploration Trench
BH6I	BH61	5-Jan-97	Original Borrow Area	Original Borrow Drillhole
7E	7E	24-Nov-96	Original Borrow Area	Original Borrow Exploration Trench
7F	7F	23-Nov-96	Original Borrow Area	Original Borrow Exploration Trench
. 7G	7G	23-Nov-96	Original Borrow Area	Original Borrow Exploration Trench
BH7G	BH7G	5-Jan-97	Original Borrow Area	Original Borrow Drillhole
BH7.5G	BH7.5G	4-Jan-97	Original Borrow Area	Original Borrow Drillhole
7H	7H	23-Nov-96	Original Borrow Area	Original Borrow Exploration Trench
7H-A	7H-A		Original Borrow Area	Original Borrow Exploration Trench
вн7н	ВН7Н	5-Jan-97	Original Borrow Area	Original Borrow Drillhole
BH7.5H	BH7.5H	4-Jan-97	Original Borrow Area	Original Borrow Drillhole
71	71	24-Nov-96	Original Borrow Area	Original Borrow Exploration Trench
BH7I	BH7I	4-Jan-97	Original Borrow Area	Original Borrow Drillhole
7J-A	7J-A		Original Borrow Area	Original Borrow Exploration Trench
8E	8E	24-Nov-96	Original Borrow Area	Original Borrow Exploration Trench
8F	8F	24-Nov-96	Original Borrow Area	Original Borrow Exploration Trench
BH8F	BH8F	4-Jan-97	Original Borrow Area	Original Borrow Drillhole
8G	8G	24-Nov-96	Original Borrow Area	Original Borrow Exploration Trench
BH8G	BH8G	3-Jan-97	Original Borrow Area	Original Borrow Drillhole
8H	8H	24-Nov-96	Original Borrow Area	Original Borrow Exploration Trench
BH8H	BH8H	3-Jan-97	Original Borrow Area	Original Borrow Drillhole
81	8I	24-Nov-96	Original Borrow Area	Original Borrow Exploration Trench
BH8I	BH8I	3-Jan-97	Original Borrow Area	Original Borrow Drillhole Original Borrow Exploration Trench
9E	9E	24-Nov-96	Original Borrow Area	Original Borrow Exploration Trench
9F	9F	24-Nov-96	Original Borrow Area	Original Borrow Exploration Trench
9G	9G	24-Nov-96	Original Borrow Area	Original Borrow Exploration Trench
9H	9H	24-Nov-96	Original Borrow Area	Original Borrow Exploration Trench
91	91	24-Nov-96	Original Borrow Area	Original Borrow Drillhole
BH10F	BH10F	11-Jan-97	Original Borrow Area Original Borrow Area	Original Borrow Exploration Trench
TP10G	TP10G BH10H	17-Jan-97	Original Borrow Area	Original Borrow Drillhole
BH10H		10-Jan-97	Original Borrow Area	Original Borrow Exploration Trench
TP10I	TP10I	17-Jan-97	Original Borrow Area	Original Borrow Exploration Trench
TP11F	TP11F	15-Jan-97		Original Borrow Exploration Trench
TP11G	TP11G	17-Jan-97	Original Borrow Area	Original Borrow Exploration Trench
TP11H	TP11H	18-Jan-97	Original Borrow Area	Original Borrow Exploration Trench
TP11I	TP111	17-Jan-97	Original Borrow Area Original Borrow Area	Original Borrow Exploration Trench
TP12F	TP12F	15-Jan-97	Original Borrow Area	Original Borrow Exploration Trench
TP12G	TP12G	16-Jan-97	Original Borrow Area	Original Borrow Drillhole
BH12H	BH12H	5-Jan-97	Original Borrow Area	Original Borrow Exploration Trench
TP12H	TP12H	1/181997	Original Borrow Area	Original Borrow Drillhole
BH12.5H	BH12.5H	6-Jan-97	Original Borrow Area	Original Borrow Exploration Trench
TP12i	TP12I	19-Jan-97		Original Borrow Drillhole
BH13F	BH13F	13-Jan-97	Original Borrow Area	Original Borrow Exploration Trench
TP13G	TP13G	16-Jan-97	Original Borrow Area	Original Borrow Drillhole
BH13H	BH13H	6-Jan-97	Original Borrow Area	Original Borrow Exploration Trench
TP13I	TP13I	19-Jan-97	Original Borrow Area	Original Borrow Exploration Trench
TP14F	TP14F	15-Jan-97	Original Borrow Area	
TP14G	TP14G BH14H	16-Jan-97	Original Borrow Area	Original Borrow Exploration Trench Original Borrow Drillhole
BH14H		9-Jan-97	Original Borrow Area	



TABLE 5.1

MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

SUMMARY OF RECENT GEOLOGICAL INVESTIGATIONS

6/5/97 15:40

Identification				
Label	Field Name	Date	Location Description	Type of Investigation
TP15F	TP15F	15-Jan-97	Original Borrow Area	Original Borrow Exploration Trench
TP15G	TP15G	16-Jan-97	Original Borrow Area	Original Borrow Exploration Trench
BH15H	BH15H	8-Jan-97	Original Borrow Area	Original Borrow Drillhole
TP15I	TP15I	22-Jan-97	Original Borrow Area	Original Borrow Exploration Trench
BH16F	BH16F	12-Jan-97	Original Borrow Area	Original Borrow Drillhole
TP16G	TP16G	16-Jan-97	Original Borrow Area	Original Borrow Exploration Trench
BH16H	BH16H	8-Jan-97	Original Borrow Area	Original Borrow Drillhole
TP16i	TP16l	18-Jan-97	Original Borrow Area	Original Borrow Exploration Trench
	TP95ME-1	15-May-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 1				Foundation Exploration Trench
95- 2	TP95ME-2	15-May-95	Main Embankment (Previous Investigations)	
95- 3	TP95ME-3	15-May-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 4	TP95ME-4	3-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 5	TP95ME-5	3-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 6	TP95ME-6	3-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 7	TP95ME-7	3-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 8	TP95ME-8	3-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 9	TP95ME-9	4-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 10	TP95ME-10	4-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 11	TP95ME-11	4-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 12	TP95ME-12	4-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 13	TP95ME-13	4-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 14	TP95ME-14	4-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 15	TP95ME-15	4-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 16	TP95ME-16	4-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 17	TP95ME-17	4-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 18	TP95ME-18	4-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 19	TP95ME-19	4-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 20	TP95ME-20	4-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 21	TP95ME-21	4-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 22	TP95ME-22	4-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 23	TP95ME-23	4-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 24	TP95ME-24	4-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 25	TP95ME-25	5-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 26	TP95ME-26	5-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 27	TP95ME-27	5-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 28	TP95ME-28	5-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 29	TP95ME-29	5-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 30	TP95ME-30	5-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 31	TP95ME-31	5-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 32	TP95ME-32	5-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 33	TP95ME-33	5-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
95- 34	TP95ME-34	5-Oct-95	Main Embankment (Previous Investigations)	Foundation Exploration Trench
05 05	P40	E 0-4 05	Main Embankment SCP (Previous Investigations)	SCP (Pond) Exploration Trench
95- 35	S19	5-Oct-95		SCP (Pond) Exploration Trench
95- 36	\$20	5-Oct-95	Main Embankment SCP (Previous Investigations)	
95- 37	\$20.5	5-Oct-95	Main Embankment SCP (Previous Investigations)	SCP (Pond) Exploration Trench SCP (Pond) Exploration Trench
95- 38	S21	21-Sep-95	Main Embankment SCP (Previous Investigations)	SCP (Pond) Exploration Trench
95- 39	\$21.2	5-Oct-95	Main Embankment SCP (Previous Investigations)	
95- 40	\$22	21-Sep-95	Main Embankment SCP (Previous Investigations)	SCP (Pond) Exploration Trench

Note: 1995 Previous Investigations are included because they have been used in the confirmation of the geological and geotechnical conditions.





TABLE 5.1

MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

SUMMARY OF RECENT GEOLOGICAL INVESTIGATIONS

:\JOB\DATA\1627\M	IISC\INVSUM.XLS			6/5/97 15:40
Identification				95
Label	Field Name	Date	Location Description	Type of Investigation
96- 26	ET-96/09/03-1	3-Sep-96	Upper Basin Liner	UBL Exploration Trench
96- 27	ET-96/09/03-2	3-Sep-96	Upper Basin Liner	UBL Exploration Trench
96- 28	ET-96/09/03-3	3-Sep-96	Upper Basin Liner	UBL Exploration Trench
96- 29	ET-96/08/14-1	14-Aug-96	Lower Basin Liner	LBL Exploration Trench
96- 30	ET-96/08/14-2	14-Aug-96	Lower Basin Liner	LBL Exploration Trench
96- 31	ET-96/08/14-3	14-Aug-96	Lower Basin Liner	LBL Exploration Trench
96- 32	ET-96/08/14-4	14-Aug-96	Lower Basin Liner	LBL Exploration Trench
96- 33	ET-96/08/14-5	14-Aug-96	Lower Basin Liner	LBL Exploration Trench
96- 34	ET-96/08/14-6	14-Aug-96	Lower Basin Liner	LBL Exploration Trench
	ET-96/08/14-7	14-Aug-96	Lower Basin Liner	LBL Exploration Trench
96- 35	ET-96/08/14-8	14-Aug-96	Lower Basin Liner	LBL Exploration Trench
96- 36		14-Aug-96	Lower Basin Liner	LBL Exploration Trench
96- 37	ET-96/08/14-9		Lower Basin Liner	LBL Exploration Trench
96- 38	ET-96/08/14-10	14-Aug-96		LBL Exploration Trench
96- 39	ET-96/08/14-11	14-Aug-96	Lower Basin Liner	LBL Exploration Trench
96- 40	ET-96/08/14-12	14-Aug-96	Lower Basin Liner	LBL Exploration Trench
96- 41	ET-96/08/14-13	14-Aug-96	Lower Basin Liner	LBL Exploration Trench
96- 42	ET-96/08/14-14	14-Aug-96	Lower Basin Liner	LBL Exploration Trenen
				Alternate Borrow Exploration Trench
96- 43	ET-96/09/13-1	13-Sep-96	Alternate Borrow Area	Alternate Borrow Exploration Trench
96- 44	ET-96/09/13-2	13-Sep-96	Alternate Borrow Area	Alternate Borrow Exploration Trench
96- 45	ET-96/09/13-3	13-Sep-96	Alternate Borrow Area	Alternate Borrow Exploration Trench
96- 46	ET-96/09/13-4	13-Sep-96	Alternate Borrow Area	Alternate Borrow Exploration Trench
96- 47	ET-96/09/13-5	13-Sep-96	Alternate Borrow Area	Alternate Borrow Exploration Trench
90- 41	<u> </u>	, o dop 0-		
00.48	ET-96/09/16-1	16-Sep-96	Alternate Borrow Area	Alternate Borrow Exploration Trench
96- 48	ET-96/09/16-2	16-Sep-96	Alternate Borrow Area	Alternate Borrow Exploration Trench
96- 49		16-Sep-96	Alternate Borrow Area	Alternate Borrow Exploration Trench
96- 50	ET-96/09/16-3	16-Sep-96	Alternate Borrow Area	Alternate Borrow Exploration Trench
96- 51	ET-96/09/16-4	10-3ep-30	Attended political and a second political and	
		40.0== 00	Alternate Borrow Area	Alternate Borrow Exploration Trench
96- 52	ET-96/09/19-1	19-Sep-96	Alternate Borrow Area	Alternate Borrow Exploration Trench
96- 53	ET-96/09/19-2	19-Sep-96		Alternate Borrow Exploration Trench
96- 54	ET-96/09/19-3	19-Sep-96	Alternate Borrow Area	Alternate Borrow Exploration Trench
96- 55	ET-96/09/19-4	19-Sep-96	Alternate Borrow Area	
		ļ		
	ET-96/08/31-1		(5.1.55.5)	Filter Sand Exploration Trench
96- 56	ET-96/08/31-2	31-Aug-96	Filter Sand Investigations (East of P. E.)	Filter Sand Exploration Trench
96- 57	ET-96/08/31-3	31-Aug-96	Filter Sand Investigations (East of P. E.)	Filter Sand Exploration Trench
96- 58	ET-96/08/31-4	31-Aug-96	Filter Sand Investigations (East of P. E.)	Filter Salid Exploration Trends
				O I Tarabas Transh
96- 59	ET-96/09/01-1	1-Sep-96	Filter Sand Investigations (East of P. E.)	Filter Sand Exploration Trench
96- 60	ET-96/09/01-2	1-Sep-96	Filter Sand Investigations (East of P. E.)	Filter Sand Exploration Trench
96- 61	ET-96/09/01-3	1-Sep-96	Filter Sand Investigations (East of P. E.)	Filter Sand Exploration Trench
30- 01		, , , <u>, , , , , , , , , , , , , , , , </u>		
	ET-06/00/02 1	2-Sep-96	Filter Sand Investigations (East of P. E.)	Filter Sand Exploration Trench
96- 62	ET-96/09/02-1	2-Sep-96	Filter Sand Investigations (East of P. E.)	Filter Sand Exploration Trench
96- 63	ET-96/09/02-2		Filter Sand Investigations (East of P. E.)	Filter Sand Exploration Trench
96- 64	ET-96/09/02-3	2-Sep-96	Filter Sand Investigations (East of P. E.)	Filter Sand Exploration Trench
96- 65	ET-96/09/02-4	2-Sep-96	Filter Cond Investigations (East of P. E.)	Filter Sand Exploration Trench
96- 66	ET-96/09/02-5	2-Sep-96	Filter Sand Investigations (East of P. E.)	Filter Sand Exploration Trench
96- 67	ET-96/09/02-6	2-Sep-96	Filter Sand Investigations (East of P. E.)	Filter Sand Exploration Trench
96- 68	ET-96/09/02-7	2-Sep-96	Filter Sand Investigations (East of P. E.)	Filter Sand Exploration Trench
96- 69	ET-96/09/02-8	2-Sep-96	Filter Sand Investigations (East of P. E.)	Filler Saild Exploration Trongit
F				Tomber Transh
96- 70	ET96-1	11-Jul-96	West end of Reclaim Barge Channel	Foundation Exploration Trench
	ET96-2	11-Jul-96	West end of Reclaim Barge Channel	Foundation Exploration Trench
96- 71		11-Jul-96	West end of Reclaim Barge Channel	Foundation Exploration Trench
96- 72	ET96-3	11-Jul-96	West end of Reclaim Barge Channel	Foundation Exploration Trench
96- 73	ET96-4		West end of Reclaim Barge Channel	Foundation Exploration Trench
96- 74	ET96-5	11-Jul-96	West end of Reclaim Barge Channel	Foundation Exploration Trench
96- 75	ET96-6	11-Jul-96		Foundation Exploration Trench
96- 76	ET96-7	11-Jul-96	West end of Reclaim Barge Channel	Foundation Exploration Trench
		1 44 14 06	West end of Reclaim Barge Channel	Troungation Exploration Trench
96- 77	ET96-8	11-Jul-96 11-Jul-96	West end of Reclaim Barge Channel	Foundation Exploration Trench

Knight Piésold Ltd. CONSULTING ENGINEERS

TABLE 5.1

MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

SUMMARY OF RECENT GEOLOGICAL INVESTIGATIONS

Identification	'MISC\INVSUM.XLS		<u> </u>	. 6/5/97
Label	Field Name	Date	Location Description	Type of lawardination
96- 79	ET96-10	12-Jul-96	West end of Reclaim Barge Channel	Type of Investigation
96- 80	ET96-11	12-Jul-96 12-Jul-96	West end of Reclaim Barge Channel West end of Reclaim Barge Channel	Foundation Exploration Trench
96- 81	ET96-12	12-Jul-96	West end of Reclaim Barge Channel	Foundation Exploration Trench
96- 82	ET96-13	12-Jul-96	West end of Reclaim Barge Channel	Foundation Exploration Trench
96- 83	ET96-15	12-Jul-96	West end of Reclaim Barge Channel	Foundation Exploration Trench
96- 84	ET-96/09/23-1	23-Sep-96	Original Borrow Area	Foundation Exploration Trench
96- 85	ET-96/09/23-2	23-Sep-96	Original Borrow Area	Original Borrow Exploration Trench
	2. 50/00/20 2	20-0cp-30	Original Bollow Alea	Original Borrow Exploration Trench
96- 86	ET-96/09/24-1	24-Sep-96	Original Borrow Area	Original Borrow Exploration Trench
96- 87	ET-96/09/24-2	24-Sep-96	Original Borrow Area	
96- 88	ET-96/09/24-3	24-Sep-96	Original Borrow Area	Original Borrow Exploration Trench
96- 89	ET-96/09/24-4	24-Sep-96	Original Borrow Area	Original Borrow Exploration Trench
96- 90	ET-96/09/24-5	24-Sep-96	Original Borrow Area	Original Borrow Exploration Trench
96- 91	ET-96/09/24-6	24-Sep-96	Original Borrow Area	Original Borrow Exploration Trench
96- 92	ET-96/09/24-7	24-Sep-96	Original Borrow Area	Original Borrow Exploration Trench
			Original Borrow Area	Original Borrow Exploration Trench
96- 93	ET-96/08/10-1	8-Oct-96	Original Borrow Area	Original Parray Contaction To
96- 94	ET-96/08/10-2	8-Oct-96	Original Borrow Area	Original Borrow Exploration Trench
·		1 2 3 3 3	James Dollott Filed	Original Borrow Exploration Trench
96- 95	ET-96/08/15-1	15-Aug-96	Original Borrow Area	Original Powers Evaluation Transfer
96- 96	ET-96/08/15-2	15-Aug-96	Original Borrow Area	Original Borrow Exploration Trench
96- 97	ET-96/08/15-3	15-Aug-96	Original Borrow Area	Original Borrow Exploration Trench Original Borrow Exploration Trench
96- 98	ET-96/08/15-4	15-Aug-96	Original Borrow Area	Original Borrow Exploration Trench
96- 99	ET-96/08/15-5	15-Aug-96	Original Borrow Area	Original Borrow Exploration Trench
96- 100	ET-96/08/15-6	15-Aug-96	Original Borrow Area	Original Borrow Exploration Trench Original Borrow Exploration Trench
96- 101	ET-96/08/15-7	15-Aug-96	Original Borrow Area	
96- 102	ET-96/08/15-8	15-Aug-96	Original Borrow Area	Original Borrow Exploration Trench
96- 103	ET-96/08/15-9	15-Aug-96	Original Borrow Area	Original Borrow Exploration Trench
96- 104	ET-96/08/15-10	15-Aug-96	Original Borrow Area	Original Borrow Exploration Trench Original Borrow Exploration Trench
96- 105	ET-96/08/15-11	15-Aug-96	Original Borrow Area	Original Borrow Exploration Trench
96- 106	ET-96/08/15-12	15-Aug-96	Original Borrow Area	Original Borrow Exploration Trench
			2010// 100	Original Borrow Exploration Trends
96- 107	ET-96/08/16-1	16-Aug-96	Original Borrow Area	Original Borrow Exploration Trench
96- 108	ET-96/08/16-2	16-Aug-96	Original Borrow Area	Original Borrow Exploration Trench
			3.13.13.13.13.13.13.13.13.13.13.13.13.13	Original Borrow Exploration Trench
96- 109	ET-96/10/29-1	29-Oct-96	Main Embankment (D/S of Final Toe)	Foundation Exploration Trench
96- 110	ET-96/10/29-2	29-Oct-96	Main Embankment (D/S of Final Toe)	Foundation Exploration Trench
			The state of the s	Toditoation Exploration Trench
96-111	ET-96/10/24-1	24-Oct-96	Filter Sand Investigations (East of P. E.)	Filter Sand Exploration Trench
96- 112	ET-96/10/24-2		Filter Sand Investigations (East of P. E.)	Filter Sand Exploration Trench
96- 113	ET-96/10/24-3	24-Oct-96	Filter Sand Investigations (East of P. E.)	Filter Sand Exploration Trench
96- 114	ET-96/10/24-4	24-Oct-96	Filter Sand Investigations (East of P. E.)	Filter Sand Exploration Trench
96- 115	ET-96/10/24-5		Filter Sand Investigations (East of P. E.)	Filter Sand Exploration Trench
				Titles dand Exploration Trench
96- 116	TP-96/08/16-Barge1	16-Aug-96	Reclaim Barge Channel	Foundation Exploration Trench
96- 117	TP-96/08/16-Barge2		Reclaim Barge Channel	Foundation Exploration Trench
96- 118	TP-96/08/16-Barge3	16-Aug-96	Reclaim Barge Channel	Foundation Exploration Trench
96- 119	TP-96/08/16-Barge4	29-Oct-96	Reclaim Barge Channel	Foundation Exploration Trench
			<u> </u>	Controduct Exploration 115/101
TP3H	TP3H	19-Jan-97	Original Borrow Area	Original Borrow Exploration Trench
4E	4E		Original Borrow Area	Original Borrow Exploration Trench
4F	4F		Original Borrow Area	Original Borrow Exploration Trench
4G	4G		Original Borrow Area	Original Borrow Exploration Trench
TP4G	TP4G		Original Borrow Area	Original Borrow Exploration Trench
4H	4H		Original Borrow Area	
41	41		Original Borrow Area	Original Borrow Exploration Trench
5E	5E		Original Borrow Area	Original Borrow Exploration Trench
5F	5F		Original Borrow Area	Original Borrow Exploration Trench
			Original Borrow Area	Original Borrow Exploration Trench
BH5F	6737 1			
BH5F 5G	BH5F 5G		Original Borrow Area	Original Borrow Drillhole Original Borrow Exploration Trench





TABLE 2.4

MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

USUAL MINIMUM CRITERIA FOR DESIGN EARTHQUAKES

Consequence	Maximum Design Earthquake (MDE)		
Category	Deterministically	Probabilistically Derived	
	Derived	(Annual exceedence probability)	
Very High	MCE ^{[a][b][c]}	1/10,000 ^{[b][c]}	
High	50% to 100% MCE ^{[d][e]}	1/1000 to 1/10,000 ^[e]	
Low	[t]	1/100 to 1/1000 ^(f)	

^a For a recognized fault or geographically defined tectonic province, the Maximum Credible Earthquake (MCE) is the largest reasonably conceivable earthquake that appears possible. For a dam site, MCE ground motions are the most severe ground motions capable of being produced at the site under the presently known or interpreted tectonic framework.

In Hydro-Quebec's practice, the MDE for Very High Consequence structures involves a combination of deterministic and probabilistic approaches that reflect current knowledge of seismo-tectonic conditions in Eastern Canada. Hydro-Quebec's deterministically derived MDE magnitude is the maximum historically recorded earthquake, increased by one-half magnitude, while their probabilistically derived earthquake has an estimated probability of exceedence of 1/2000

^c An appropriate level of conservatism shall be applied to the factor of safety calculated from these loads, to reduce the risks of dam failure to tolerable values. Thus, the probability of dam failure could be much lower than the probability of extreme event loading.

^d MDE firm ground accelerations and velocities can be taken as 50% to 100% of MCE values. For design purposes the magnitude should remain the same as the MCE.

^e In the High Consequence category, the MDE is based on the consequences of failure. For example, if one incremental fatality would result from failure, an AEP of 1/1000 could be acceptable, but for consequences approaching those of a Very High Consequence dam, design earthquakes approaching the MCE would be required.

^f If a Low Consequence structure cannot withstand the minimum criteria, the level of upgrading may be determined by economic risk analysis, with consideration of environmental and social impacts.

Knight Piésold Ltd. CONSULTING ENGINEERS

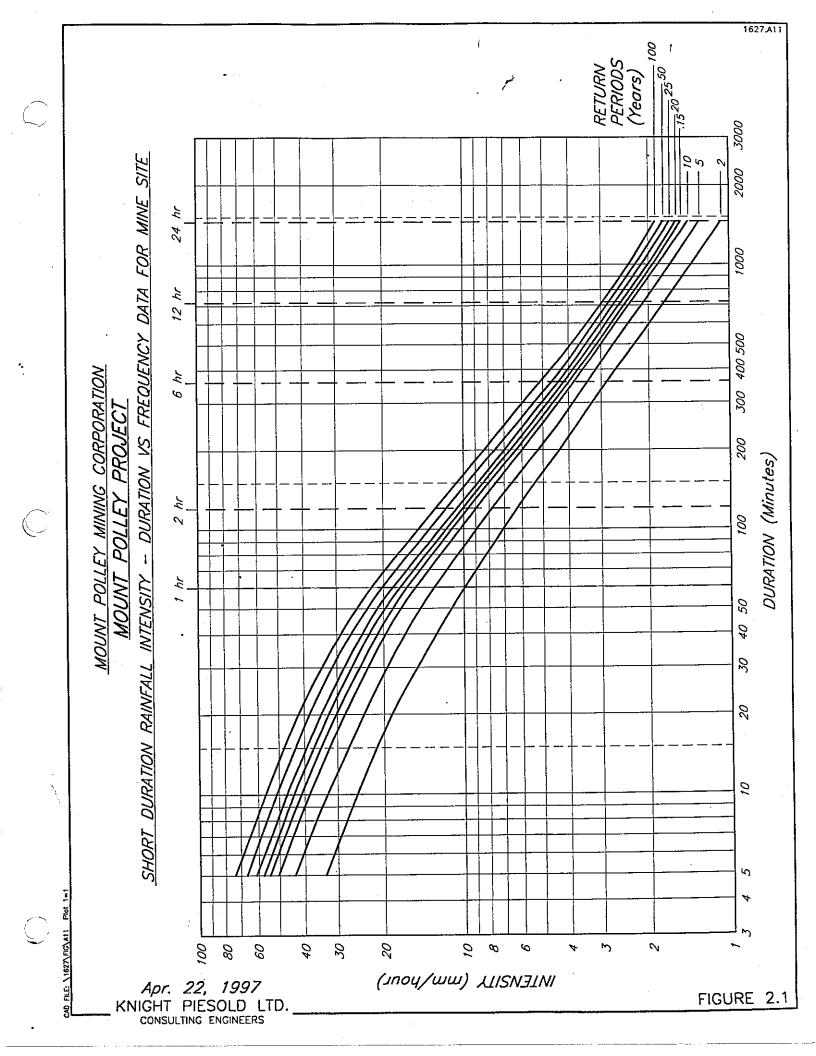
TABLE 5.1

MOUNT POLLEY MINING CORPORATION MOUNT POLLEY PROJECT

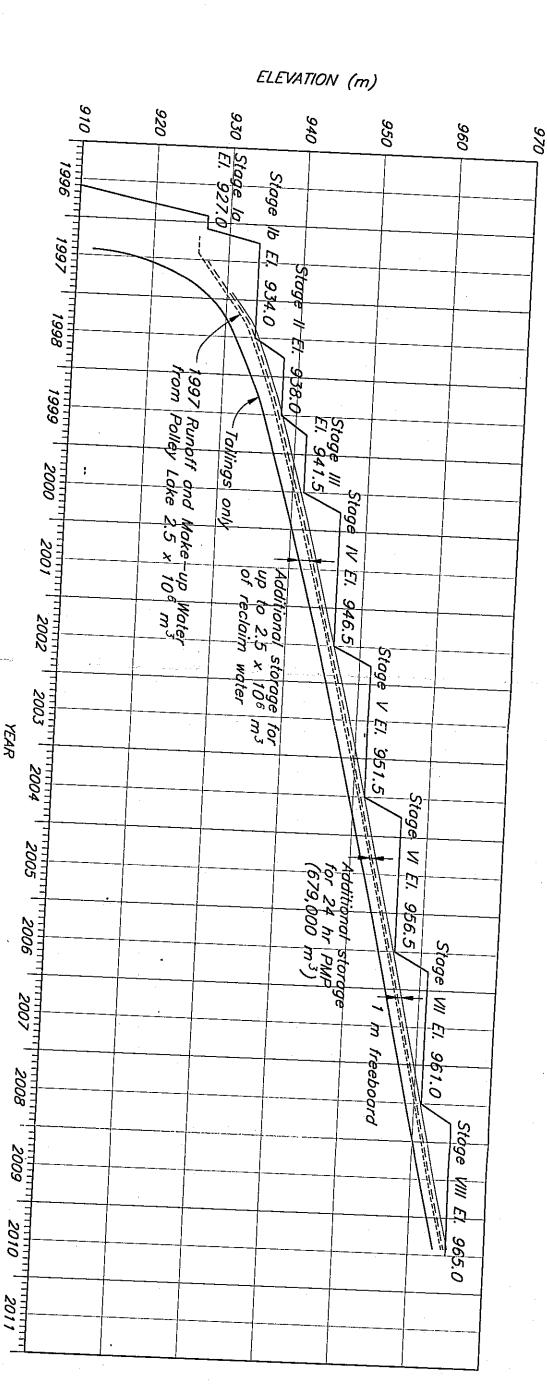
SUMMARY OF RECENT GEOLOGICAL INVESTIGATIONS

J:\JOB\DATA\1627\N	VISCHNVSUM.XLS			6/5/97 15:4
Identification	:			
Label	Field Name	Date	Location Description	Type of Investigation
96- A1	96- A1	21-Jul-96	Main Embankment	Foundation Auger/SPT Drillhole
96- B1	96- B1	21-Jul-96	Main Embankment	Foundation Auger/SPT Drillhole
96- C1	96- C1	21 -J ul-96	Main Embankment	Foundation Auger/SPT Drillhole
CPT 96-1	CPT 96-1	27-Jul-96	Main Embankment	Foundation CPT (Cone) Drillhole
CPT 96-2	CPT 96-2	27-Jul-96	Main Embankment	Foundation CPT (Cone) Drillhole
CPT 96-3	CPT 96-3	27-Jul-96	Main Embankment	Foundation CPT (Cone) Drillhole
CPT 96-4	CPT 96-4	28-Jul-96	Main Embankment	Foundation CPT (Cone) Drillhole
CPT 96-5	CPT 96-5	28-Jul-96	Main Embankment	Foundation CPT (Cone) Drillhole
PRW 96-1	PRW 96-1	28-Jul-96	Main Embankment	Foundation Auger Pressure Relief Well
PRW 96-2	PRW 96-2	27-Jul-96	Main Embankment	Foundation Auger Pressure Relief Well
PRW 96-3	PRW 96-3	27-Jul-96	Main Embankment	Foundation Auger Pressure Relief Well
PRW 96-4	PRW 96-4	27-Jul-96	Main Embankment	Foundation Auger Pressure Relief Well
PRT-1	PRT-1	10-Aug-96	Main Embankment	Foundation Pressure Relief Trench
PRT-2	PRT-2	13-Aug-96	Main Embankment	Foundation Pressure Relief Trench
PRT-3	PRT-3	24-Sep-96	Main Embankment	Foundation Pressure Relief Trench
PRT-4	PRT-4	28-Sep-96	Main Embankment	Foundation Pressure Relief Trench
		·		
GW96-1A	GW96-1A	29-Nov-96	Perimeter Embankment (D/S of Seepage Collection Pond)	Groundwater Monitoring Well Drillhole
GW96-1B	GW96-1B	29-Nov-96	Perimeter Embankment (D/S of Seepage Collection Pond)	Groundwater Monitoring Well Drillhole
GW96-2A	GW96-2A	10-Nov-96	East of Perimeter Embankment	Groundwater Monitoring Well Drillhole
GW96-2B	GW96-2B	10-Nov-96	East of Perimeter Embankment	Groundwater Monitoring Well Drillhole
GW96-3A	GW96-3A	4-Nov-96	Main Embankment (D/S of Seepage Collection Pond)	Groundwater Monitoring Well Drillhole
GW96-3B	GW96-3B	4-Nov-96	Main Embankment (D/S of Seepage Collection Pond)	Groundwater Monitoring Well Drillhole
GW96-4A	GW96-4A	18-Nov-96	Right Abutment of Final Main Embankment	Groundwater Monitoring Well Drillhole
GW96-4B	GW96-4B	18-Nov-96	Right Abutment of Final Main Embankment	Groundwater Monitoring Well Drillhole
GW96-5A	GW96-5A	21-Nov-96	Northwest side of Tailings Storage Facility	Groundwater Monitoring Well Drillhole
GW96-5B	GW96-5B	21-Nov-96	Northwest side of Tailings Storage Facility	Groundwater Monitoring Well Drillhole
GW96-9	GW96-9	21-Nov-96	Main Embankment (D/S of Final Toe)	Groundwater Monitoring Well Drillhole
96- 1	ET-96/08/19-1	19-Aug-96	Upper Basin Liner	UBL Exploration Trench
96- 2	ET-96/08/19-2	19-Aug-96	Upper Basin Liner	UBL Exploration Trench
96- 3	ET-96/08/19-3	19-Aug-96	Upper Basin Liner	UBL Exploration Trench
96- 4	ET-96/08/19-4	19-Aug-96	Upper Basin Liner	UBL Exploration Trench
96- 5	ET-96/08/19-5	19-Aug-96	Upper Basin Liner	UBL Exploration Trench
96- 6	ET-96/08/19-6	19-Aug-96	Upper Basin Liner	UBL Exploration Trench
96- 7	ET-96/08/19-7	19-Aug-96	Upper Basin Liner	UBL Exploration Trench
96- 8	ET-96/08/19-8	19-Aug-96	Upper Basin Liner	UBL Exploration Trench
96- 9	ET-96/08/19-9	19-Aug-96	Upper Basin Liner	UBL Exploration Trench
96- 10	ET-96/08/19-10	19-Aug-96	Upper Basin Liner	UBL Exploration Trench
96- 11	ET-96/08/20-1	20-Aug-96	Upper Basin Liner	UBL Exploration Trench
96- 12	ET-96/08/20-2		Upper Basin Liner	UBL Exploration Trench
96- 13	ET-96/08/20-3	20-Aug-96	Upper Basin Liner	UBL Exploration Trench
96- 14	ET-96/08/20-4	20-Aug-96	Upper Basin Liner	UBL Exploration Trench
96- 15	ET-96/08/20-5	20-Aug-96	Upper Basin Liner	UBL Exploration Trench
1		_		
96- 16	ET-96/08/27-1	27-Aug-96	Upper Basin Liner	UBL Exploration Trench
96- 17	ET-96/08/27-2	27-Aug-96	Upper Basin Liner	UBL Exploration Trench
96- 18	ET-96/08/27-3	27-Aug-96	Upper Basin Liner	UBL Exploration Trench
96- 19	ET-96/08/28-1	28-Aug-96	Upper Basin Liner	UBL Exploration Trench
96- 20	ET-96/08/28-2		Upper Basin Liner	UBL Exploration Trench
96- 21	ET-96/08/28-3		Upper Basin Liner	UBL Exploration Trench
96- 22	ET-96/08/28-4	28-Aug-96	Upper Basin Liner	UBL Exploration Trench
96- 23	ET-96/08/28-5	28-Aug-96	Upper Basin Liner	UBL Exploration Trench
96- 24	ET-96/08/28-6	28-Aug-96	Upper Basin Liner	UBL Exploration Trench
96- 25	ET-96/08/28-7	28-Aug-96	Upper Basin Liner	UBL Exploration Trench
	E1-00/00/20-/	40-Mug-30	oppor beautifuler	Toor exhibitation traugit





0,001 CLAY Project: 1627 Date: April 15, 1997 Fine Medium 0.01 SILT Coarse 0.1 US STANDARD SIEVE SIZES #16 #30 #40 GRAIN SIZE IN MILLIMETERS MOUNT POLLEY MINING CORPORATION SAND TAILINGS GRADATION LIMITS Medium -O-Bulk Tails (calculated from Slime and Sand Tails Components) Coarse -A-Slime Tails Median -D-Sand Tails Median # 3/8" 10 3/4" GRAVEL SIEVE OPENING IN INCHES 6" 3" 1 1/2" 1" Coarse AANewbulls xls PSA Plut All 5/7/97, 3:05 PM 90 COBBLES BOULDERS 1000 2 901 8 8 2 8 20 8 30 2 FIGURE 3.1 PERCENT FINER BY WEIGHT KNIGHT PIESOLD LTD. CONSULTING ENGINEERS



NOTES

Filling schedule based on tailings dry density of 1.1 t/m³ in Year 1, 1.2 t/m³ in Year 2 and 1.3 t/m³ for remainder. All construction periods shown are for July through September, except Stage lb, which is February through March.

Embankment crest elevations for each of the staged expansions will be determined annually based on tailings production, in—situ density and water management requirements.

MOUNT POLLEY PROJECT
TAILINGS AREA FILLING SCHEDULE AND STAGED CONSTRUCTION

FOR 6,500,000 tpy THROUGHPUT

MOUNT POLLEY MINING CORPORATION

GRAIN SIZE IN MILLIMETERS

0.1

0.01

0.001

20

10

1000

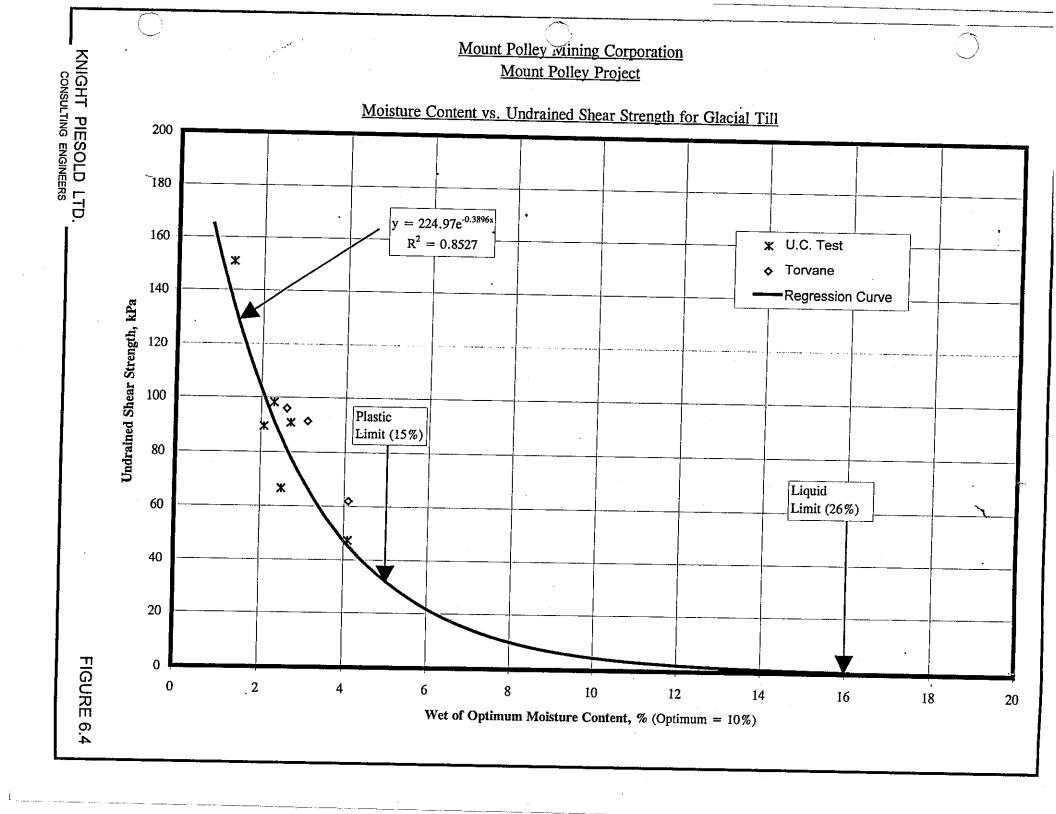
Note:

* = These are 100% limits.

100

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FIGURE 6.3



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20 -

10 -

1000

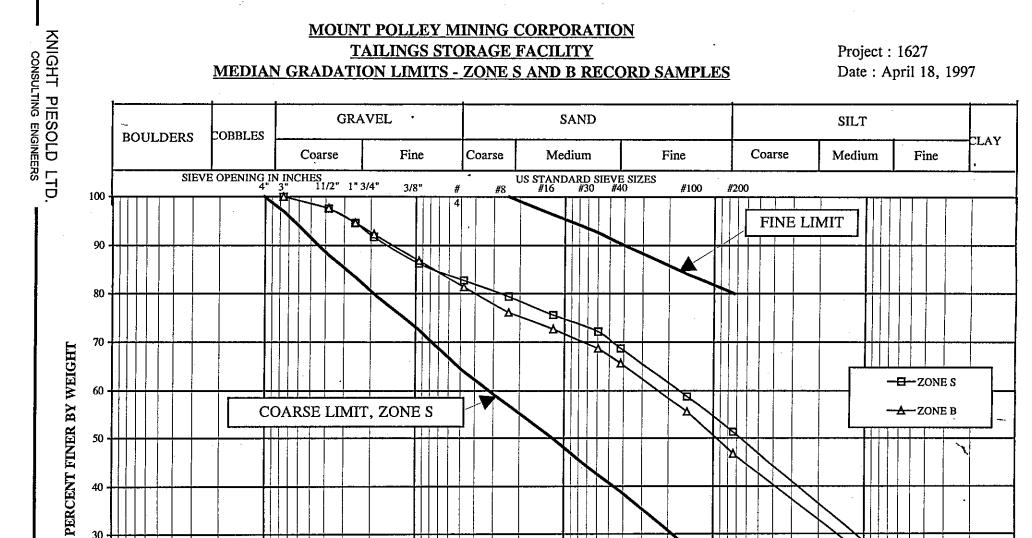
Note:

* = These are 100% limits.

100

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FIGURE 6.5

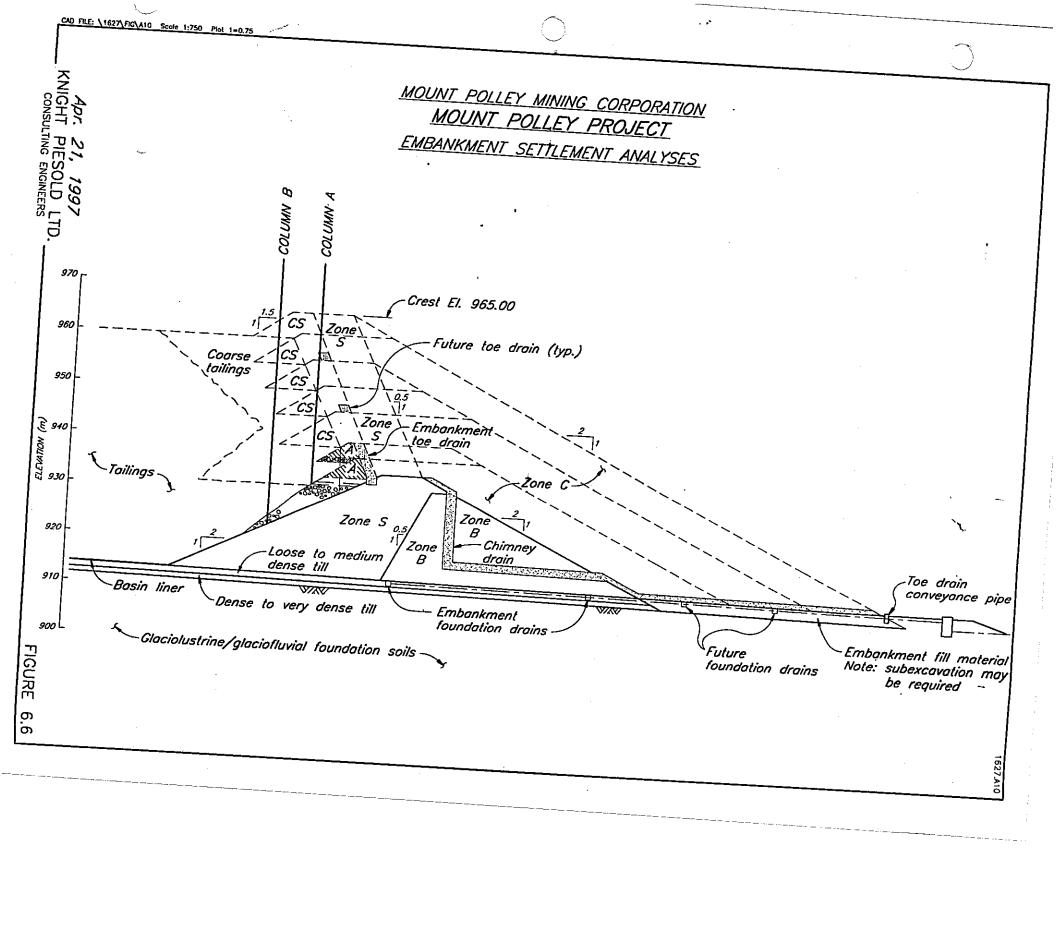


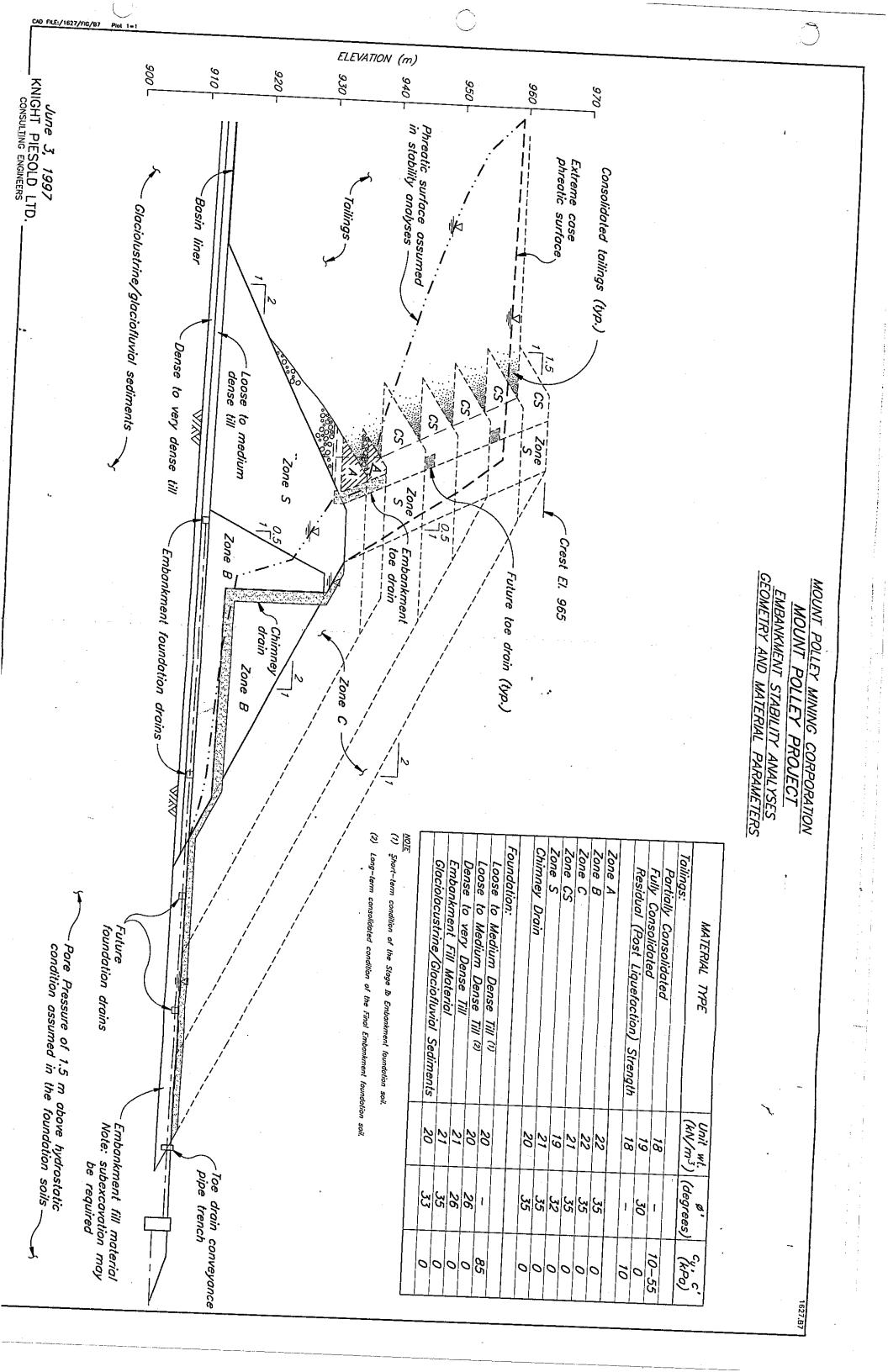
GRAIN SIZE IN MILLIMETERS

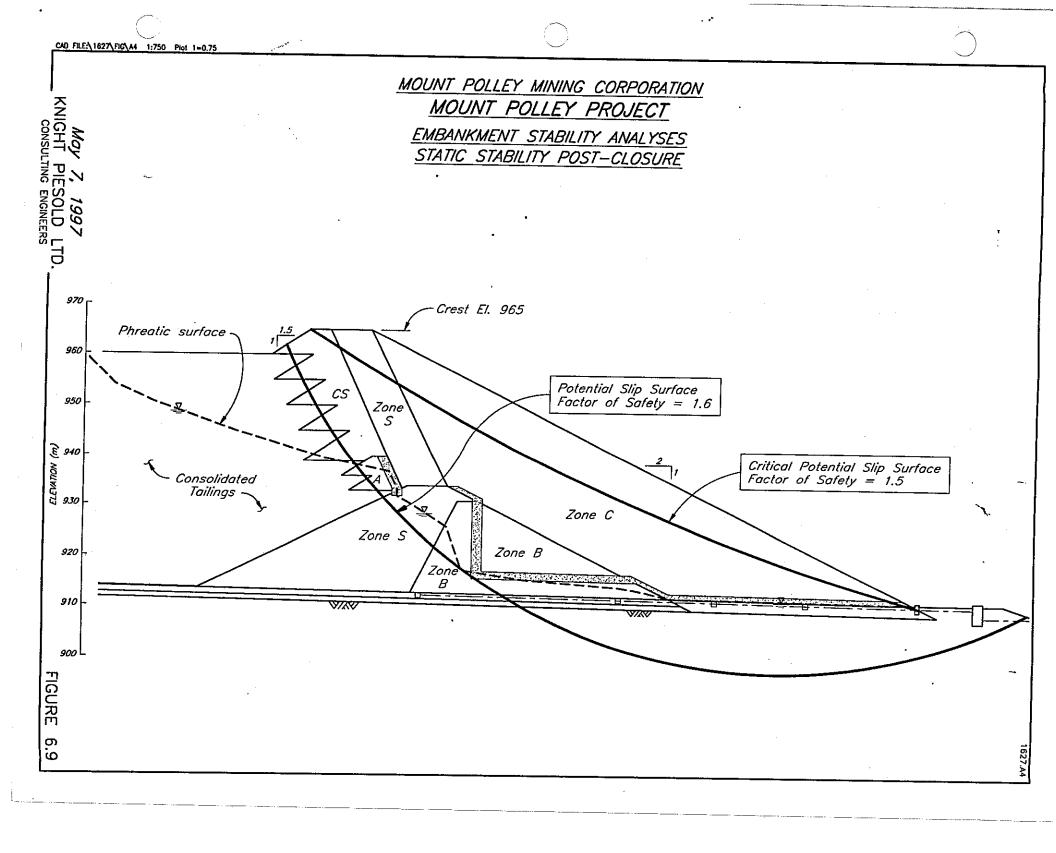
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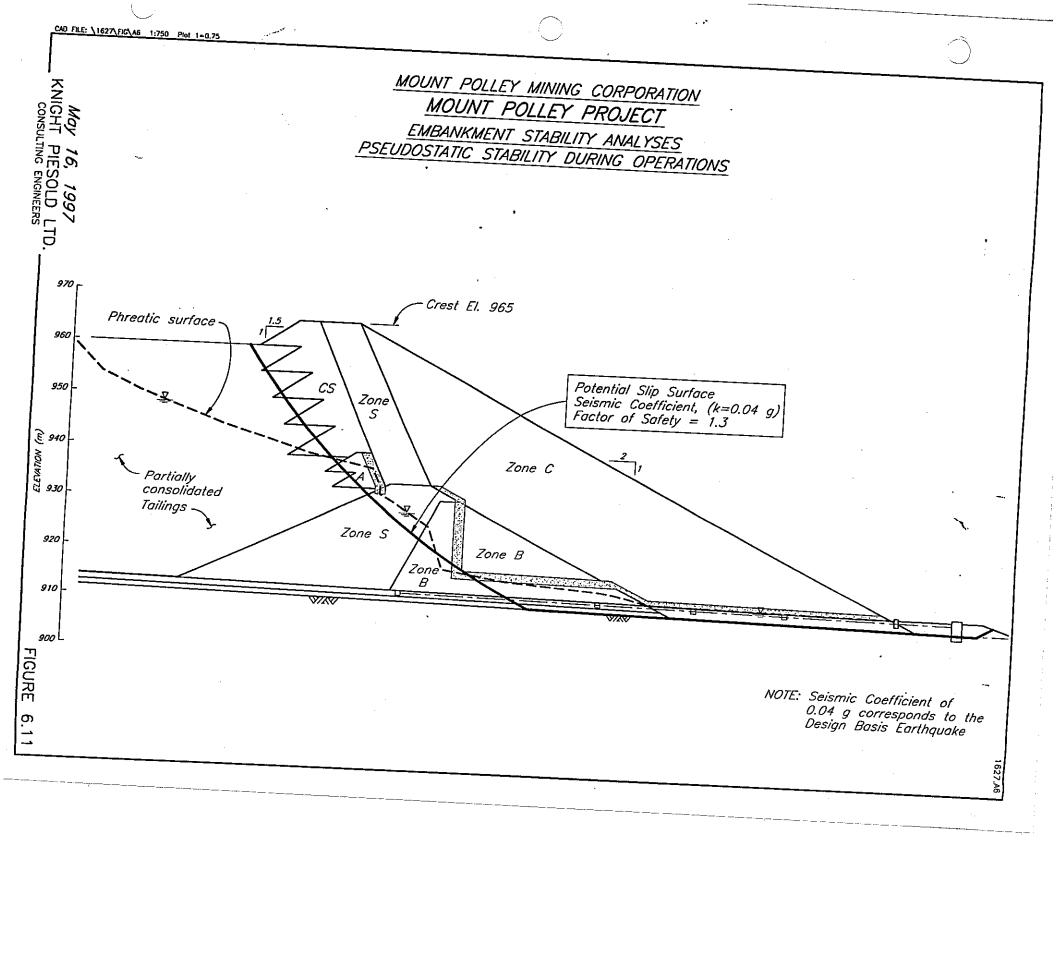
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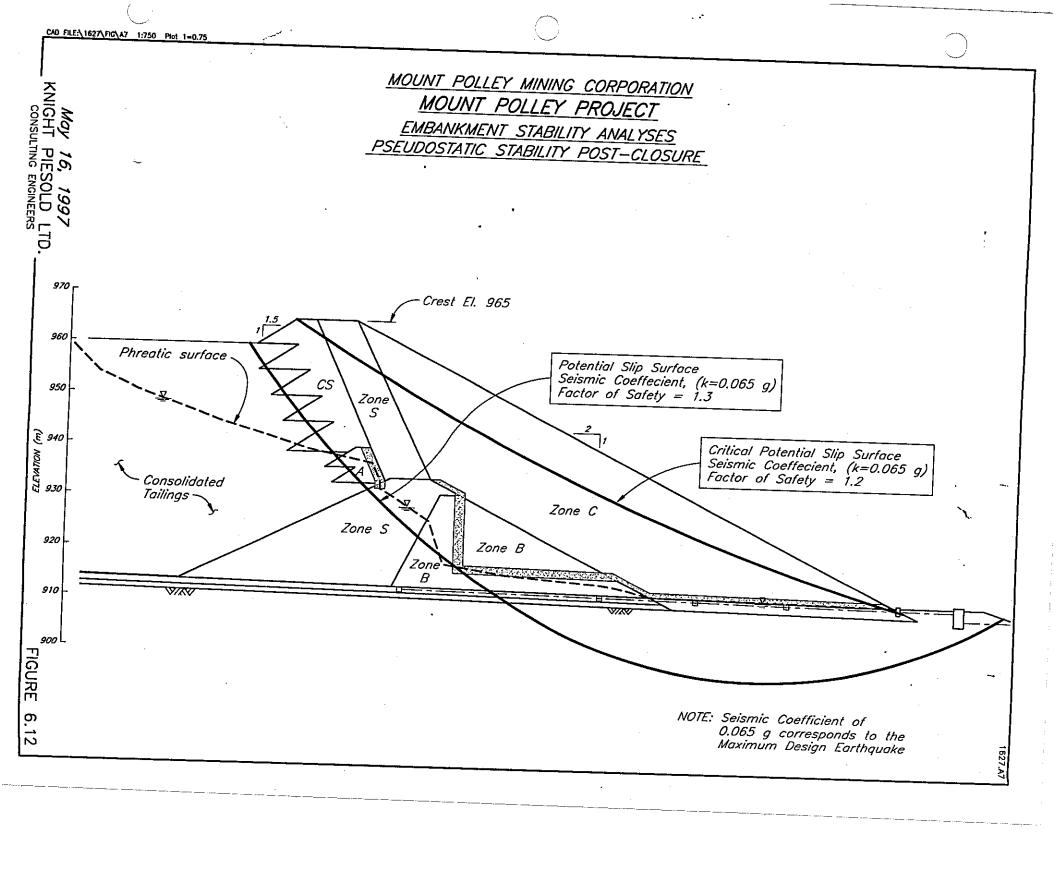
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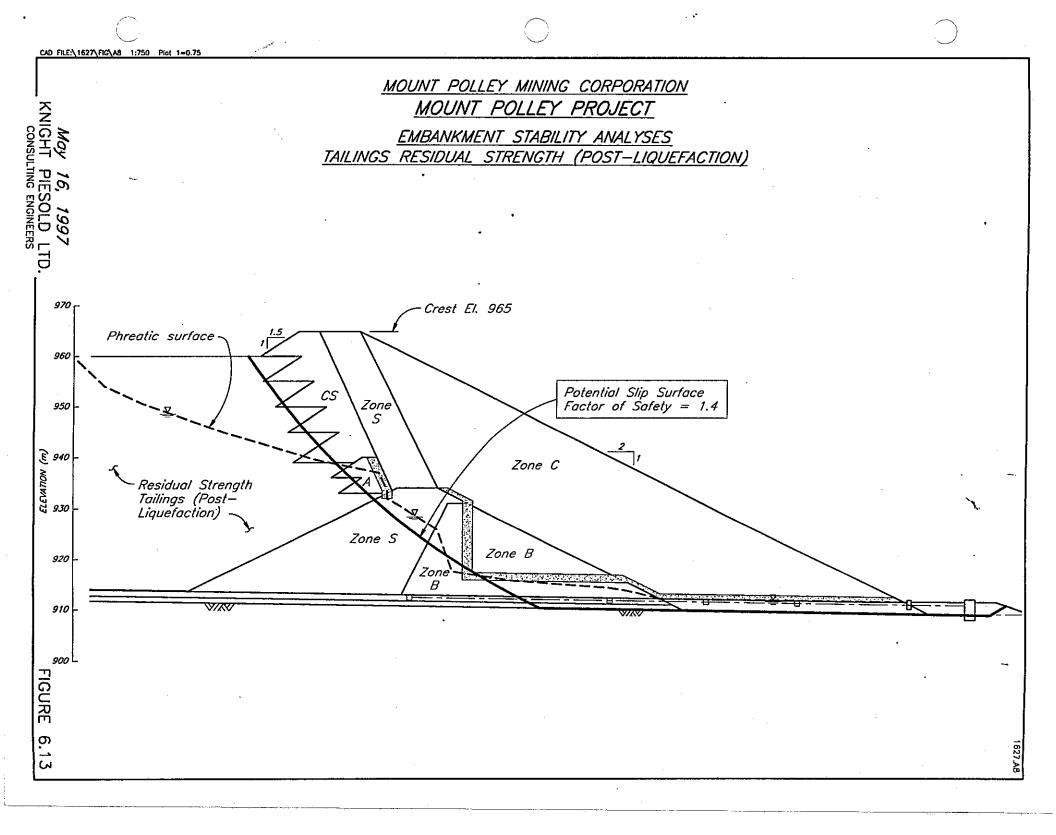












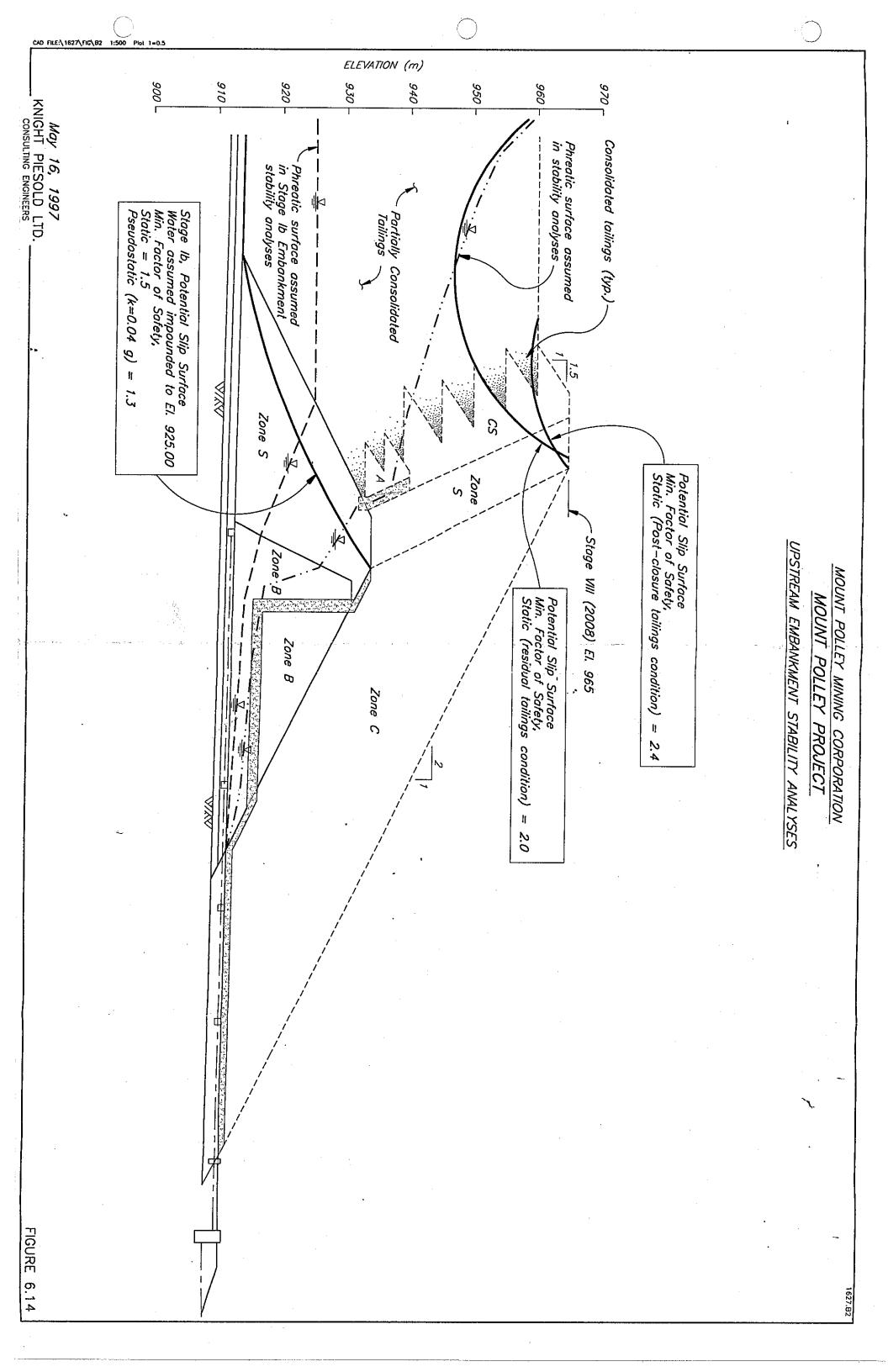
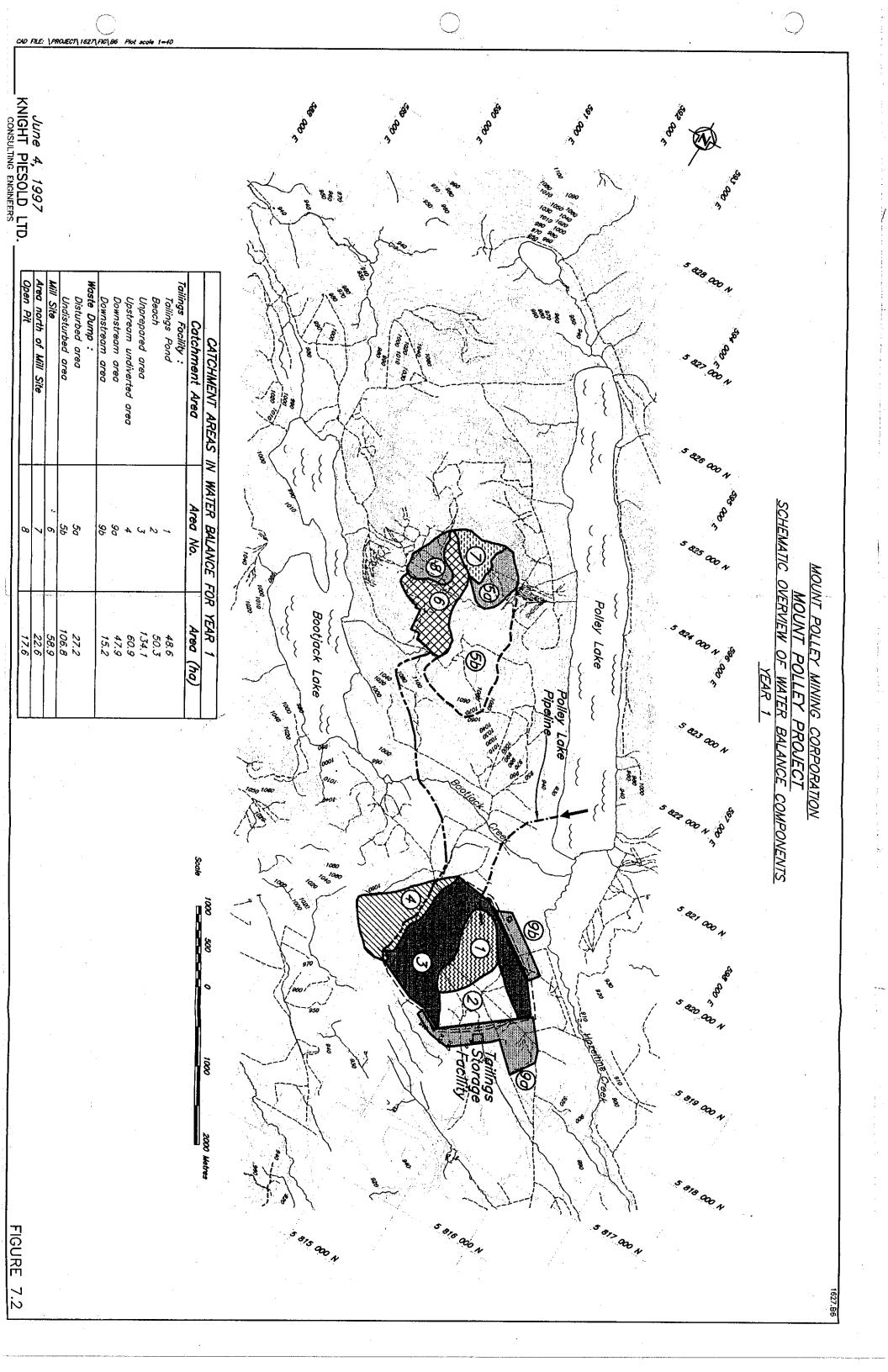
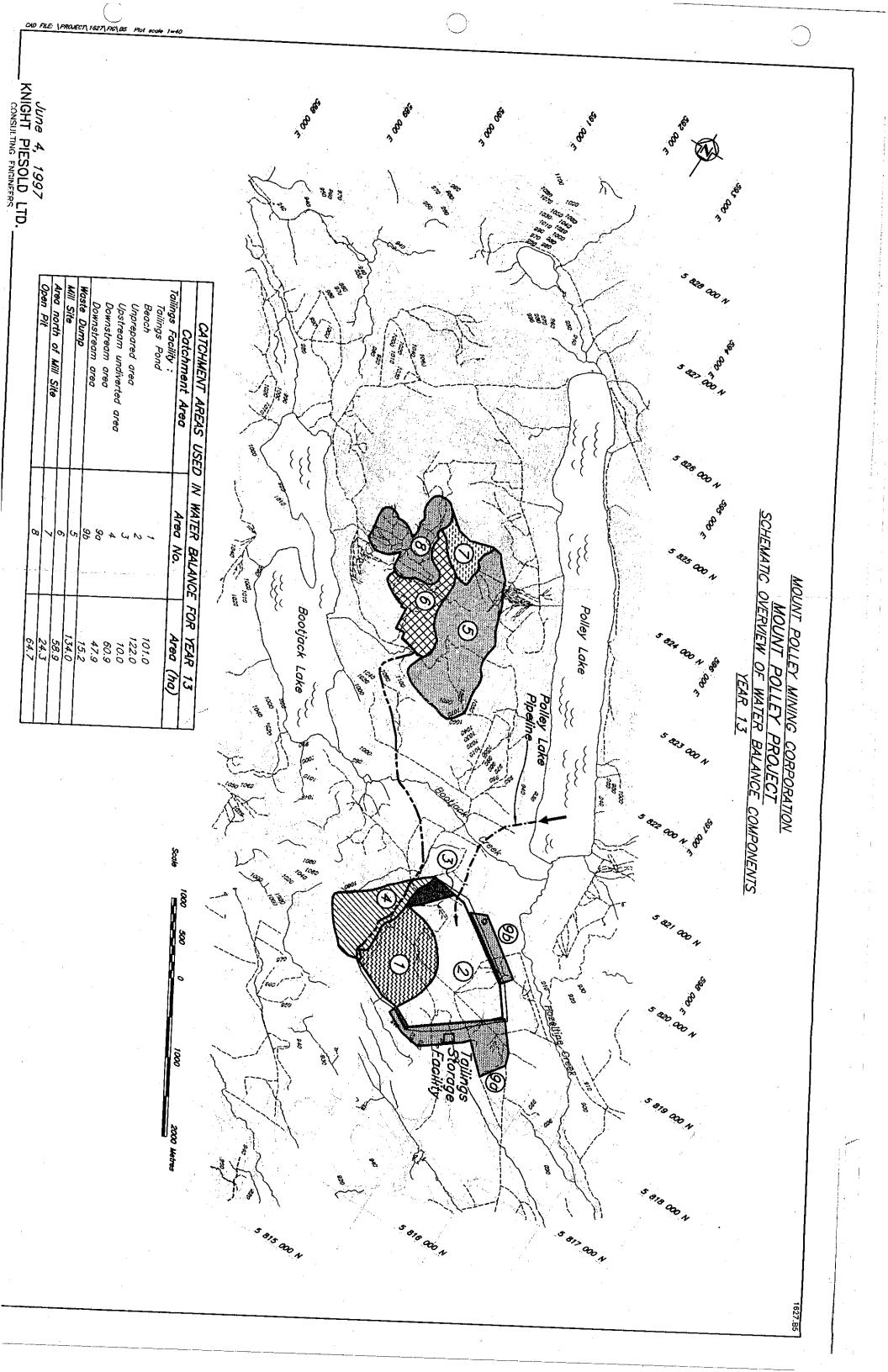


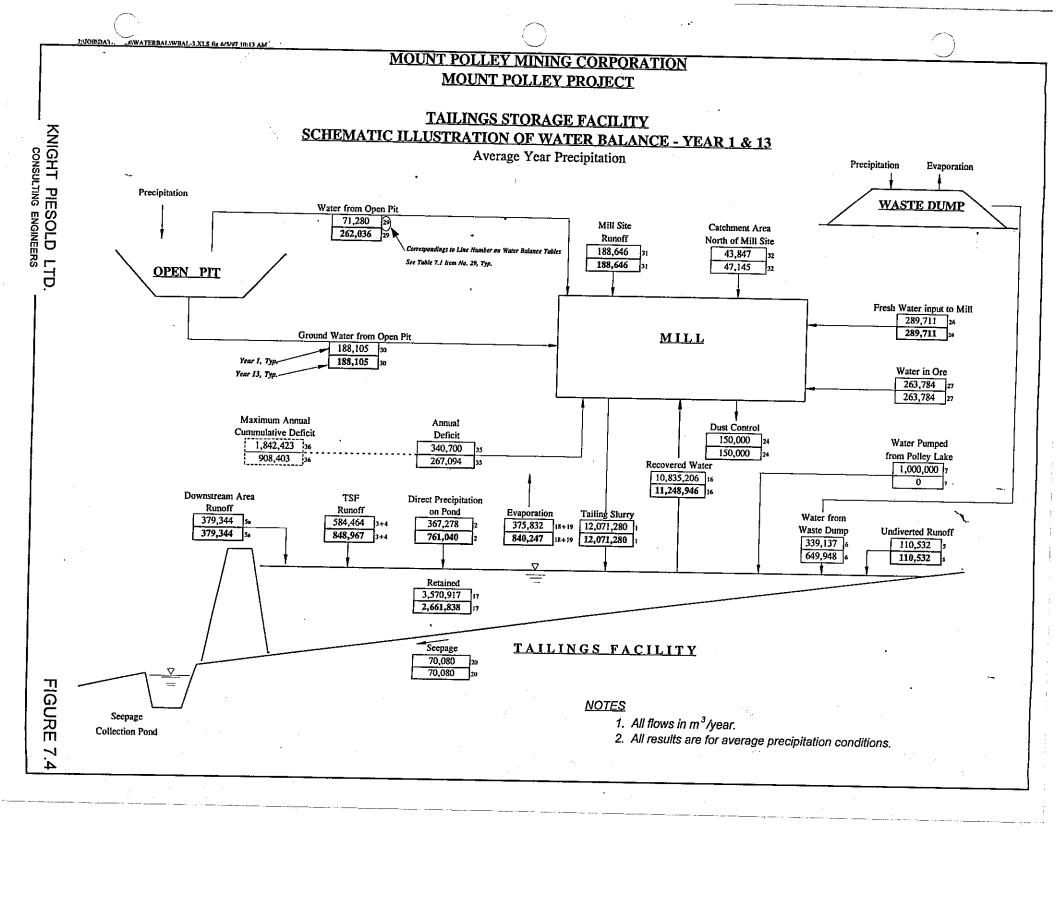
FIGURE 6.15

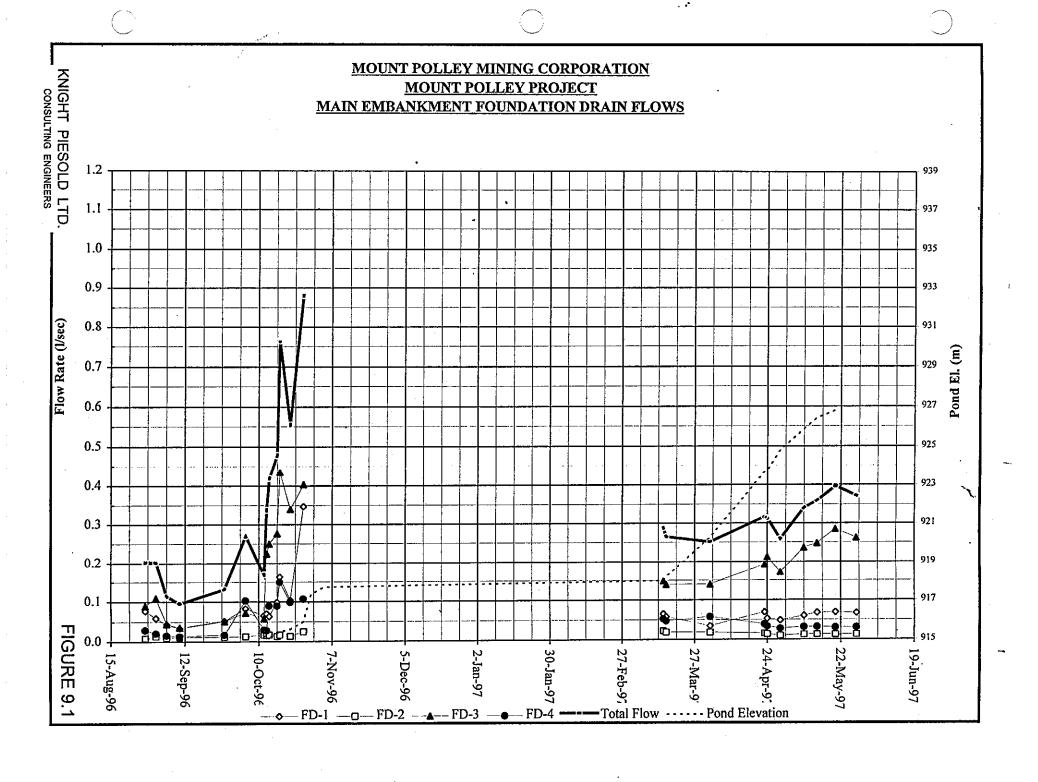
MOUNT POLLEY MINING CORPORATION

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18-Dec

20-Nov

23-Oct

25-Sep

28-Aug

15-Jan

12-Feb

12-Mar

9-Apr

7-May

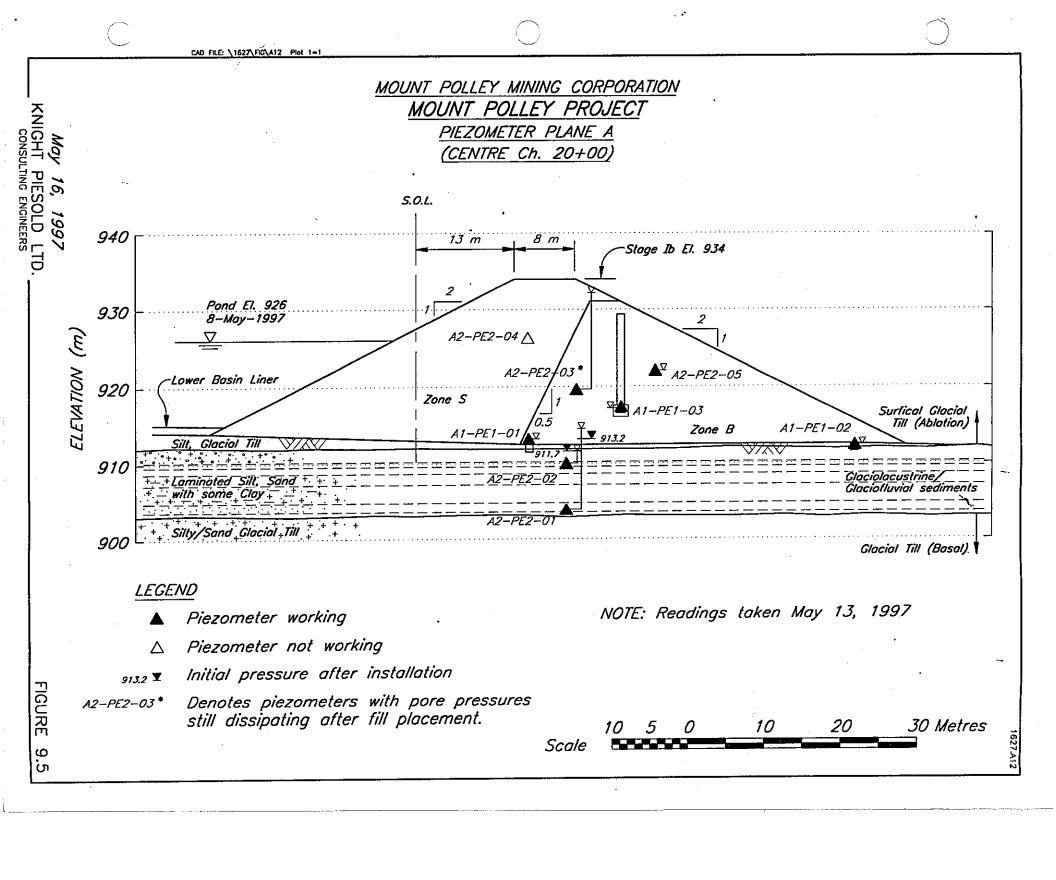
4-Jun

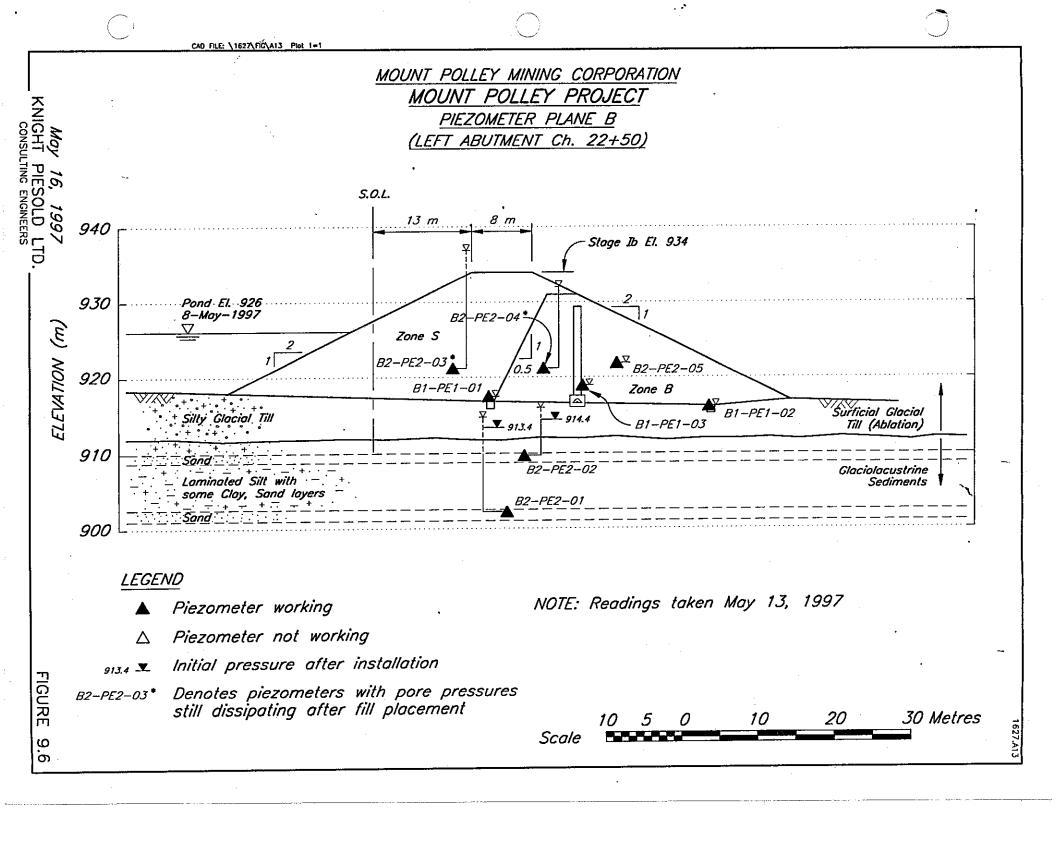
FIGURE 9.2

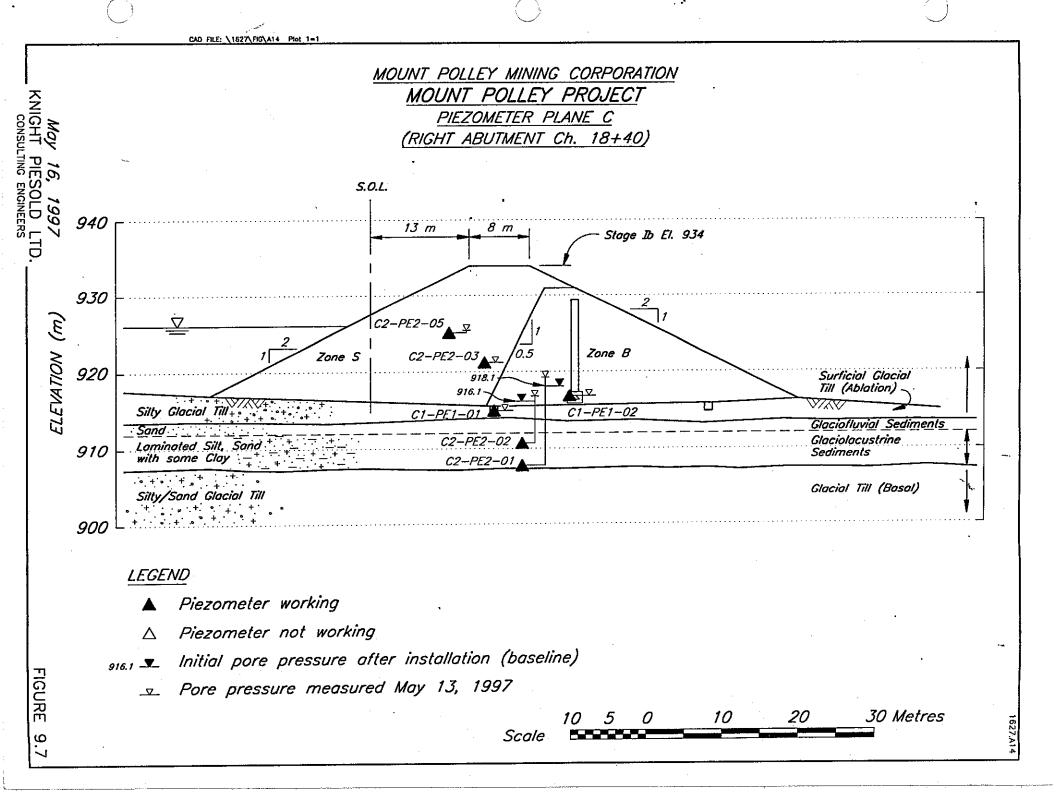
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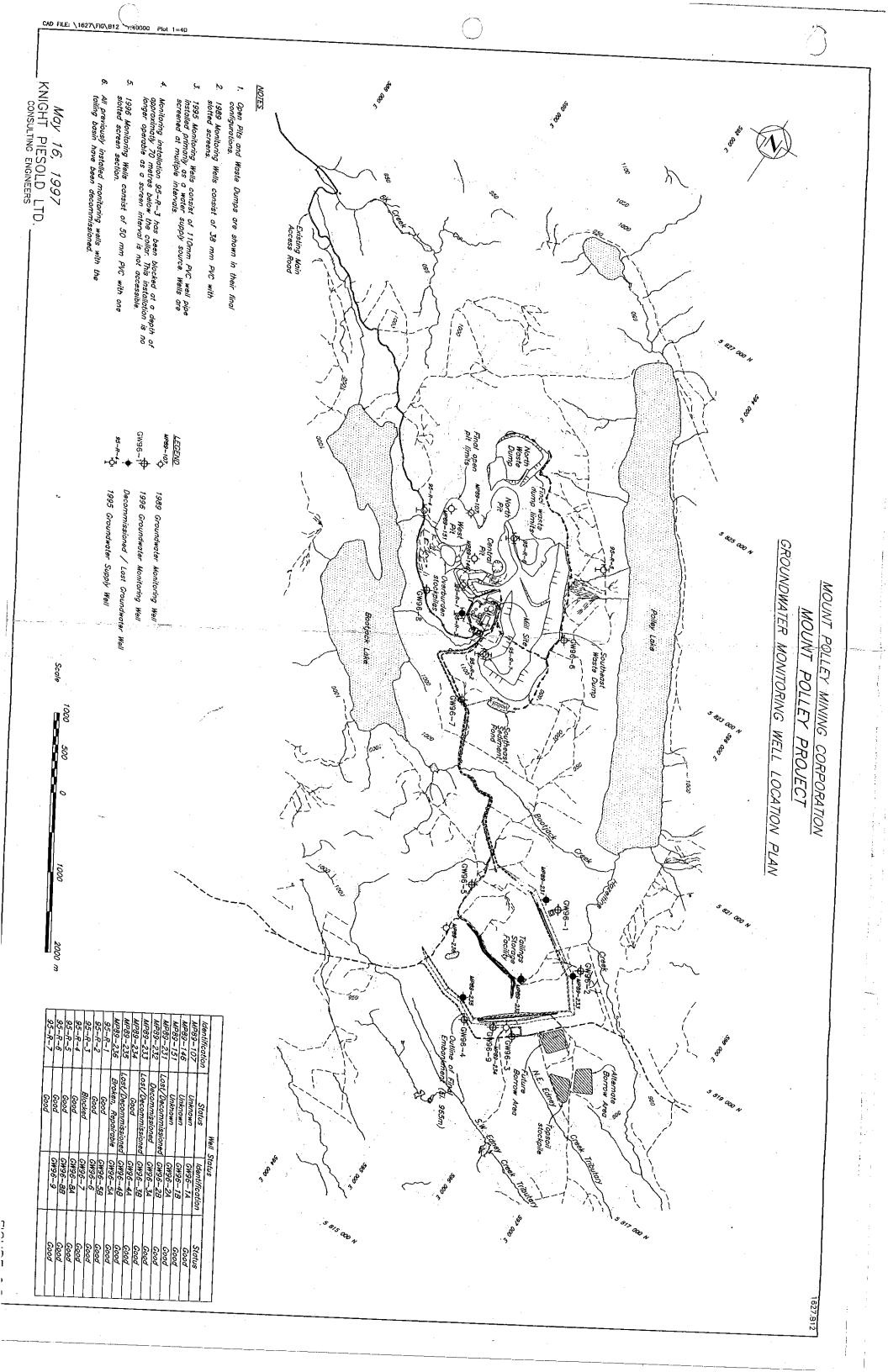
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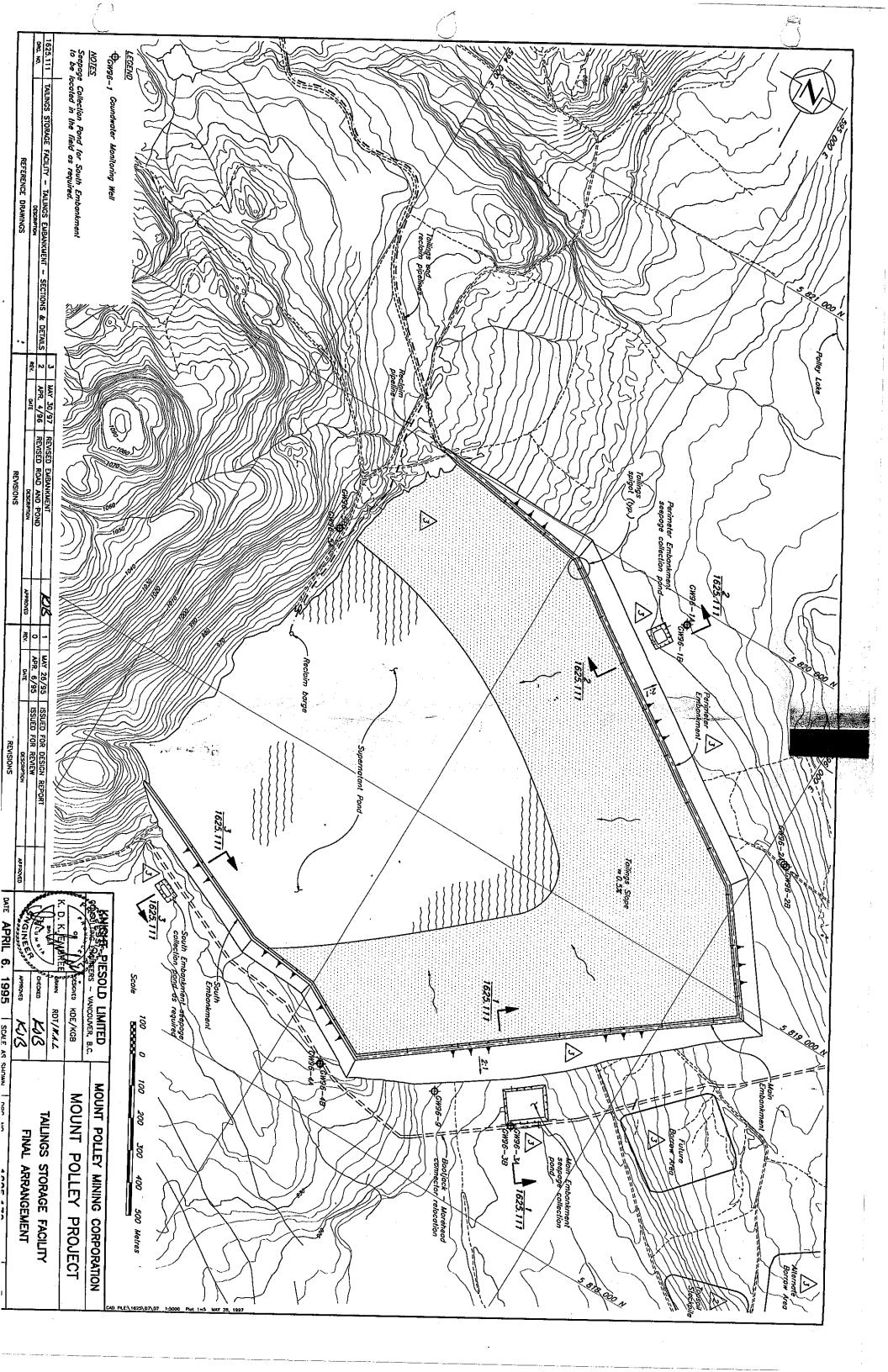
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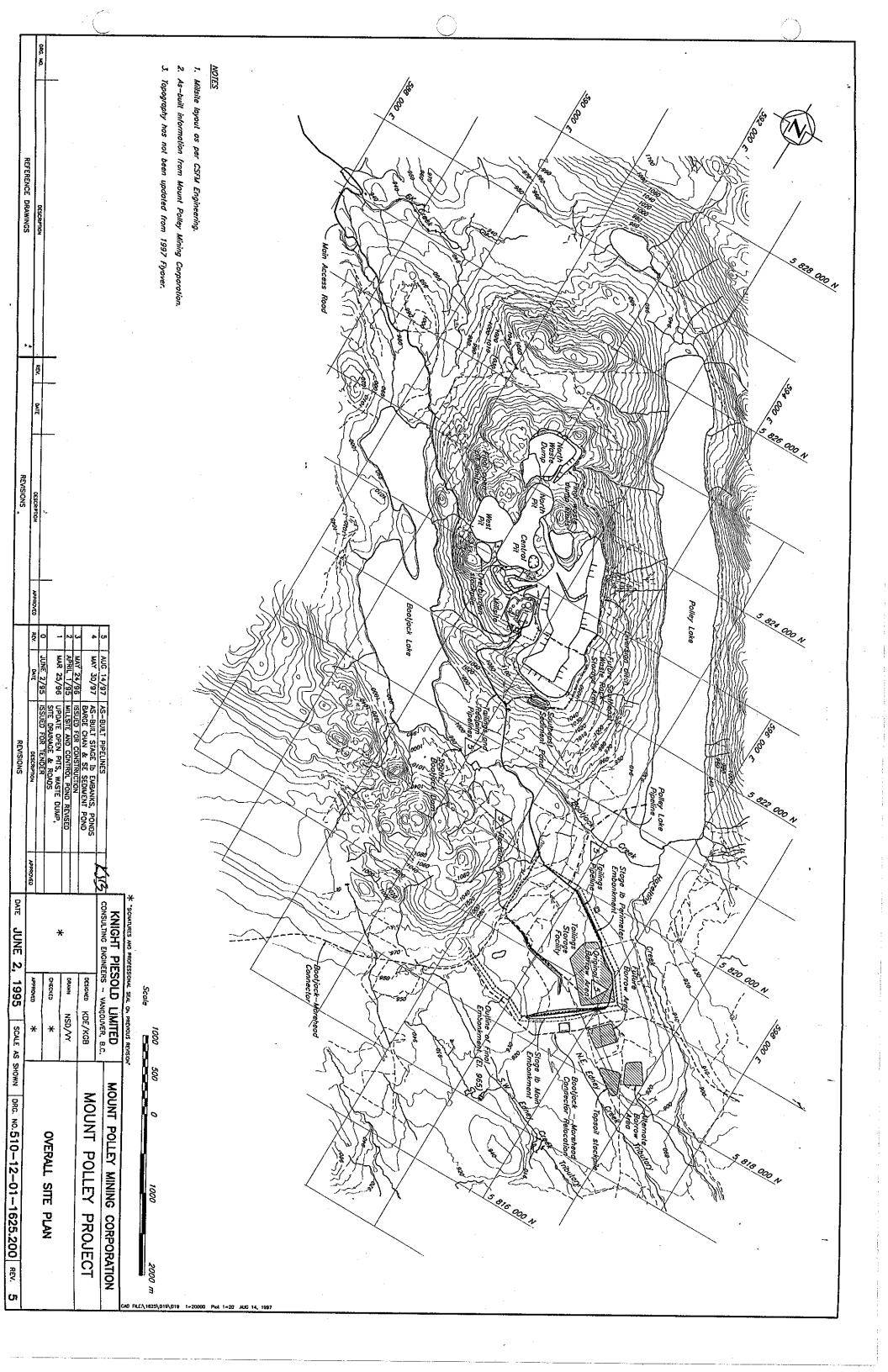


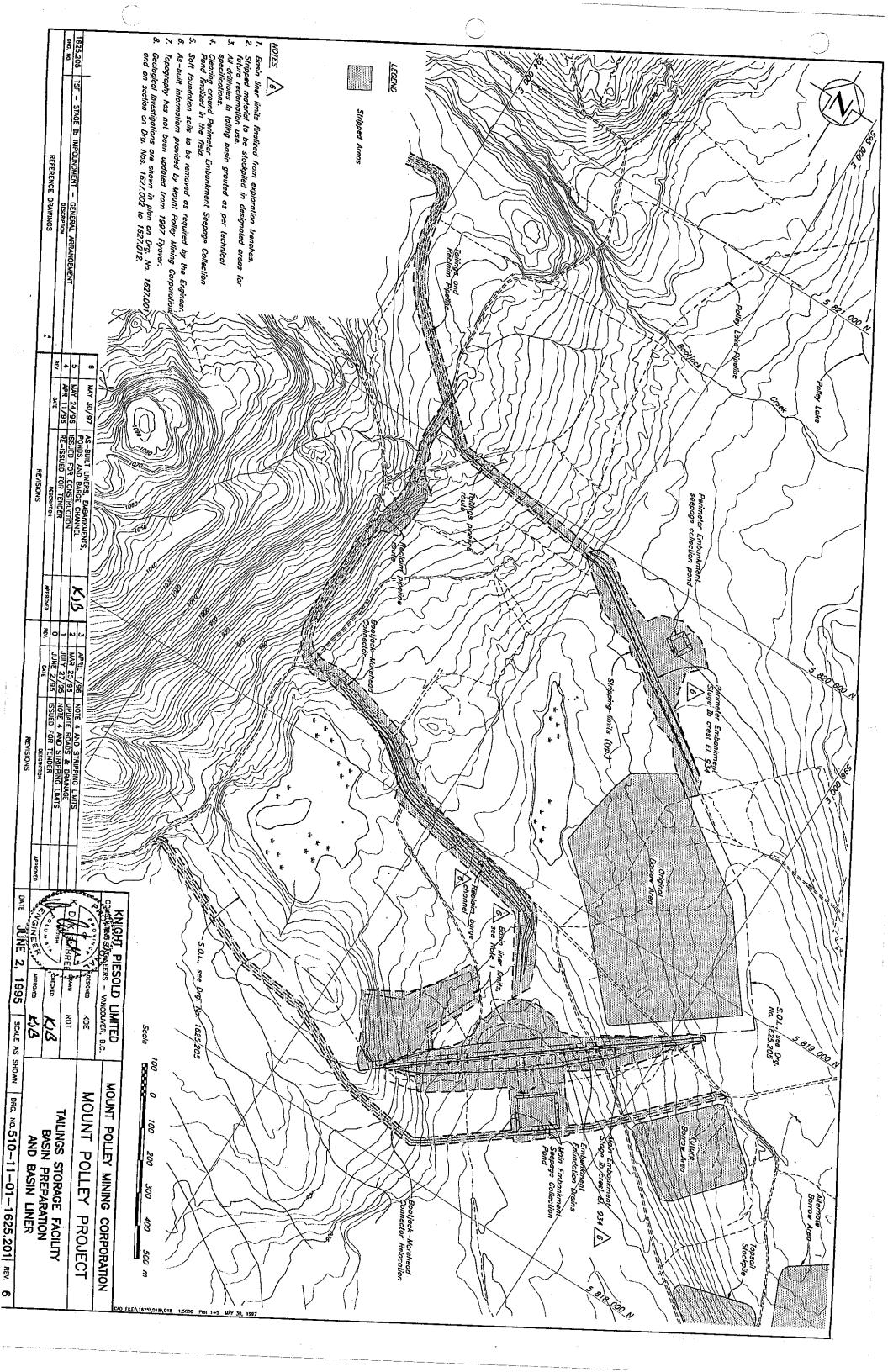


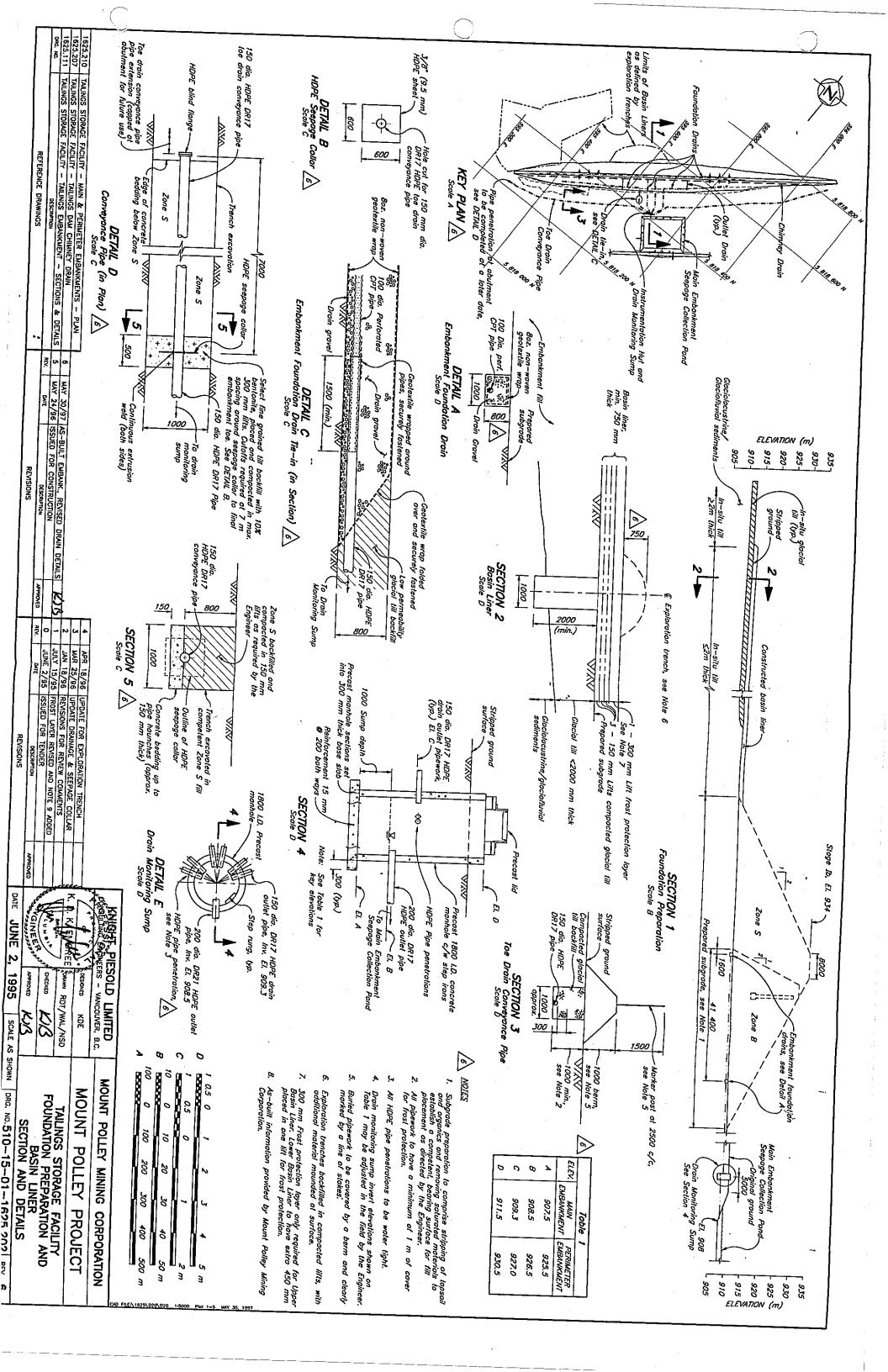


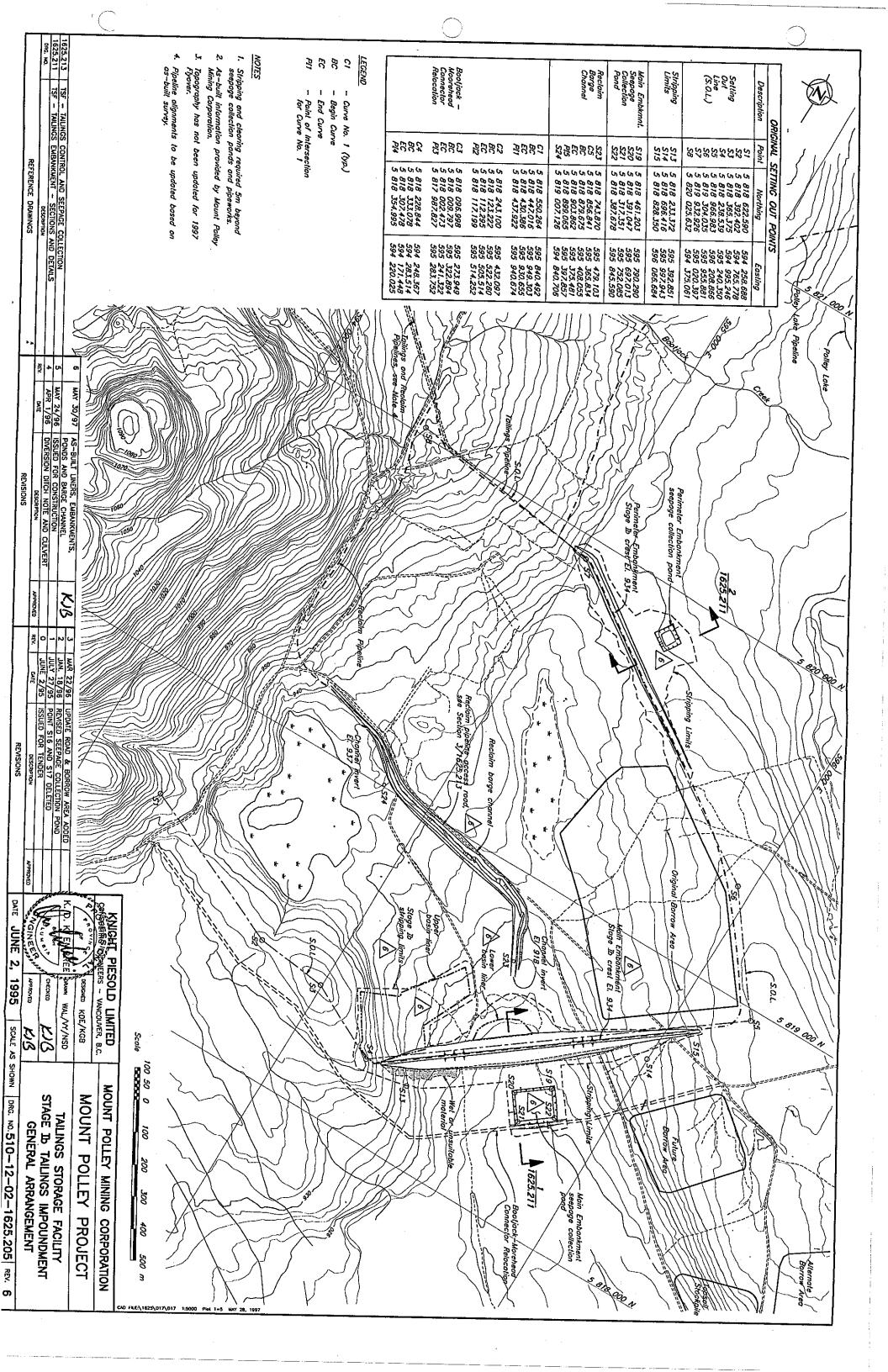


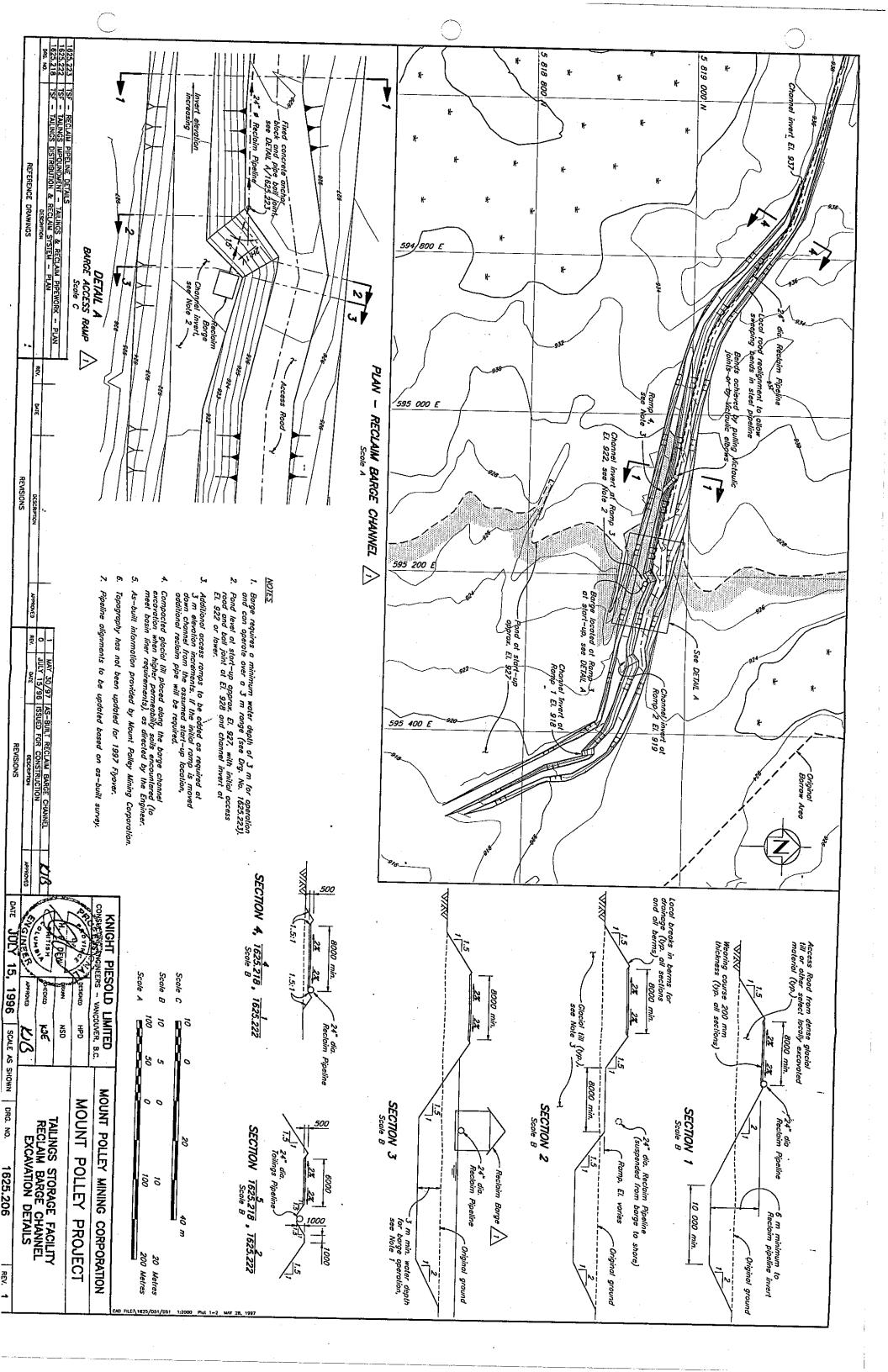


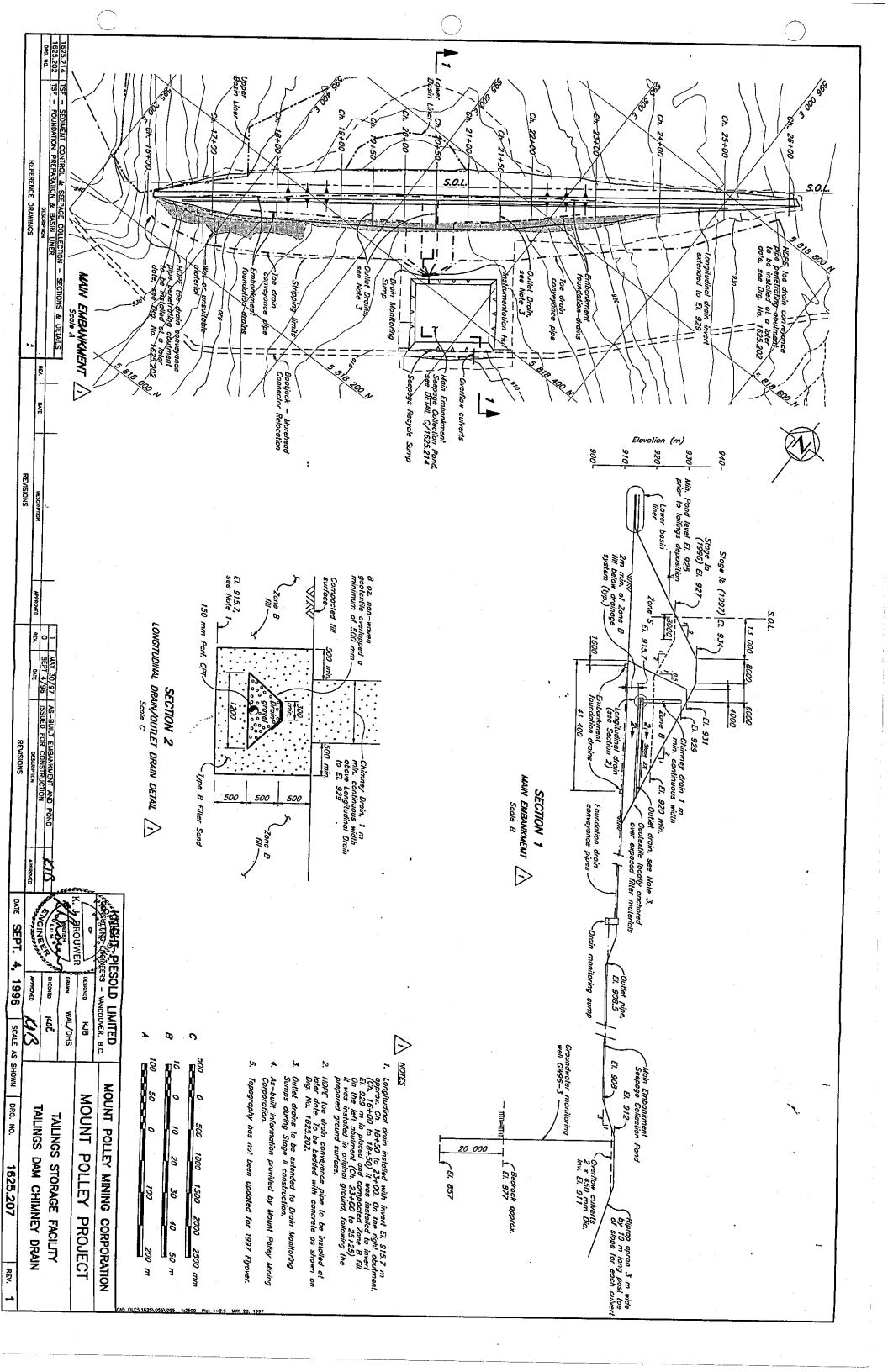


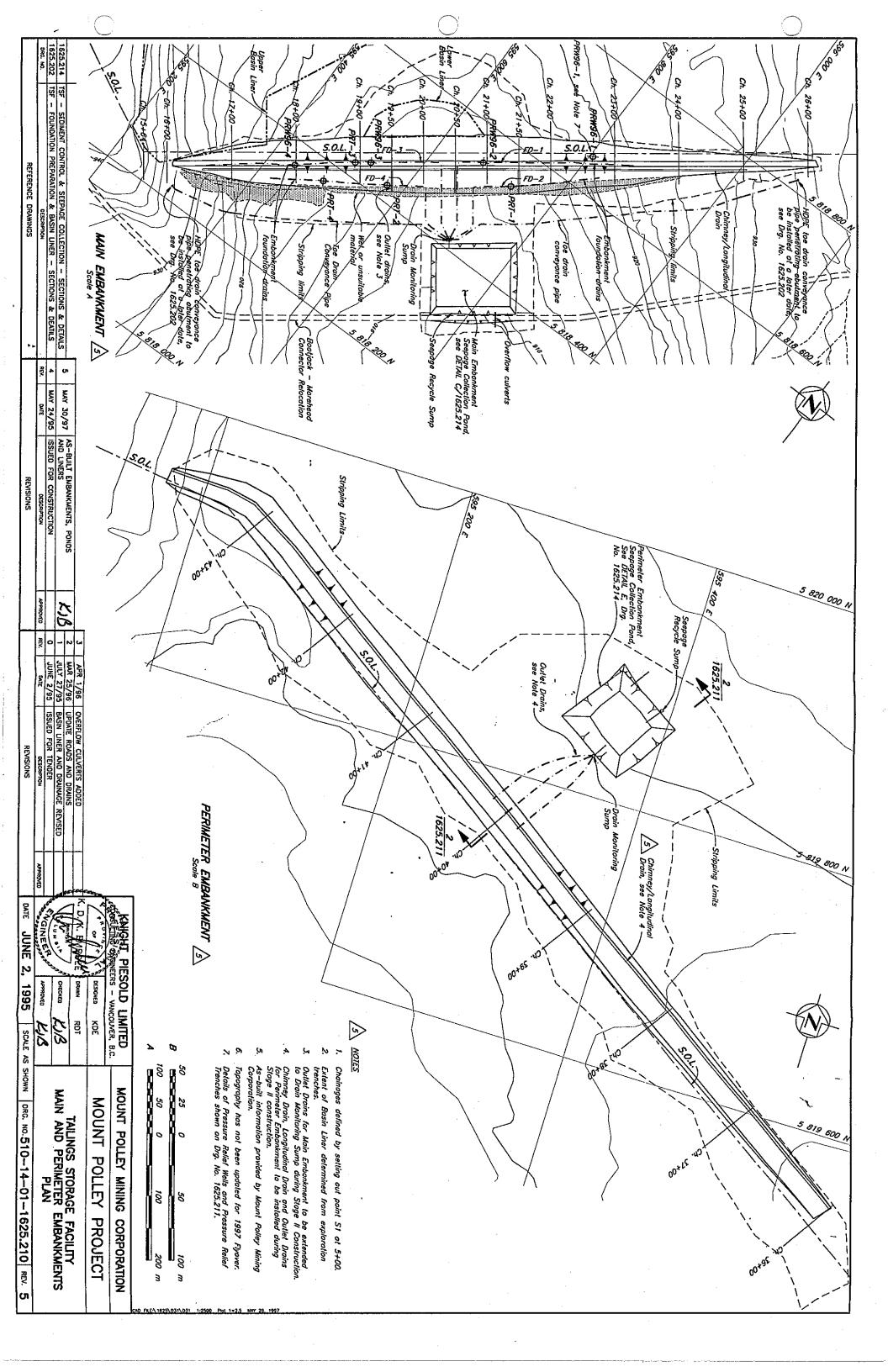


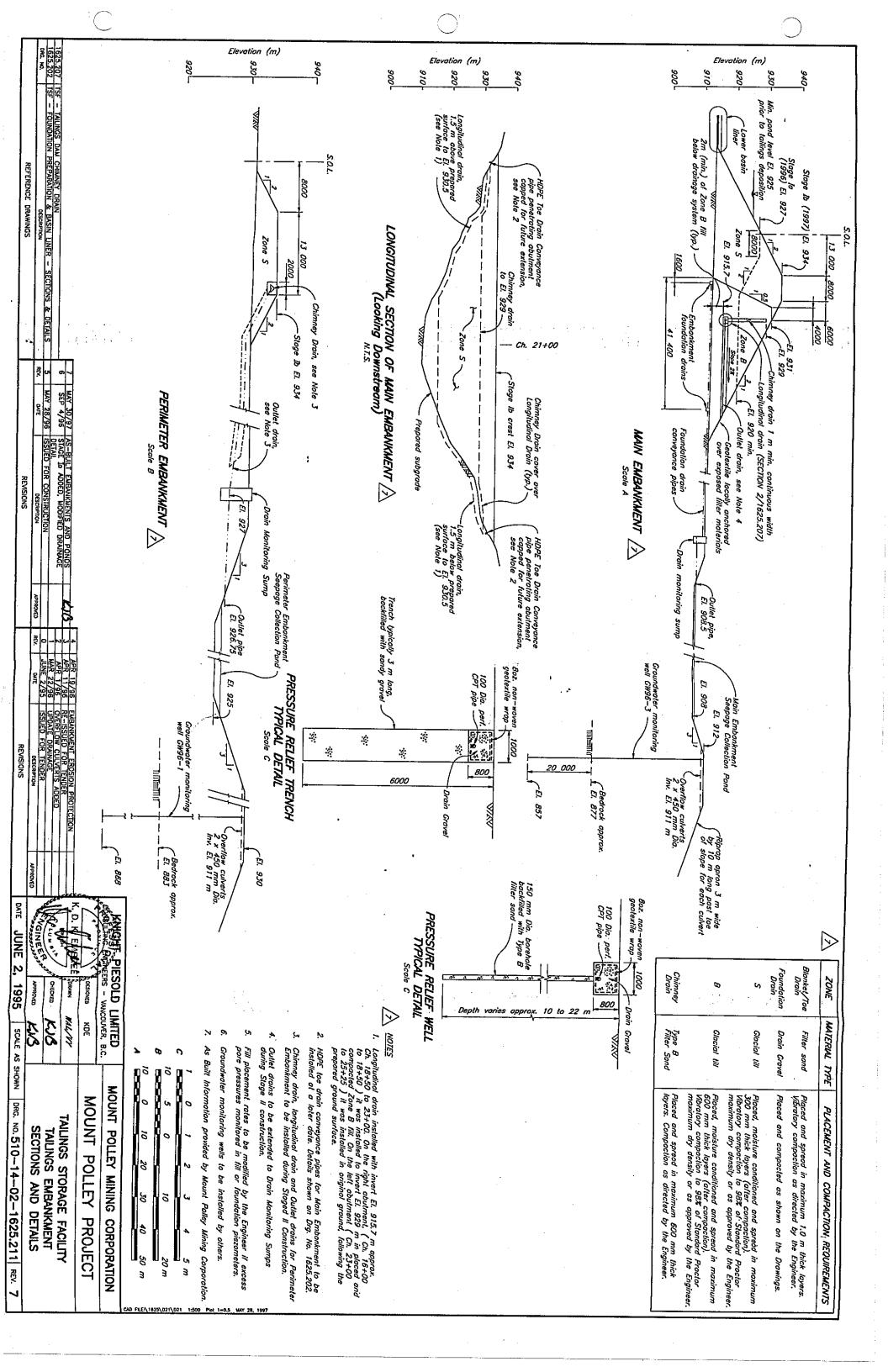


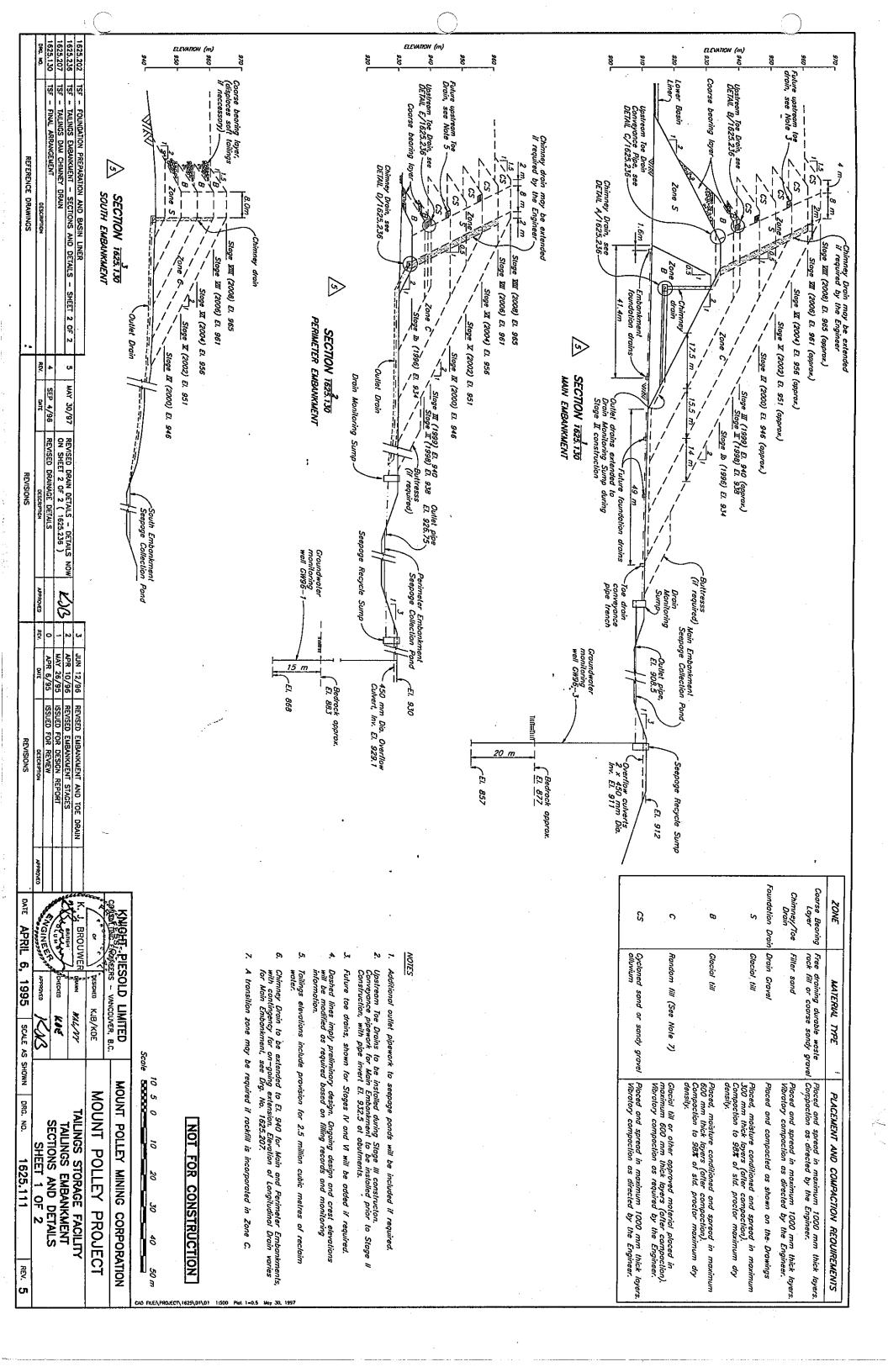


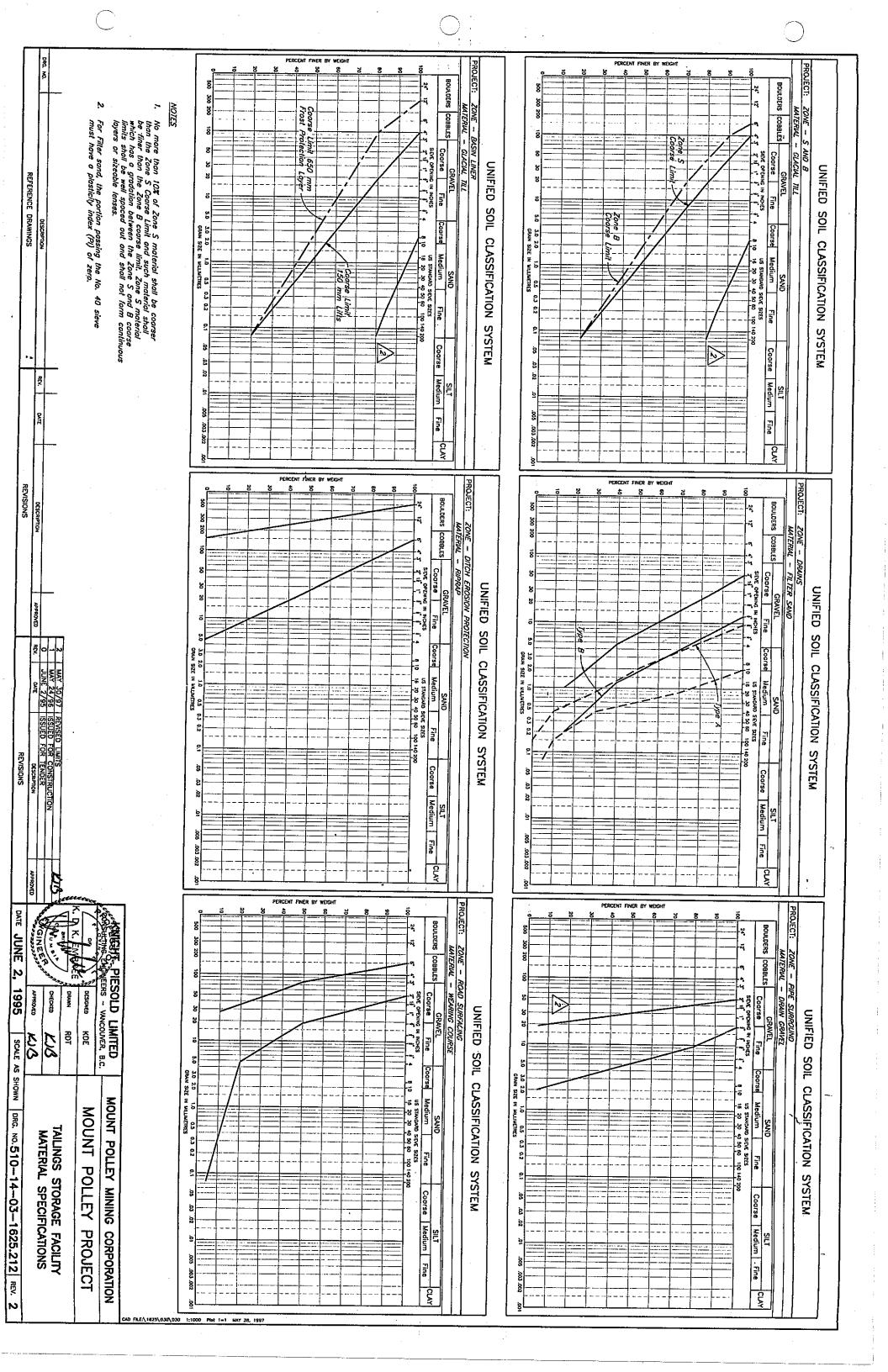


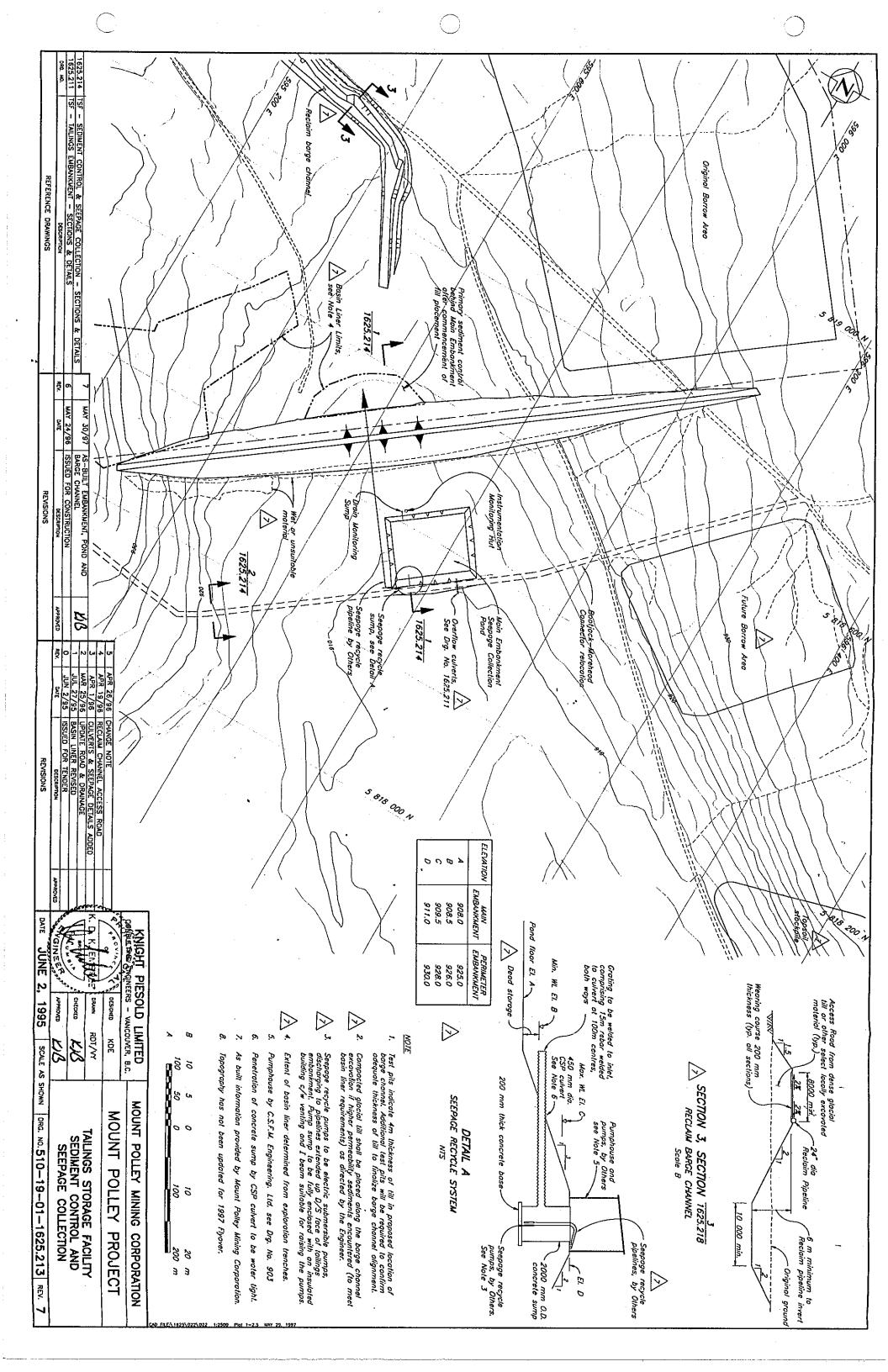


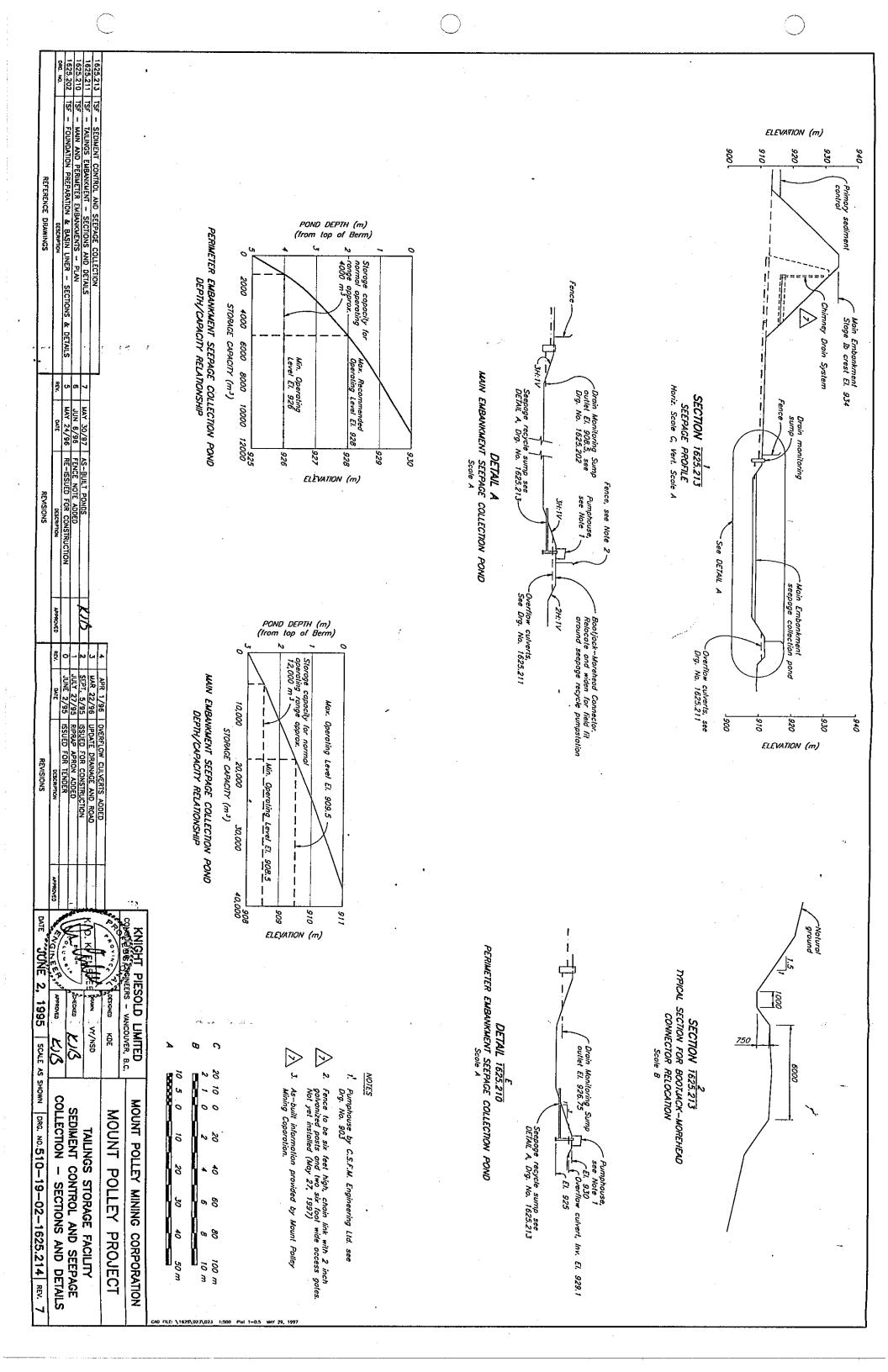


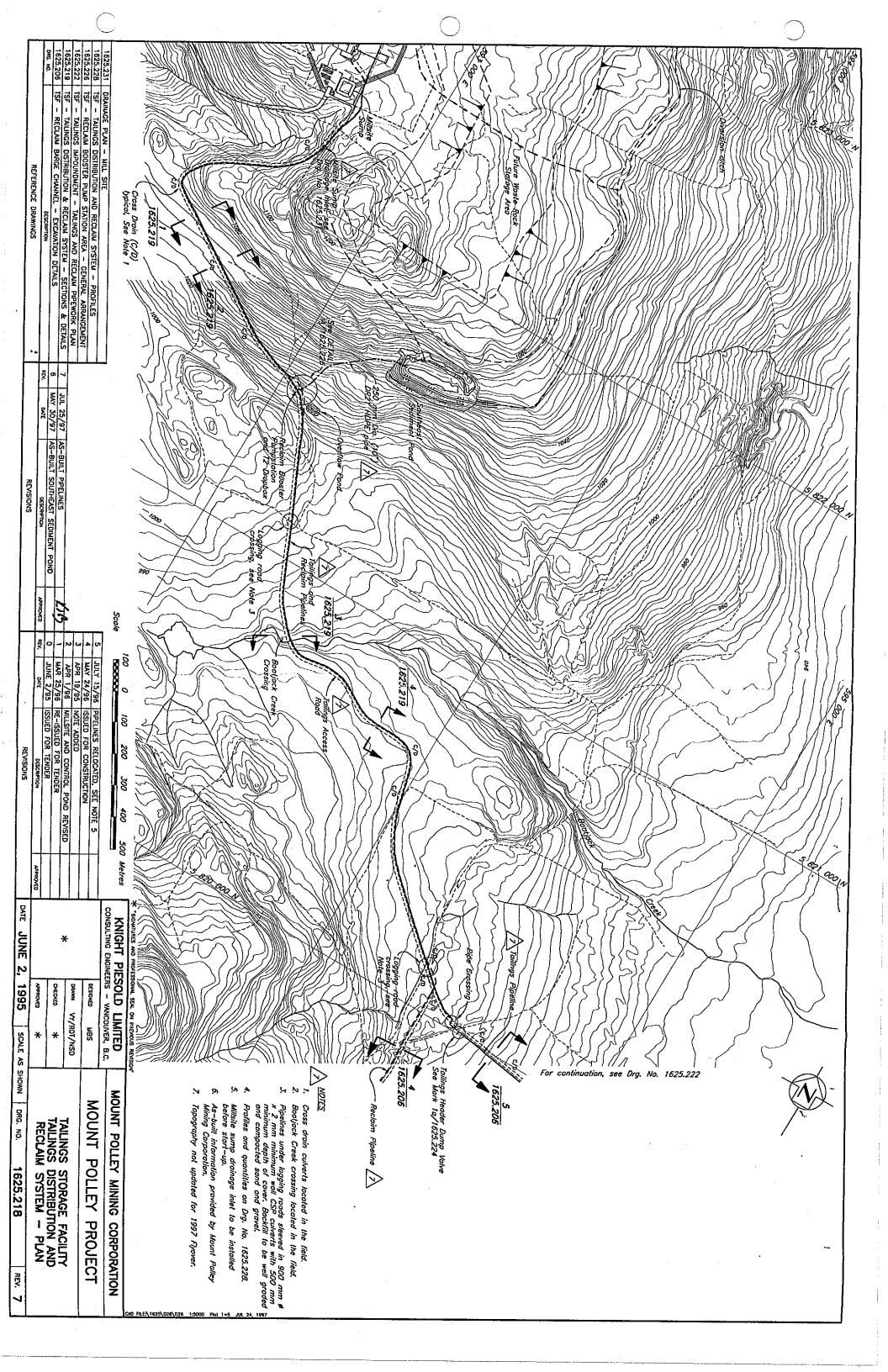


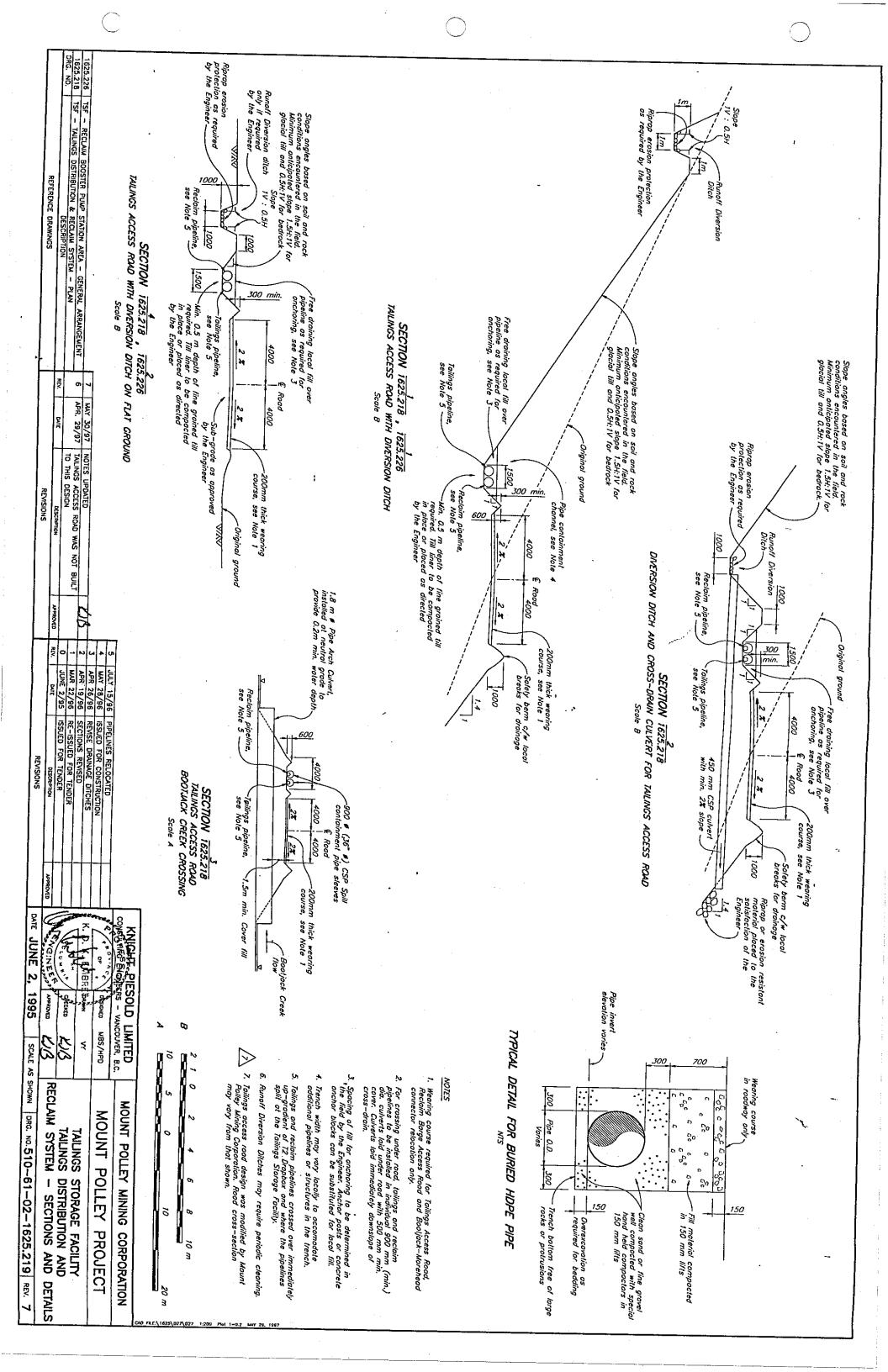


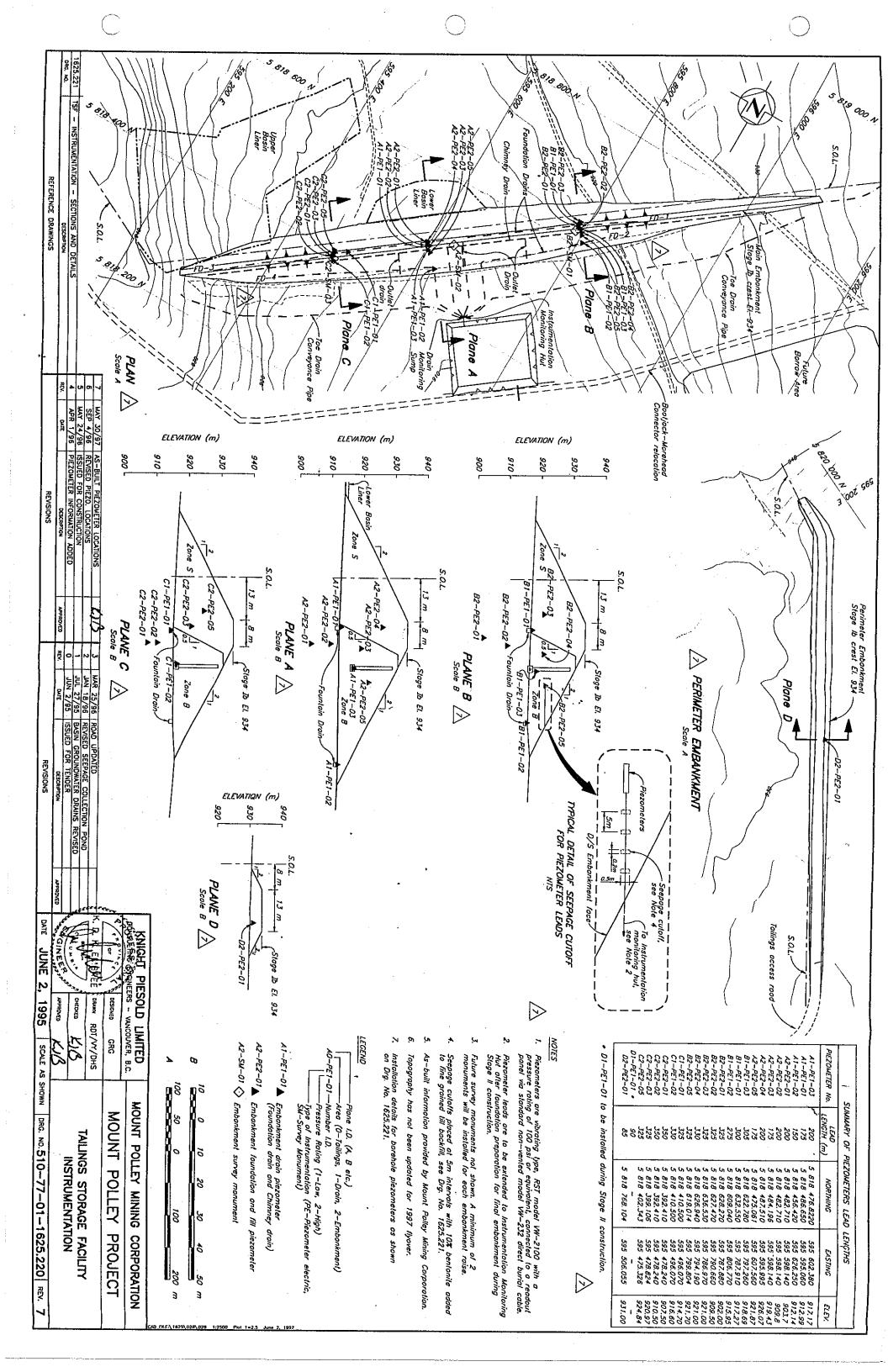


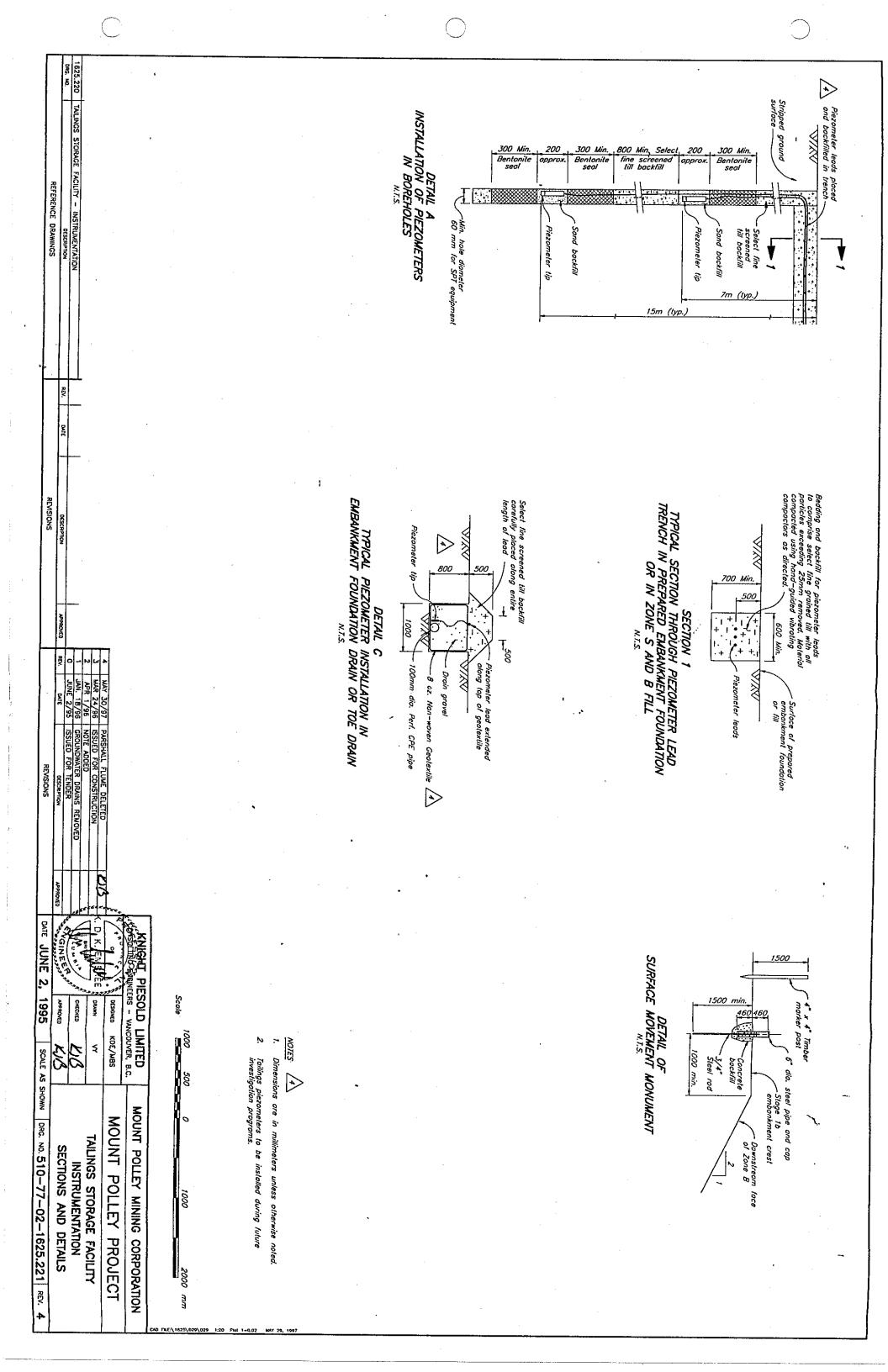


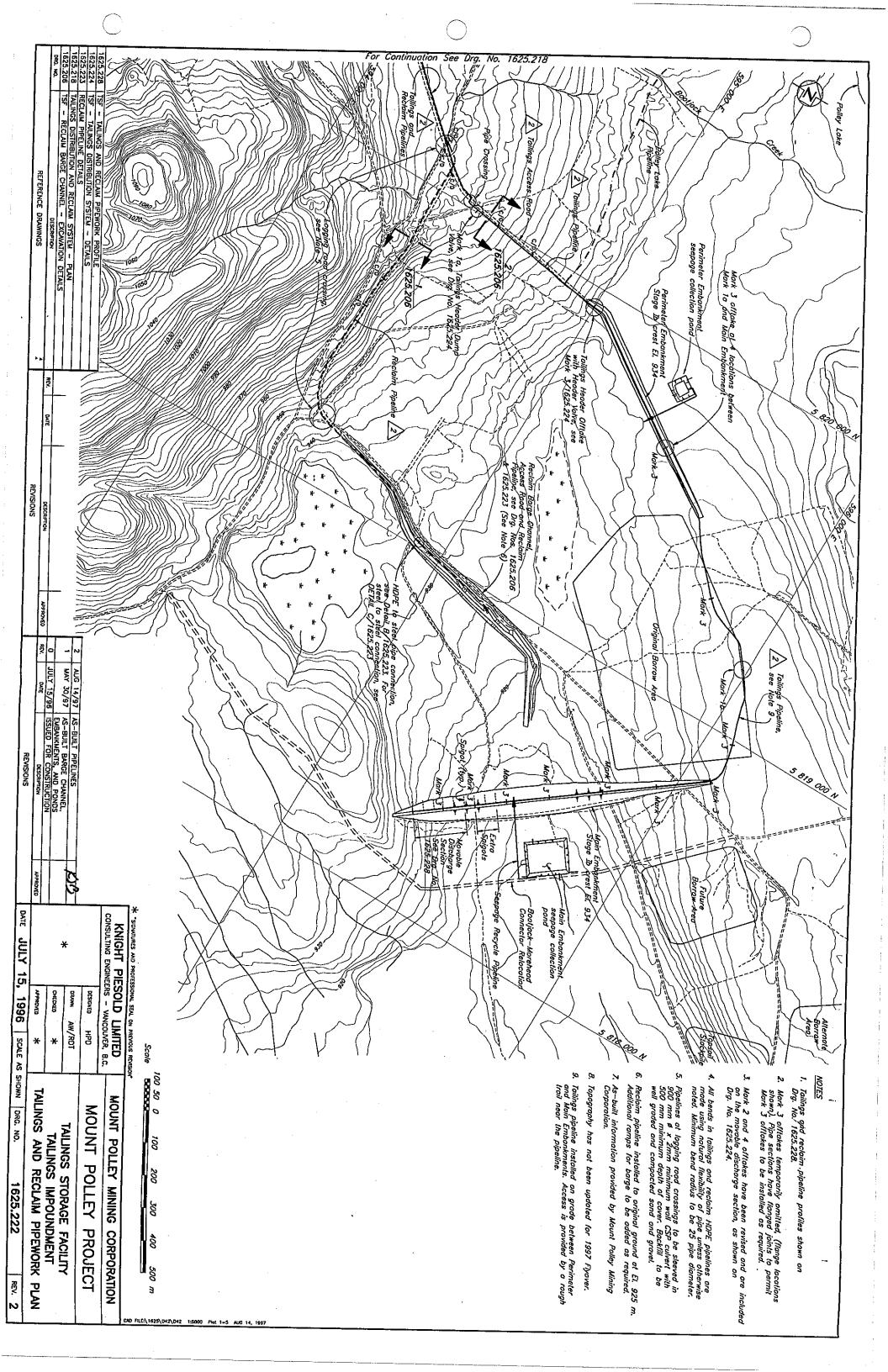


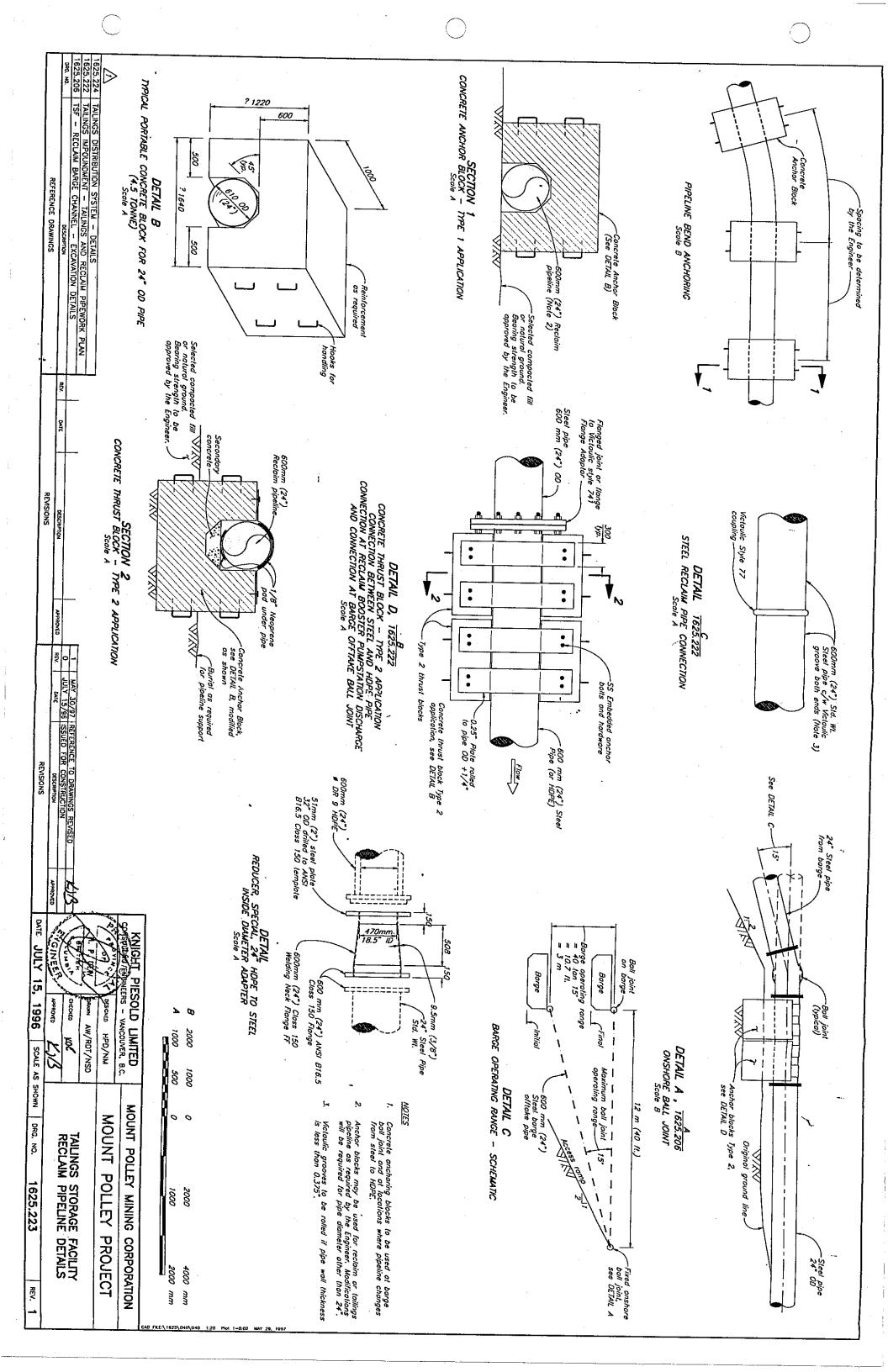


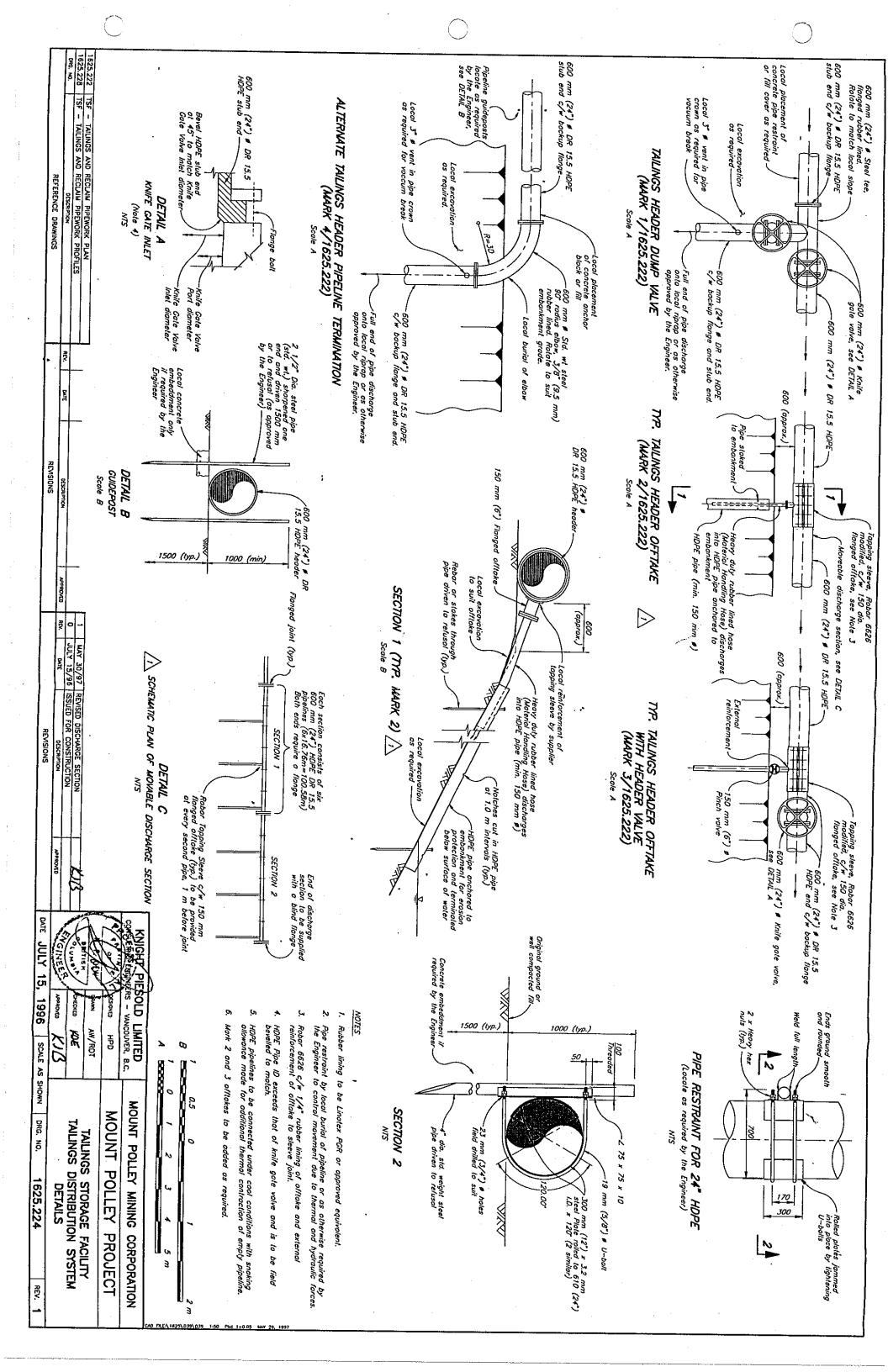


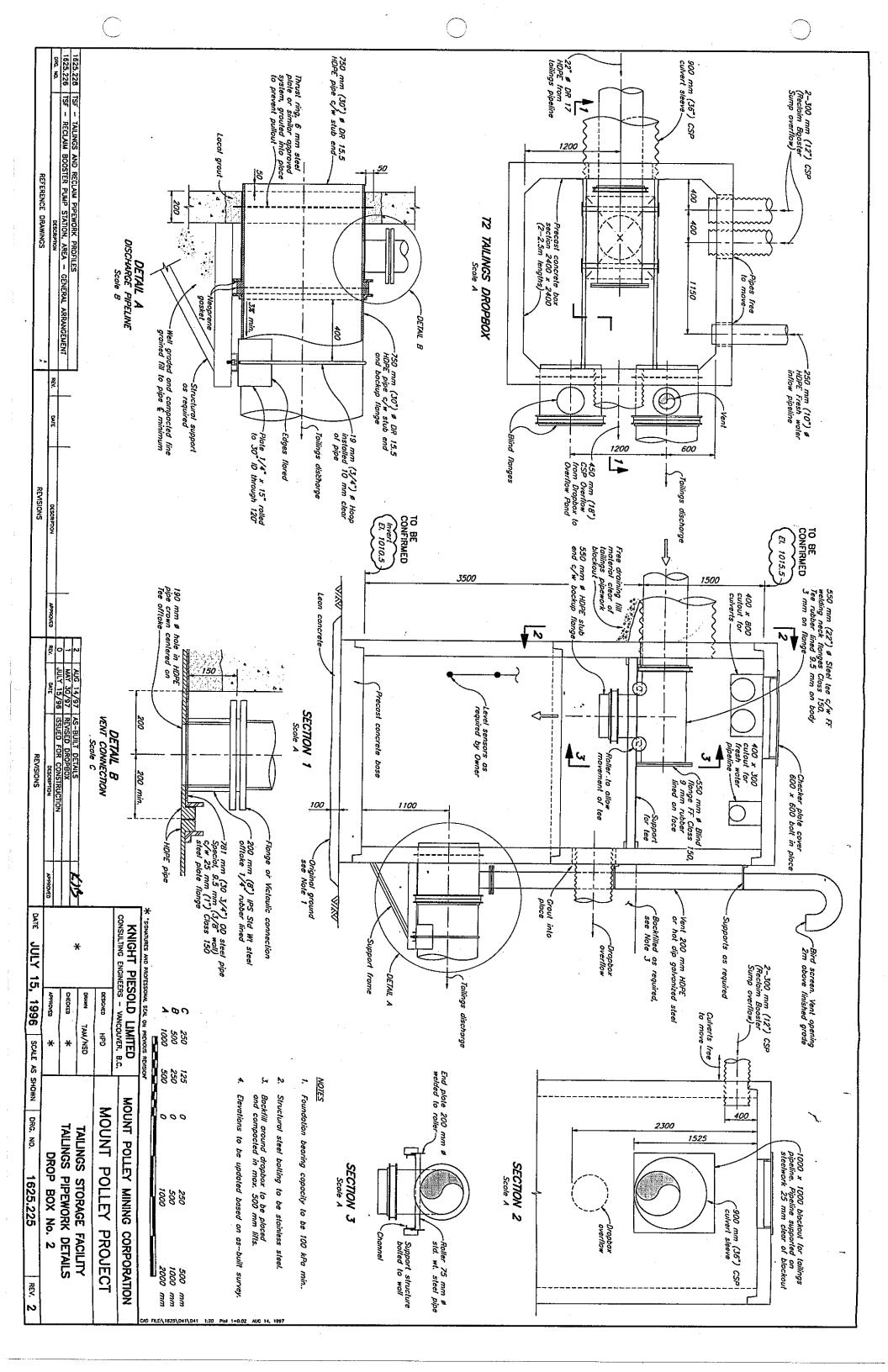


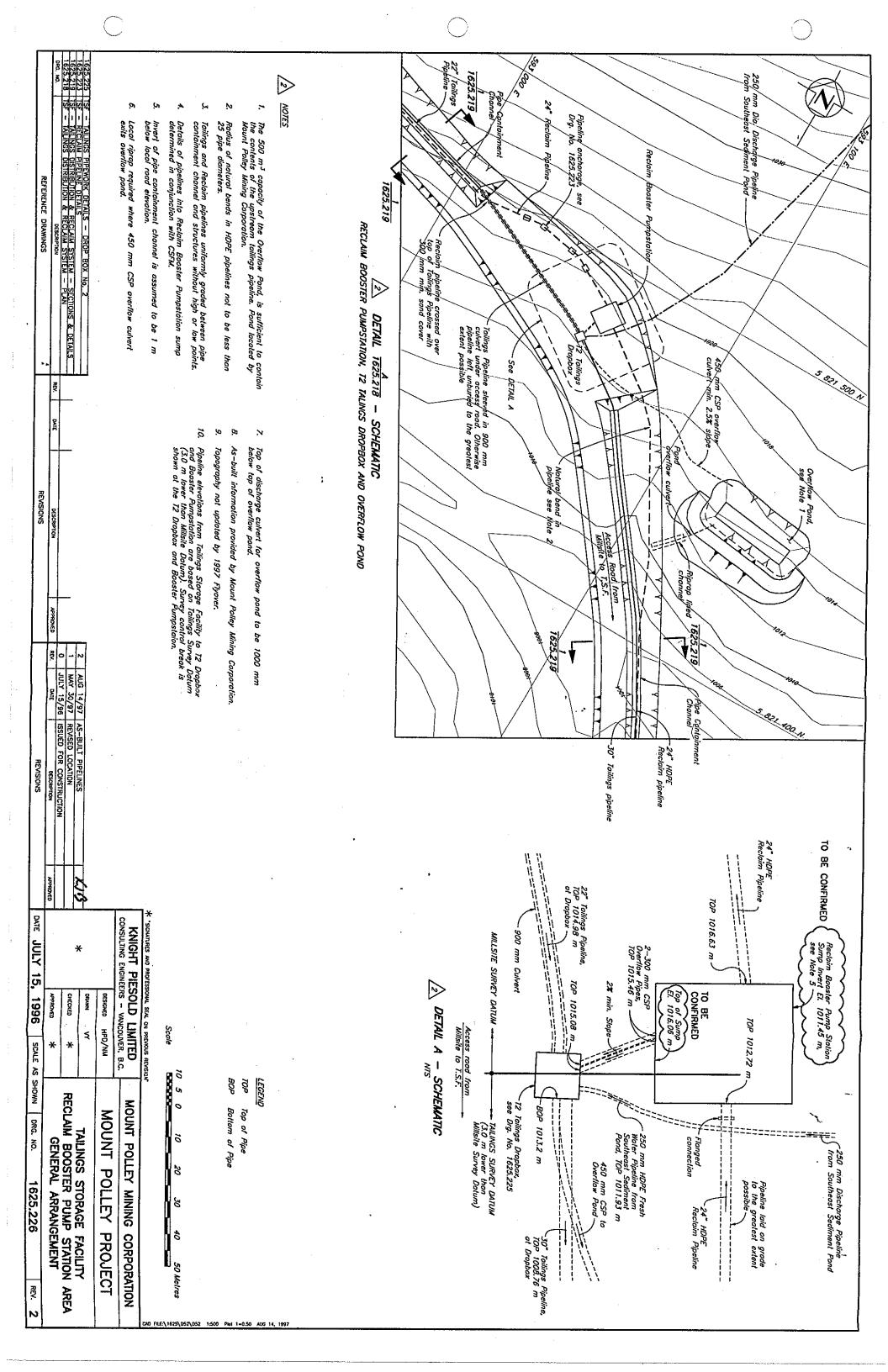


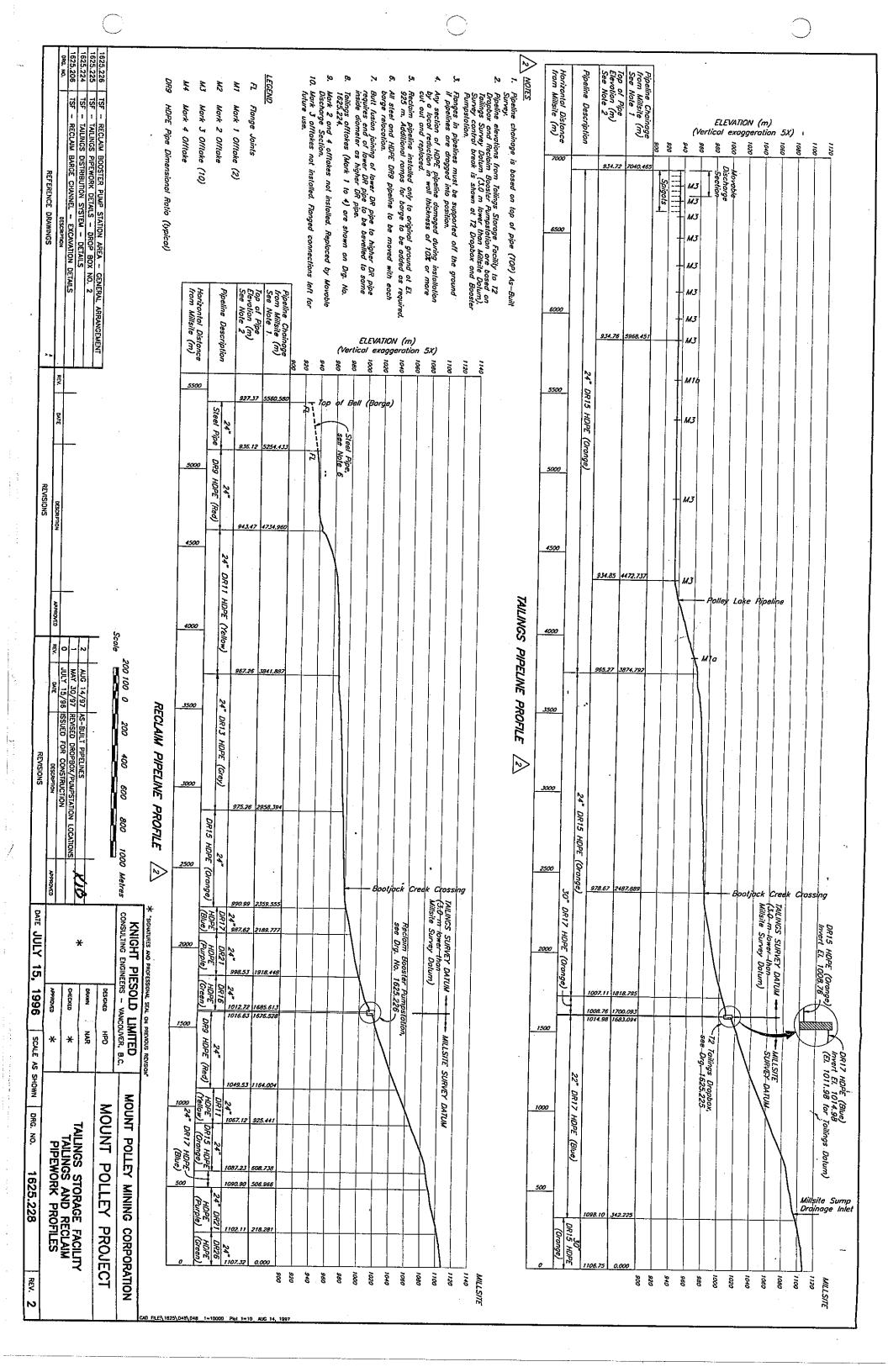


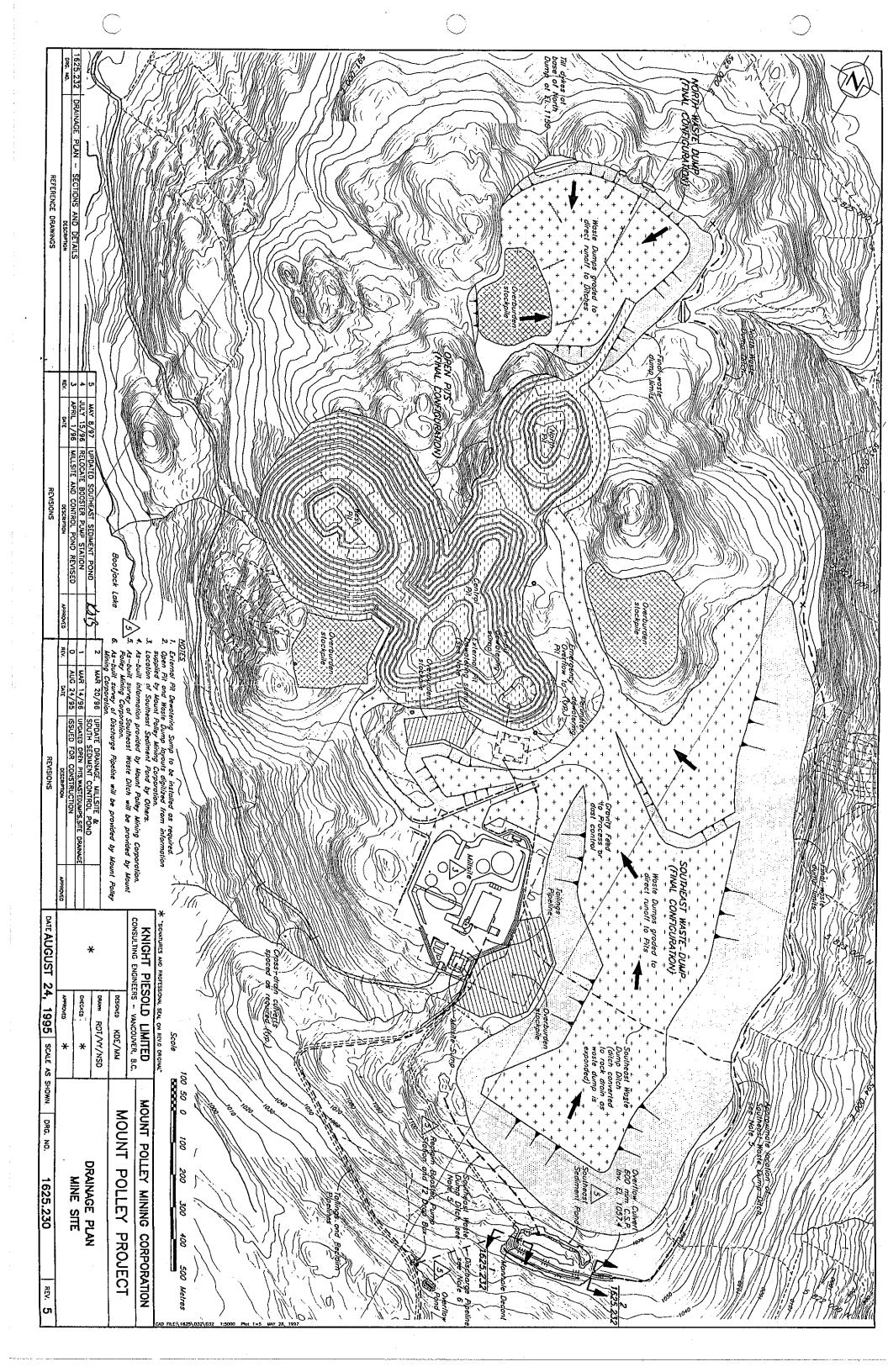


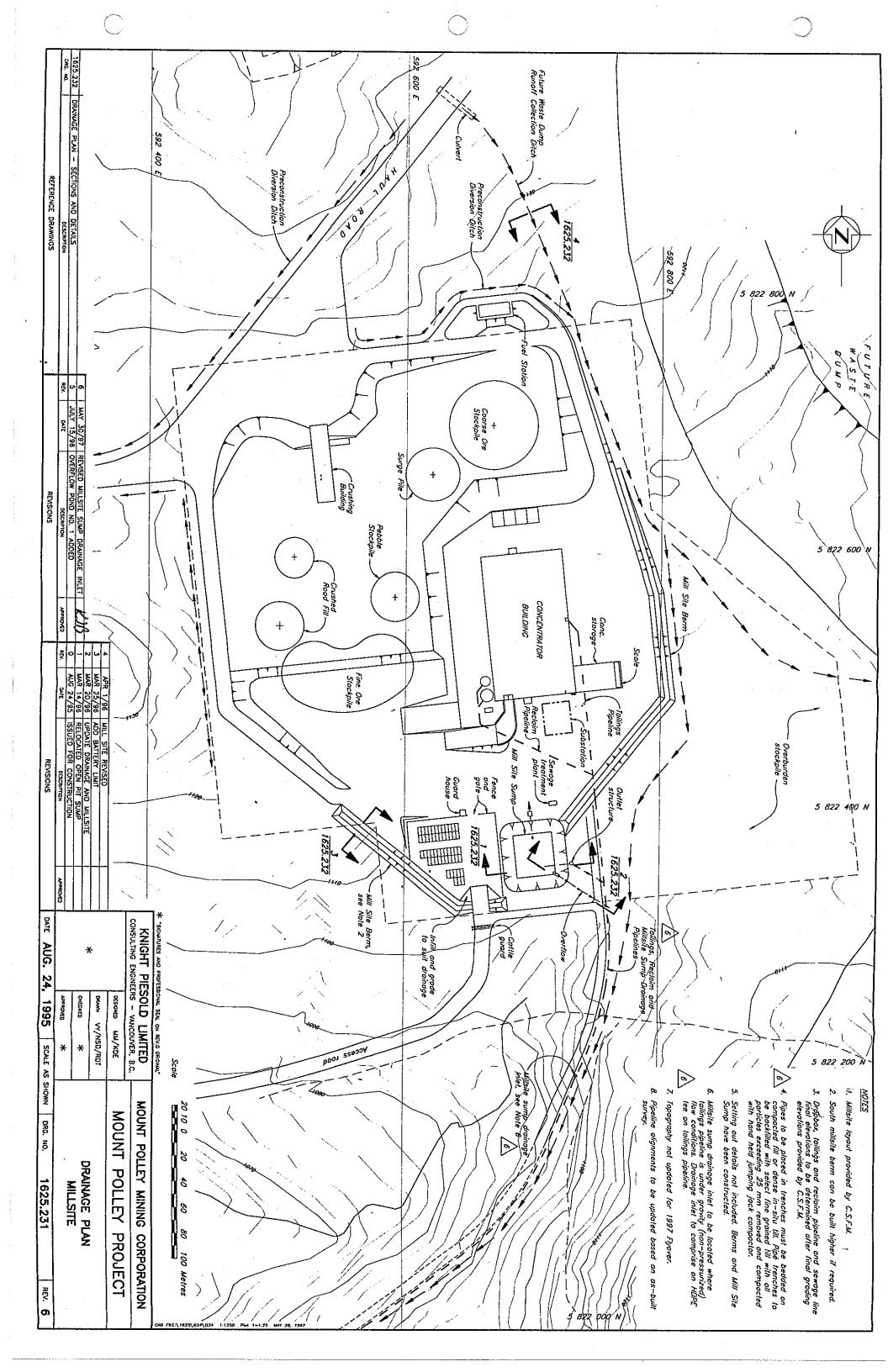


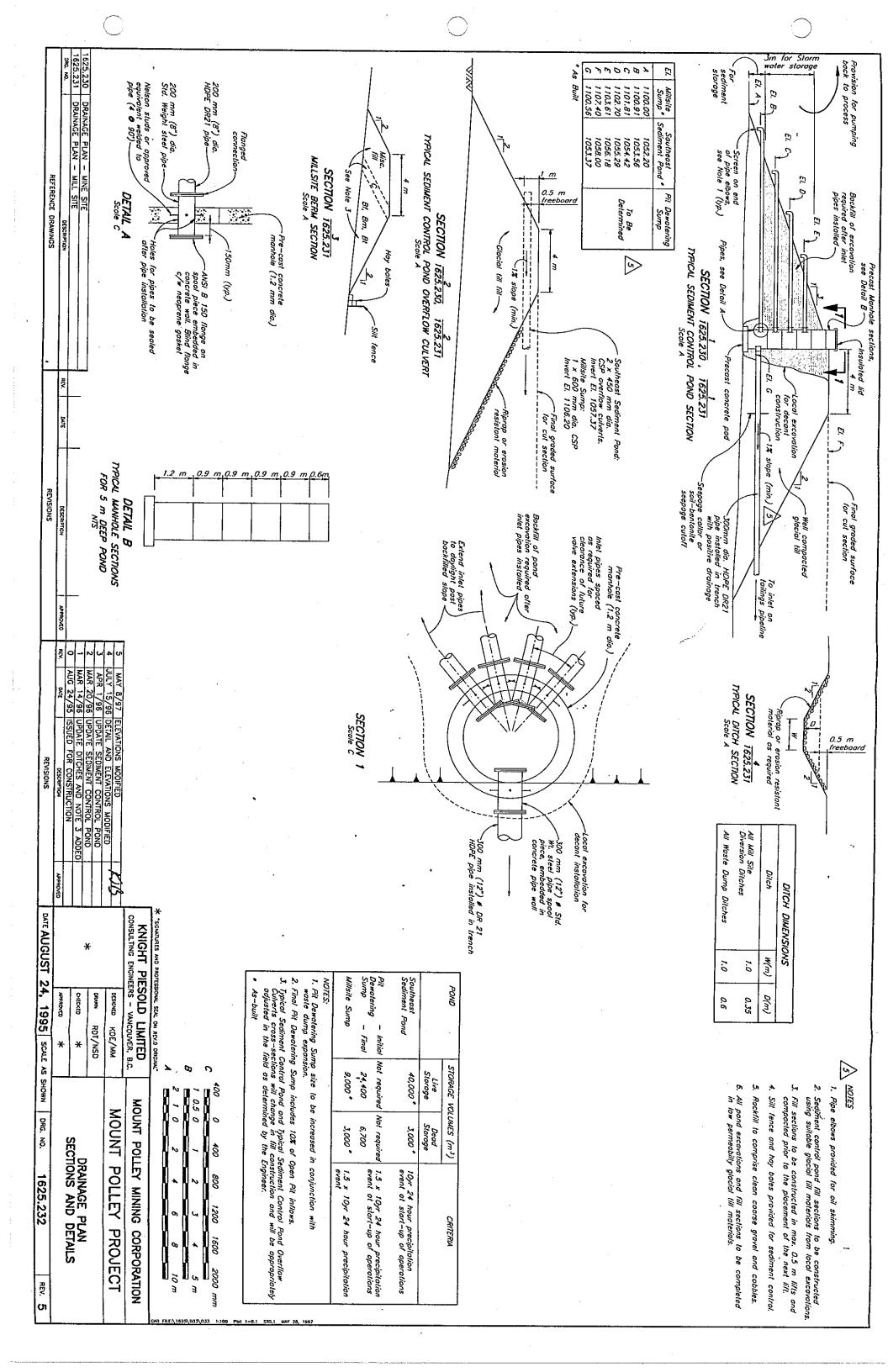


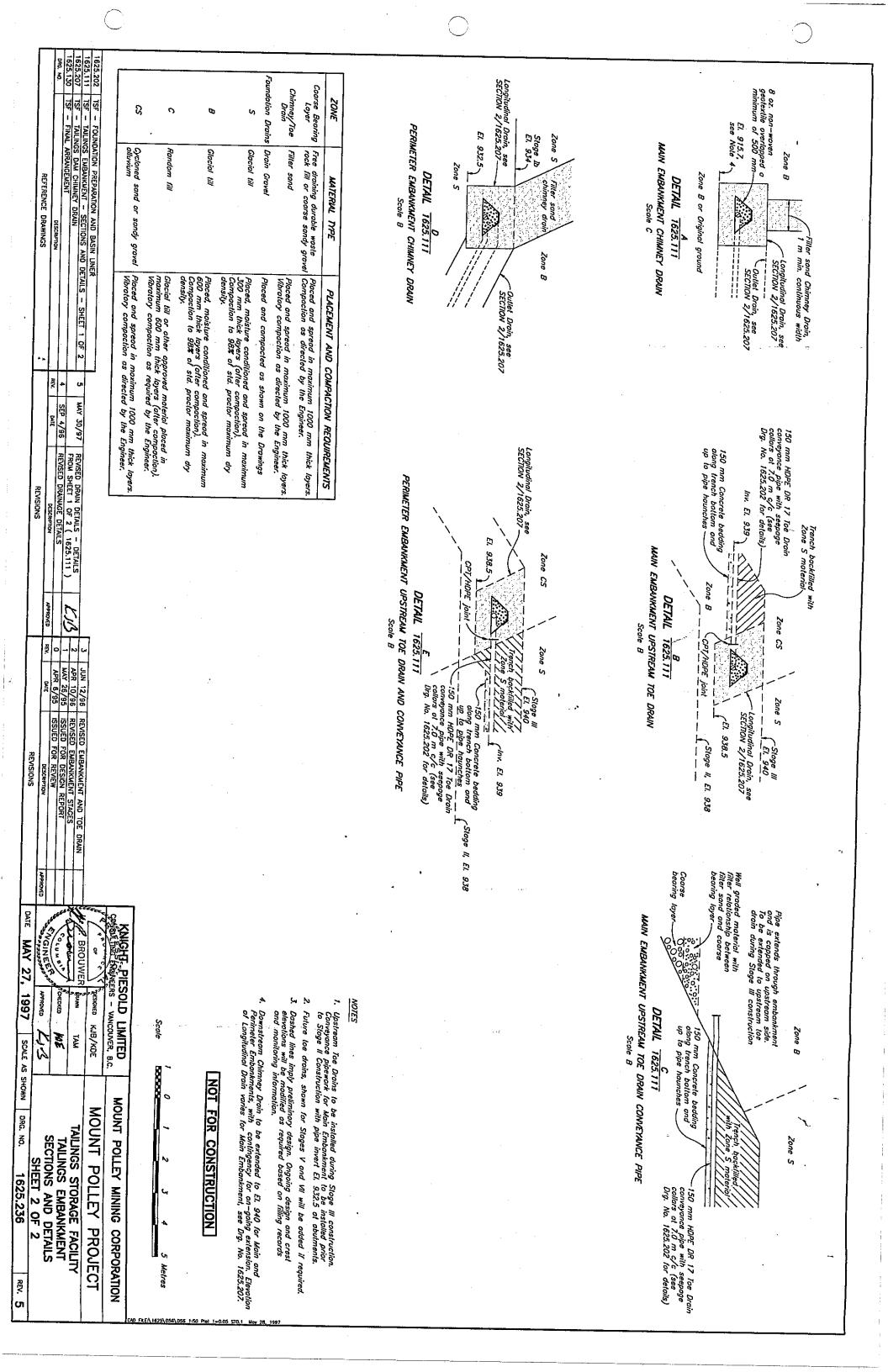










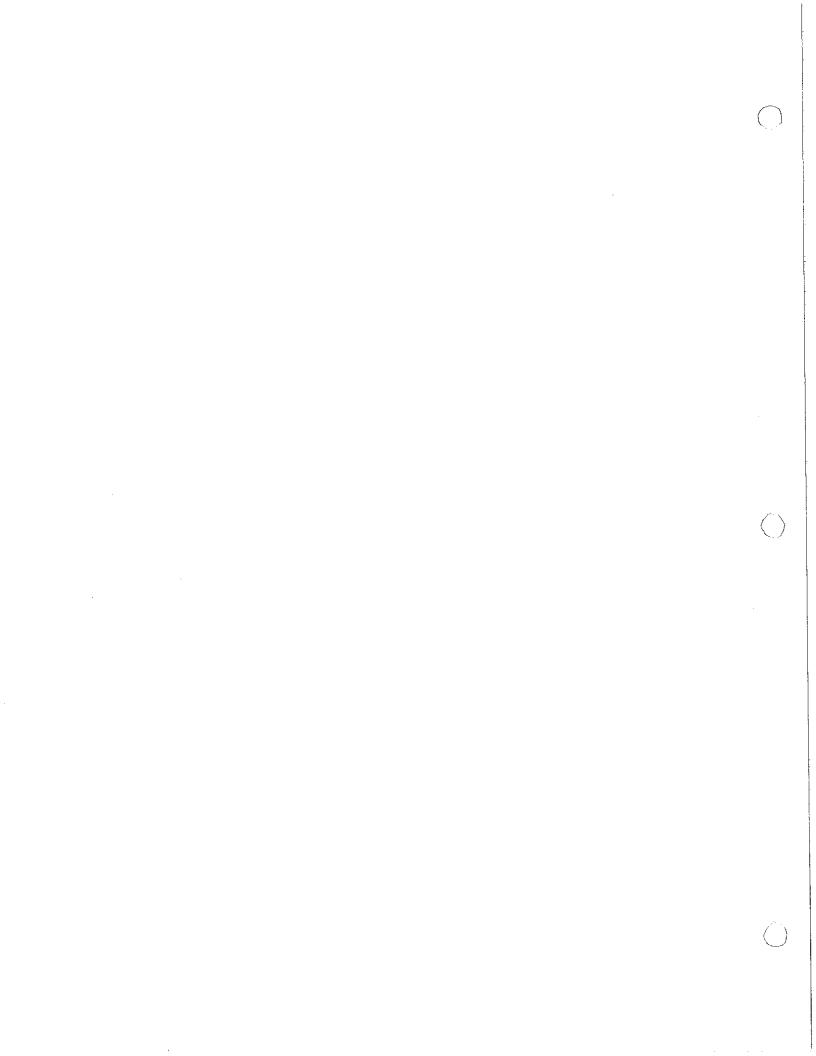




APPENDIX B

SUPPLEMENTARY INVESTIGATION REPORTS
B1 - R. E. GRAHAM FIELD REPORT
B2 - CONETEC FIELD REPORT







B1 - R. E. GRAHAM FIELD REPORT



Association des Ingénieurs-Conseils du Canada



R. E. GRAHAM ENGINEERING LTD.

August 12, 1996

302 - 1777 Third Avenue Prince George, B.C., V2L 3G7 Phone (604) 564-4304 Fax (604) 564-9323 E-mail graham@mag-net.com

North American Construction Ltd. 2150 Steel Road Prince George, B.C. V2N 2H9 File No. J-1245

Attention: Mr. Randy Peppler, A.Sc.T.

Dear Sir:

RE:

MT. POLLEY PROJECT EMBANKMENT FOUNDATION PIEZOMETER INSTALLATIONS

On July 21 to 24, 1996, we supervised the installation of six vibrating wire piezometers in drill holes in native soil below the centre of the proposed tailings dam. Mr. Bill Porter of North American Construction Ltd. (NAC) helped coordinate the work. Mr. Peter Procter, E.I.T. of Knight Piesold Ltd. (KPL) was on site to specify requirements for soil testing and piezometer installations.

Test hole logs describing soil conditions and showing details of the piezometer installations are on Plates 1 to 7, attached. Coordinates and elevations, as surveyed by NAC, are provided on the logs. An explanation of the terms and symbols used on the logs is attached for your reference.

Judging by the adjacent ground, about 0.5 m of soil had been stripped from the sites prior to the installations. Our drilling subcontractor, Peace Drilling and Research of Taylor, B.C., drilled the holes using 150 mm diameter solid augers and 200 mm outside diameter hollow-stem augers. In addition to our original scope of work, we were directed by KPL to perform continuous Standard Penetration Tests to about 10 m depth, log soil conditions, obtain samples, and install additional bentonite seals. Samples were submitted to KPL for laboratory analyses. Two piezometers were set in each of the holes except at Plane B, where sloughing soil conditions required the upper piezometer (B2-PE2-02) to be set in an adjacent hole. Prior to, and following installation of the piezometers, we hooked the indicator cables to the readout box to check that the sensors were operating.

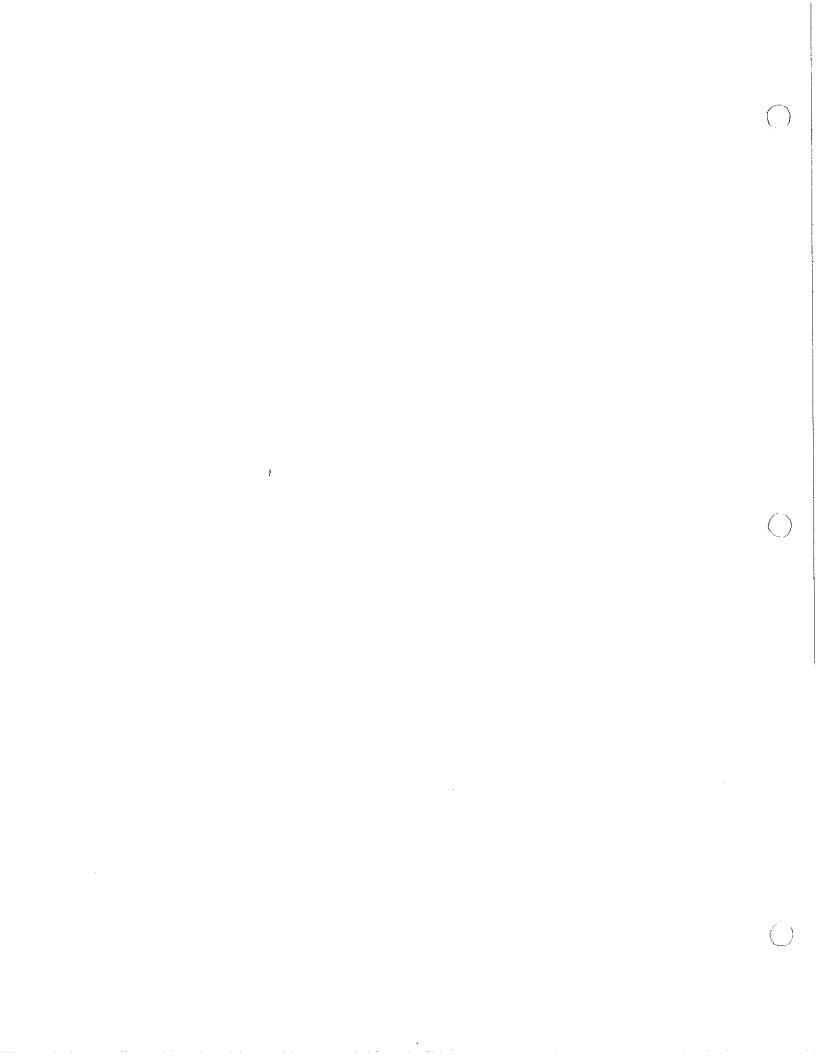
Thank you for the opportunity to work on this project. Please call me or Dave McDougall, P.Eng. if you have any questions, require any additional information, or if we can be of further assistance.

Yours truly,

R.E. Graham Engineering Ltd.

Per: J.D. Clarke, E.I.T.

JDC/pm Enclosures



图

R. E. GRAHAM ENGINEERING LTD.
302-1777 THIRD AVENUE
PRINCE GEORGE, B.C., V2L 3G7

TEST HOLE LOG

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ŀ	-	-		\dashv	\dashv		\dashv		_	\neg									$ /\rangle$	<u>600</u>			
ŀ	_		_					_						- be	low 3 rves 1	.65 m, stro 0-20 mm	ngly varv thick, an	red, ev-brown.	\Box	1	ļ		
)							_[4				20		-,	IX	26 500	İ	KK4	₊.o +
-	_	_				_	_					-	Ш	Ï						1	 	7 A 3	H
ŀ	\dashv									\dashv	_	1							\mathbb{N}	24			j.
ł	-		\dashv	-						1	_	1	Ш						ľÅ	1 24 600	Switched		
Ī												5							E	_	from solid to hollow		+
	\Box											↓	Ш						\mathbb{N}	24	stem auger © 4.9 m.		-
1		\dashv			-					}		-					-		ΙŇ	600	pp = 225 - 325 kPc		5.5
ŀ		-		\dashv	-		-				-	1						— 906.9 —	\vdash	É	<u></u>		Ţ
ı												16	\prod	SILT, tr	ace o	f fine sand	, sensitiv	e,	\mathbb{N}	21	Seepage from 5.8		1
							_			\Box		۱ <u> </u>	Ш			stic to low			ĮŇ	1500	to 9.75 m.		}
ŀ	_											-		be	ilow, c	B to 6.4 m prey, satura	ted.	2300	1	1			-
I								-		\vdash		1			•	=				7	pp = 150 kP	°	
ŀ	-							-			-	1,							JX	(13 600	5]	1/1/	Ţ
Ì												Ι΄			7	7 to 97 -	danth	eilt	1	4	—		Į.
ſ												4	$\parallel \parallel$	- ir	orn /. Jickina	3 to 8.7 n in hole.	r gebtu.	SIIL	Γ	7	Sampler sank		}
1				$\vdash \vdash$					<u> </u>	\square		-			-				JX	600		Y/X/	ŀ
-		\vdash	-	$\vdash \dashv$		-			-	\vdash		-	$\parallel \parallel$							1	01 7002	144	7.9
1				H					-	\Box	-	+8							\vdash	1			I
]							$ \rangle$	0 400	Sampler sank under weight		
ļ												4						•	V	\\	of rods.	ا المتناة	}
			<u> </u>					<u> </u>	_		 	4]							├	ដ្ឋា្ឋា	8.9
								-	-			+9	$\parallel \parallel$							/ 당	Piez	الرزرالي	9.1
7				H				-	-		 	1							//	1 600	0 ←	1///	
			-		-		-	-	 	t^{-}		1		.0.75				903.0	E			1///	
i													Ш	9.75— SAND	(TILL)	, silty, gra	velly, very	,	4	1	74		
						ļ						\downarrow_{10}	, Ш			grey-brow			4	(1)46		YZZ	
														-	Contin	ued on foll	owing she	eet.			PLATE NO). 1	



R. E. GRAHAM ENGINEERING LTD.
302-1777 THIRD AVENUE
PRINCE GEORGE, B.C., V2L 3G7

TEST HOLE LOG

<u>}_</u>	<u> </u>	_	IDO		-11 C	- A1/		l1	245	Ina:	TC () E	NVEST.96/07/21-24 DRILL SOLID/HOLLOW STEM AUGER HOLE NO. 96-A1	1
LOG	GE		JDC						243	UA				-
		A P	OCK	ICTH T PE	NETE	ROME	TER E	ROG.			Ξ			4
		× N ⊗ R	IATUR ESIDI	AL V	ANE ANE	SHEA	ir st ar s	TREN	STH GTH		7	님	SURFACE ELEVATION 912.67 m SIDENTS SOIL DESCRIPTION STANDPIPE OR PIEZONETER INSTALLATION	ı
	50		100)	150)	200		250	kPa	DEPTH	SYMBOL	SOIL DESCRIPTION	ı
		Wp	207	VATE	-Ö 30%	,Wn	40%	, м	50%		DE	S	SOIL DESCRIPTION S E INSTALLATION	ľ
	102	•	-24/	`_	3 <u>0</u> ^	<u>.</u>	102		302	\top	4.0	\neg	Continued from previous sheet.	_
\vdash	十	十	_	\top	\neg	_	1	十			-10	Ш	SAND (TILL), silty, gravelly, very]
┝━┼	1	7	\dashv	\top	1		7	丁					dense, grey-brown, moist.	
	十	十			\exists				T				125 mm Penetration,	┨
	-	7	\neg	\neg		\neg							100 on rock.	_
	1										-11			4
											•			-
		\perp	\perp			_ .		\perp					.	Н
	\perp	_	_		4	_ _	_	_		-				Н
	_	_	_		_	_	4	4	\bot					1
 	_	-		-	+	\dashv		-	-		-12		from 12.25 to 12.7 m SHT non-	d.
┡	-+		-				-	+					- 1 0 1 2.20 to 2.7 to 512.5 to 10.1	П
┞╌┼	-	-	\dashv	\dashv	+	-+	+	+	+	+			plastic, trace of fine sand.	
 	ᅪ		-+	+	\dashv	+	-			+			- below 12.7 m, some silt, brown.	
┝┼			\dashv	\dashv	-+	\dashv	十	\dashv	_	 			\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\	Ц
	一	-	1	+	7		\dashv	1	T		-13			Ц
	_		寸	1										H
			╗											H
													350 - 13.7	Н
					_	_	1	_	4.		14	Ш		Н
		_		_	_	\perp	_	_		_ _	-			Н
			_				\dashv	\dashv	_	- -	ł			Ħ
				}		\dashv		\dashv			-	\mathbb{H}		П
	\dashv		-				-+	+	╁	+	ł		14.9	П
	-		┵	\dashv		\dashv	-	-		-	 15		70	\square
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		一		寸		_	十	十	1		1	Γ"	897.3	Ц
				寸]	İ	Bentonite	Н
											1		End of Hole & 13.4 III.	${f H}$
									\Box		1		Seepage as noted.	H
											1		Artesian conditions, likely from silt layer coarse sand	Н
			\Box				\dashv		_	_ _	-		from 5.8 to 9.75 m depth. Hole quicking	H
			ļ ļ		_						-		and sloughing in silt layer.	Ħ
									\dashv	-	+	1		\prod
			_						-	\dashv	1			
-	_		-		}				-+	- -	1		NOTES:	
-			 				_		十		1		1) Ground surface cut down roughly	-
			_								7		0.5 m at hole location.	4
			_								Ţ	ľ	2) Piezometers are vibrating wire type	Н
		Ι.]		(SINCO Model 52611030), rated for 0-100 psi, connected to surface	Н
											1	ļ	via direct burial, polyethylene jacket	H
									_		4		cable (SINCO Model 50612604).	H
		<u> </u>					\square				1			+
<u> </u>		L.			\perp	انا					-			H
<u> </u>	_	<u> </u>	<u> </u>				Щ				4	ļ		T
 		<u> </u>				<u> </u>			_		\dashv	1		
-	 	-	-	\vdash		ļ		\vdash			-	-		
1	-		├-	-		<u> </u>	<u> </u>			-	+		PLATE NO 2	Ī

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R. E. GRAHAM ENGINEERING LTD.
302-1777 THIRD AVENUE

PRINCE GEORGE, B.C., V2L 3G7

TEST HOLE LOG

LOGGED	JDC	FILE	NO.	J-1	245	DA	TE	0F	INVEST.96/07/21-24 DRILL SOLID AUGER HOLE NO. 96-Blg
	STRENGT								
8	POCKET I NATURAL RESIDUAL	PENETRON VANE SH	METER LEAR :	ROG. STREN	стн		(E)	١,	
50	100	150	20	0	NGTH 250	kPa	Ξ.	BOL	SURFACE ELEVATION 916.97 m W COMMENTS
	WAT	ER CON	TENT	-∆			DEPTH	₹	SOIL DESCRIPTION
10%	w, D	30% ^w "	40	% W.	50%	·	_	S	
┠╌┼╌┼┈			-			 	0		DEPTH ELEV.
	 		\vdash	\dashv		\vdash			SiLT (TiLL), clayey, some sand, some gravel, compact to dense, brown,
					<u> </u>			\mathbb{N}	moist.
								M	- above 0.6 m, wet with slight orange mottling.
	<u> </u>				_	-	- 1		- from 0.6 m to 1.8 m, gravelly. $30 \times 30 \times 30 \times 30 \times 30 \times 30 \times 30 \times 30 $
┠╌┼╾┼╾	1- 1-					-			
 				+		\vdash			1.5
┠┼┼				十		1.		Щ	
				<u>,</u>			. 2		
				_	-	-	-		$\begin{array}{ c c c c c c c c c c c c c c c c c c c$
┠┷┼┷┼┈	 		\vdash	_	- -	-	1		- from 2.4 to 3.0 m, slight laminated
┠─┼─┼	 - -				-	\vdash			structure in places, clayey and sandy zones. $ \begin{array}{c c} 31 \\ \hline 650 \end{array} $ 2.7
	- -				_	1	- 3		- below 3.0 m, sandy.
							[]		
	 			_	<u>.</u>				$\left \begin{array}{c} \left \begin{array}{c} 40\\ 500 \end{array}\right \right $
	 			\dashv					- © 3.6 m, 40 mm thick varved
┡┼┼	 	-		-		-			layer,
					\dashv	十	4	HII	$\left \begin{array}{c} \left \begin{array}{c} 21 \\ 600 \end{array} \right \right $
									$-$ @ 4.8 m, 75 mm thick fine to $\begin{pmatrix} 48 \\ 600 \end{pmatrix}$
<u> </u>		<u> </u>			_	-			medium grained, silty sand layer.
	-		\vdash			-	- 5	Ш	SILT, some fine sand to sandy, dense
 	 		-	-		\vdash			to very dense, non plastic to low $\left X \right _{\overline{500}} = \left \overline{A} \right _{$
	1								plasticity, structureless, grey, damp to moist.
									$\frac{64}{300}$ pp >450 kPa 5.8
						ļ	- 6		
 		$\vdash\vdash$				╁	1		- from 6.1 to 6.7 m, clayey, weakly varved with occasional fine sand
	 			+	-	 	1		in varves, low plasticity.
							1		
					\bot		7		86 500 mm
<u> - -</u>				\dashv		┼			1/ \
 				+	\dashv	+-			SAND, fine to medium grained, silty, $\sqrt{\frac{61}{61}}$
				\dashv	\dashv	1			dense, grey, saturated.
						<u> </u>	- 8	1	Seepage from 7.3 to 8.2 m.
							٦		l an IXXXI I
 		<u> </u>			-	\vdash	Ì		SILT, some fine sand, very dense,
 	 		\vdash	+		+-	1		non—plastic to low plasticity, slight varved structure, grey, moist.
							9		- from 8.8 to 10.4 m, frequent wet,
							[]		Title to frieddam graffied, agridy rayers X 100 seepage
	<u> </u>			_		-			and zones (<150 mm thick).
 - - -	"			+	+	-	1		
1-1-			} - 	-		-	\mathbf{I}	$\parallel\parallel$	
				-		+-	10	Щ	Continued on following sheet. PLATE NO. 3
						Į	<u> </u>	L	Continued on following Sheet.

图

R. E. Graham Engineering Ltd. 302-1777 THIRD AVENUE PRINCE GEORGE, B.C., V2L 3G7

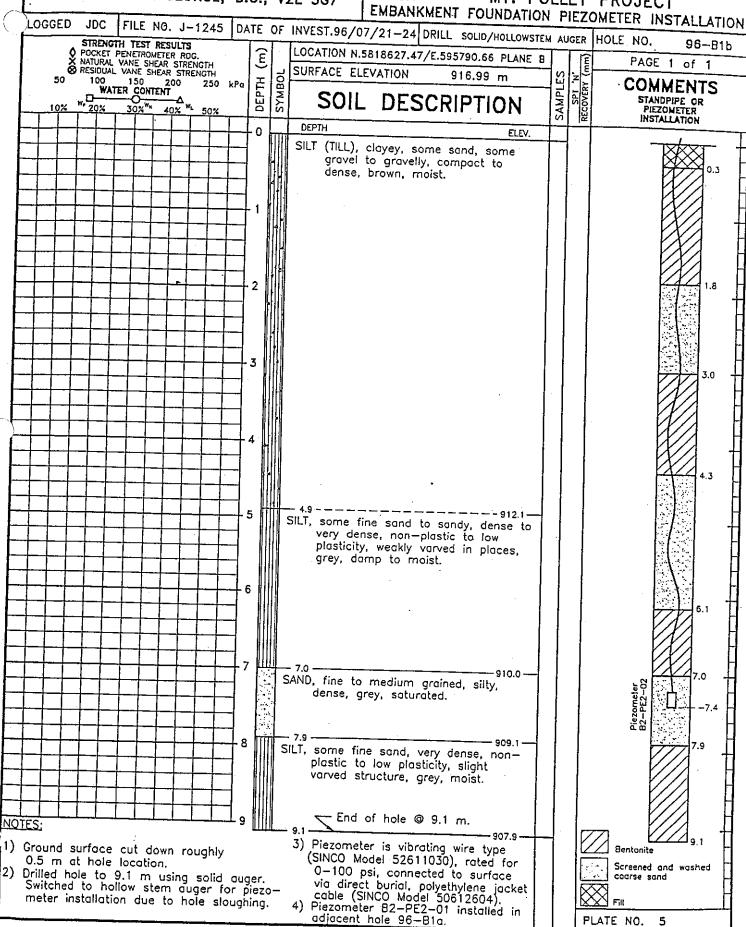
TEST HOLE LOG

Jan	250	JD	~	EU	- L	10		1245		TC :	٥٢				HOLE NO. 96-B1g
12000	GED			H 11				124	, 10,	T) 			- 1	
ŧ	<u> </u>	POCI	KET	PENE	TROM	IETER	RDC). 		\mathbb{E}		LOCATION N.5818628.27/E.595787.88 PLANE B	100	SPI 'N' RECOVERY (mm)	PAGE 2 of 2
		POCI NATU RESI	DUAL	VANE	E SH	EAR :	STRE	NGTH		I	占	SURFACE ELEVATION 916.97 m	ĘĻ	7 Y	COMMENTS
	50			ER S	CONT	20 ENT	10	250	kPa	I L	ξ E	SOIL DESCRIPTION	ĮŽ,	S S	STANDPIPE OR PIEZOMETER
	10%	w _p □-	2%)****	40	-∆ %	50%	:	8	Š	JOIL DESORIE HOR	S	æ	INSTALLATION
								_ _		10	<u> </u>	Continued from previous sheet.	4		
	_	_										SILT, some fine sand, very dense,	1		XX
	- -						\vdash	-				non-plastic to low plasticity, slight varved structure, grey,	b		l XXXI H
-		+-	-							1		moist.	IX	<u>65</u> 460	l XXX H
	+	+						-	-	١.,		 from 11.0 to 11.9 m, clayey, 			
	1	<u> </u>								-11		intermediate plasticity.			
													\times		ρρ >450 kPa
	<u> </u>							_		-					l 💢 H
-						\vdash				-		- from 11.9 to 12.4 m, occasional			l <i>777</i> 7''' H
┢╌┼╴		+						+		12		(<20 mm thick) sand seams.	\mathbb{N}	84 460	
 	+	+-						-			Ш		<u> </u>	700	12.2
 -	1										$\parallel\parallel$	 from 12.5 to 13.0 m, silty clay, 			
								\perp		1		intermediate plasticity.	\times	j	
	+	+-				$\sqcup \downarrow$		\dashv		13		- from 13.0 to 13.9 m, strongly]	Possible
		-		\vdash		-				1		varved, occasional (<20 mm thick) wet sand seams.		1	Seepage from 13.0
\vdash	+-	+						-				 below 13.4 m, grey-brown. 	∇	72	to 13.9 m.
	1	1]			ĮŇ	300	PP = 430 KPG ///
		<u> </u>								14	Ш	— from 14.0 to 14.9 m, silty clay,		Ì	
	╝	ļ									$\ \ $	intermediate plasticity.	\triangleright	,	pp = 250 kPa
		-								-	Ш			1	to 350 kPa
$\vdash \vdash$		+-		_					\dashv	1				ļ	≥ 14.6 1
	+	+-								15		 from 14.9 to 15.5 m, strongly 	-	,	₩ C3
\Box	_	 								† 13	Ш	varved.	$ $ $ $	62 450	15.2
											Ш		1	4	
		_	<u> </u>	<u> </u>						-		 below 15.5 m, wet, very stiff to stiff with depth. 	1		pp = 300 kPa
		_	ļ	_	<u> </u>					4		5 ,,,, 5,,,, 25,,,,,,,,,,,,,,,,,,,,,,,,,		1	
-		-		┢		-	-	\dashv		 16				1	pp = 100 kPa
 	十		┼				-			1			K	4	16.3
	1	1	<u> </u>]	ш	16.5 900.5	-}-	┪	16.5
												/			
				_						+		End of hole @ 16.5 m.		İ	Bentonite
	-		ļ	\vdash	<u> </u>	-	<u> </u>		_ _	-		Seepage as noted.			Service and market
\vdash		+-	├	 	\vdash	-	 	┝╌┼	+	1	-	Hole sloughing in sand layer			Screened and washed coarse sand
	+	+-	+	 	-	1			_	1	1	from 7.3 to 8.2 m.			Bentonite and Silt
1	_	\top	1	1					\dashv	1		Hole filled with water to 3.0 m depth	1		
										I		when left open for 2 hr period.	1	1	XX 5101
										_		when tere open for 2 in period.			Slough
				 _	<u> </u>		ļ	\sqcup	_	4		NOTES:	j		1
-	4	+	┼	┼	-	 	-	\vdash		-		1) Ground surface cut down roughly			
+	+	+	\vdash	 		+	-	\vdash		+		0.5 m at hole location. 2) Piezometer is vibrating wire type		1	1
) +		+-	╁	T	\vdash	+	 	 	\dashv	1		(SINCO Model 52611030), rated for			
	\dashv	+	1	T	1	T	1]		0-100 psi, connected to surface via direct burial, polyethylene jacket	١	Ĭ	
												cable (SINCO Model 50612604).			
	\perp	_			ļ	_	<u> </u>		\perp	+		3) Piezometer B2-PE2-02 installed in adjacent hole 96-B1b.			1 51 175 115
								1		ı	1	adjacent noie 30-610.			PLATE NO. 4

P

R. E. GRAHAM ENGINEERING LTD. 302-1777 THIRD AVENUE PRINCE GEORGE, B.C., V2L 3G7

TEST HOLE LOG





R. E. Graham Engineering Ltd. 302-1777 THIRD AVENUE PRINCE GEORGE, B.C., V2L 3G7

TEST HOLE LOG

/- ^E												EMBANKMENT FOUNDATION PLEZOMETER INSTALLATIO	
ن .	OGG	ED_	JDC	F	ILE	NO.	: J-	124	5 DA	TE ()F	INVEST.96/07/21-24 DRILL SOLID/HOLLOWSTEM AUGER HOLE NO. 96-C1	╝
Γ			STREN([LOCATION N.5818392.41/E.595478.24 PLANE C SURFACE ELEVATION 915.71 m SOIL DESCRIPTION STANDPIPE OR PIEZONETER INSTALLATION	
1		×	NATURA RESIDU	Lν	ANE S	SHEAR	STRE	NGTH		(m)	ᆜ	SURFACE ELEVATION 915.71 m SEE COMMENTS	
	:	50	100		150		00	25	0 kPa	Τ̈́	Ä	STANDPIPE CR	
			w, 1 20%	AIG	-Ö,	W,	∆ _w	50	17	DEPTH	SYMBOL	SOIL DESCRIPTION S PREZONETER INSTALLATION	ļ
\vdash	-	0 <u>% </u>	20%	T	30%	┬-1	1	30.	<u>`</u>	-0		DEPTH ELEV.	\top
r	╅									- 0		SILT (TILL), clayey, some sand, some	
									ļ			gravel, compact to dense, low $X = \frac{1}{430}$	Щ
	I			1	_		_					plasticity, brown, moist. — slight orange mottling above 0.6 m.	Н
┸		<u> </u>		\perp	_ _			-				29 500	Н
┝	+-	 		+		+				1			H
\vdash		┼		+	+	+	-		<u> </u>		4		
上	1-	1		7		\top					-	$\left \begin{array}{c} \left \begin{array}{c} \frac{29}{600} \\ \end{array} \right \right $	
													\mathbb{H}
L	_ _	<u> </u>		\downarrow		_	╄-			2		V <u>48</u>	+
-		-		+		+	-	\vdash	_				Н
┢	- -	┼	 -	+	+	+	+	\vdash	_	1		- below 2.4 m, grey-brown, then	H
┢		-		+	+	+-	\vdash		_	i	Ш	grey with depth. $\left \begin{array}{c} 25 \\ 500 \end{array}\right $	
t		\top		\top		Ť				3		- below 3.0 m, sandy.	4
												28 \\ \lambda_{28}	Н
L		_	$\vdash \vdash$	_		-	1			-		3.4	Н
-		\vdash	 	+	-		+-			-			Н
7	+-	╁		\dashv	╬	+	╁			┧.	Ш	100 mm 100 lon rock,	
ſ		\vdash		_		┪	+-			† 4			Ц
ı				\Box								Seepage	Н
		Ţ										- @ 4.6 m, 200 mm thick wet seam. \ 4.6 \ \frac{4.5}{2} \ \ \ \ \ \ \ \ \ \ \ \ \	Н
ŀ		-		\dashv	-	+	ļ		 	-		- © 4.9 m, approx. 50 mm thick fine 200 ESG 4.7 to medium sand layer, brown.	H
ŀ	 -	+	 	+		+	┿	$\left \cdot \right $		- 5		/ soturated/ / 「 a SIH b l e e	╵╢
ŀ	+	十	\vdash	╁	-		+-		-	1	1	CLAY, silty, occasional fine sand in	
ı]		varves, (varves ~10 mm thick), very seepage stiff, intermediate plasticity,	Н
						\bot	1	ļļ		4		brown, moist.	H
Į		-	│ 	+	-	- - -			-	 6		- © 5.6 m, 50 mm thick silt layer, trace of fine sand, non-plastic,	+
-		+-	┼─┼	\dashv		- -	+	-		-			
ŀ	+-	+	 	_	\dashv	+	+-	+		1	m	SILT, some fine sand to find sandy,	Ц
İ]		non-plastic, brown, saturated.	-
Ţ		-	\prod	_[[~	1				ļ 7		$\left \begin{array}{c} \left \frac{28}{510} \right \leftarrow \end{array} \right $	+
ŀ	_	+-		-			+-	 		-		- below 7.3 m. grev, sensitive.	Н
ł	-	+-	+	十	-+	+	+-	1		1	$\parallel \parallel$		
ł	\dashv	+		+	\dashv	+-	+-			1			
t										18	$\parallel \parallel$, #
	\perp			I	\bot					↓	$\parallel \parallel$	8.4 -8. -8. -8. -8. -8. -8. -8. -8.	' H
ŀ	+	_	+	\dashv			+	-		-	\mathbb{H}	1111 8.4	Н
ŀ	+		++	+			+	+-		┨		SAND (TILL), gravelly, silty, very dense, grey, moist.	
ł		+	 - -	\dashv	-	+	+-	-		┨_			\prod
1	\dashv	+	 	7			+	1		 9		X 84 450	Ļ
4]	H	9.4	-
[\bot			\bot	\bot	\perp	1_	1		4		- below 9.6 m, SILT (TILL), sandy,	H
- -		+	+	-		_	+	-	-	4	$\parallel \parallel$	some gravel.	
ł		+	+-			-	+	+	-	+10	Ш	Continued on following sheet. PLATE NO. 6	
- 1	- 1	1		- [1	1	i İ	1	1	Continued on following street.	

图

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PRINCE GEORGE, B.C., V2L 3G7

TEST HOLE LOG

CLIENT NORTH AMERICAN CONSTRUCTION LTD.
PROJECT MT. POLLEY PROJECT
EMBANKMENT FOUNDATION PIEZOMETER INSTALLATION

)		EU E NO	1 1246	DATE	ΛΕ II	1755 05 /07 /	21-24 DBILL	SOLID /HOLLOWST	FW ALI	GER	HOLE NO.	96-C1
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EXPLANATION OF TERMS AND SYMBOLS USED ON TEST HOLE LOGS

SOIL DESCRIPTION

Soil is classified in accordance with the International Society of Soil Mechanics and Foundation Engineering (ISSMFE) system as described in the 1992 Canadian Foundation Engineering Manual (CFEM). Descriptions for each soil type encountered are divided by contact lines at appropriate depths. Each description has a corresponding graphic symbol which relates to soil type.

Major Soil Division:

The major soil division is the main fraction of soil and constitutes at least 35% by weight. Soil is classified as GRAVEL, SAND, CLAY, SILT, or ORGANICS according to the criteria on pg. 3.

Where applicable, a bracketed term such as (FILL) or (TILL) is included to describe soil genesis.

Grain size and Shape:

Grain size descriptions for soils follow the criteria on pg. 3.

The shape of coarse and oversized particles is described as:

angular - sharp corners subangular - slightly rounded corners subrounded - no angular corners

rounded - smooth rounded surface platy — flat, plate shaped

Soil Composition: -

The following terms are used to describe the percentage of soil components by weight based on laboratory sieve analyses or field estimates:

Descriptive Term	Percentage Range
"and" and sand, and gravel etc. "and" and sand, and gravel etc. "some" some silt, some gravel etc. "trace" trace sand, trace silt etc.	>35% 20 - 35% 10 - 20% 1 - 10%

The amount of cobbles and boulders, in increasing proportion, is described as:

isolated — - occasional — - frequent — - numerous

Compactness and Consistency:

The following terms are used to describe the compactness of cohesionless soils based on the Standard Penetration Test (SPT) or field estimates:

Descriptive Term	SPT 'N' Index
very loose	0 - 4
loose	4 - 10
compact	10 - 30
dense	30 — 50
very dense	over 50

The following terms are used to describe the consistency of fine-grained soils based on unconfined compressive strength as determined by field or laboratory tests, or estimates:

Descriptive Term	Unconfined Compressive Strength (kPa)
very soft	<25
soft	25 — 50
firm	50 – 100
stiff	100 200
very stiff	200 - 400
hard	>400

Structure:

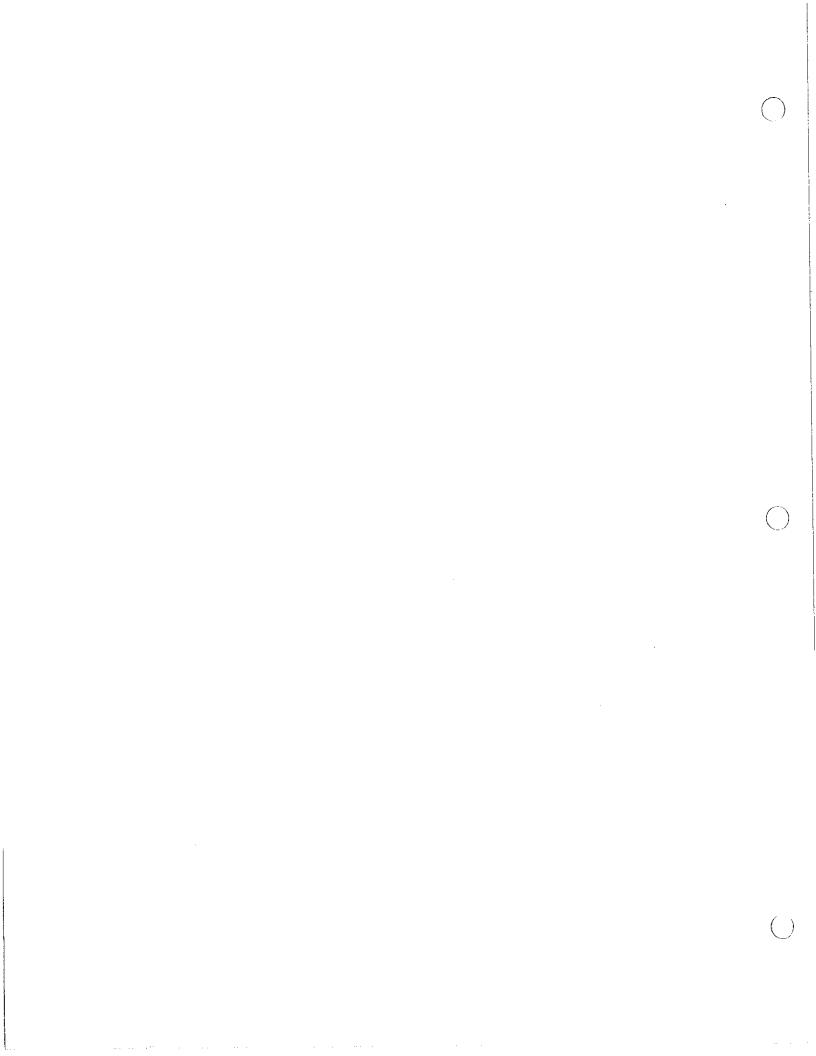
Soil macrostructure and microstructure is described.

Plasticity:

Plasticity of fine-grained soils is estimated or determined from Atterberg Limit tests based on the plasticity chart on pg. 3.



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EXPLANATION OF TERMS AND SYMBOLS USED ON TEST HOLE LOGS

SOIL DESCRIPTION (cont.)

Colour and Odour:

Colour and odour of soils is described, especially where it may indicate organic inclusions or give evidence of soil contamination.

Inclusions:

The quantity of inclusions is described using the same relative amount terms used for cobbles and

Water Content:

Soil moisture, in increasing amount, is subjectively described as:

dry — damp — moist — wet — saturated — excess water

SOIL SAMPLES

Graphic symbols indicate the depth and condition of soil samples:

Disturbed

Undisturbed

Undisturbed samples may be taken with tubes, from blocks, or by coring. All other types of samples

FIELD TESTS

Field test results commonly reported on test hole logs include:

Standard Penetration Test (SPT) (ASTM D1586):

The SPT results are reported as the 'N' index over the length of the recovered sample, at the appropriate depth. The 'N' value denotes the number of blows of a 63.5 kg hammer, freely dropping 760 mm, required to drive a 50.8 mm diameter split—spoon sampler 305 mm.

Dynamic Cone Penetration Test (DCPT):

Dynamic cone penetration test results are shown graphically. The number of blows required to drive a solid cone 305 mm is shown opposite the depth. The method of driving the cone is the same as for the SPT test described above.

Field Vane Test (FVT) (ASTM D2573-72):

Undrained shear strength of cohesive soils is measured using a 100 mm long by 50 mm diameter vane. Test results for peak and residual strengths are graphically reported at the appropriate depths using the following symbols:

Peak Shear Strength 🗙 Residual Shear Strength

Pocket Penetrometer and Torvane Tests:

Unconfined compressive strength and undrained shear strength measurements may be taken on samples or excavation walls using a pocket penetrometer or hand torvane. Pocket penetrometer results are shown graphically using \Diamond symbols. Torvane results are reported using the same symbols used for the field vane test.

LABORATORY TESTS

The following symbols are used to denote laboratory test results:

Natural water content, W_N (ASTM D2216). Atterberg Plastic Limit, W_p (ASTM D424).

Atterberg Liquid Limit, WL (ASTM D423).

Mechanical grain size (sieve) analysis or hydrometer test, or both (ASTM D422). MA

Unconfined compressive strength test on an undisturbed sample (ASTM D2166).

SO, Test for concentration of water-soluble sulphates.

Unit weight of soil or rock.

Dry unit weight of soil or rock.

COMMENTS

Groundwater conditions are indicated using the following symbols: groundwater table

Comments often included are additional test results, drilling progress, monitoring equipment installation details, and other relevant information.

24 R. E. GRAHAM ENGINEERING LTD.

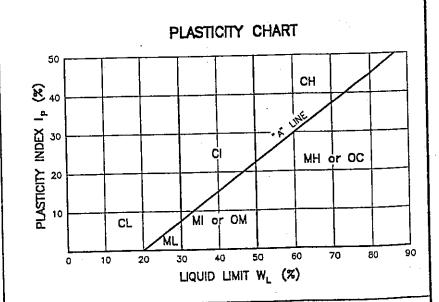
ISSMFE SOIL CLASSIFICATION SYSTEM

ł							
		MAJOR	DIVISION	GROUP SYMBOL	GRAPHIC SYMBOL	TYPICAL DESCRIPTION	LABORATORY CLASSIFICATION CRITERIA
<u> </u>				GW	4	WELL-GRADED GRAVEL AND SANDY GRAVEL MIXTURES WITH LESS THAN 5% FINES.	$C_u = \frac{D_{so}}{D_{to}} > 6$, $C_c = \frac{(D_{so})^2}{D_{to} \times D_{so}} = 1$ to 3
		무 문 문	CLEAN GRAVEL	GP		POORLY-GRADED GRAVEL AND SANDY GRAVEL MIXTURES WITH LESS THAN 5% FINES.	NOT MEETING ABOVE REQUIREMENTS.
	<u>.</u>	GRAVEL - 60.0 r DIAMETER		GM		SILTY GRAVEL AND SILT—SAND—GRAVEL MIXTURES WITH MORE THAN 15% FINES.	ATTERBERG LIMITS BELOW "A" LINE.
	20 CE	2.0	DIRTY GRAVEL	GC		CLAYEY GRAVEL AND CLAY-SAND-GRAVEL MIXTURES WITH MORE THAN 15% FINES.	ATTERBERG LIMITS ABOVE "A" LINE.
	COARSE-GRAINED			SW		WELL-GRADED SAND AND GRAVELLY SAND MIXTURES WITH LESS THAN 5% FINES.	$C_u = \frac{D_{60}}{D_{10}} > 6$, $C_c = \frac{(D_{20})^2}{D_{10} \times D_{60}} = 1$ to 3
	COARS	o.o. ER	CLEAN SAND	SP		POORLY-GRADED SAND AND GRAVELLY SAND MIXTURES WITH LESS THAN 5% FINES.	NOT MEETING ABOVE REQUIREMENTS.
		SAND 0.06 - 2.0 r DIAMETER		SM		SILTY SAND AND SILT-GRAVEL-SAND MIXTURES WITH MORE THAN 15% FINES.	ATTERBERG LIMITS BELOW "A" LINE.
		0.0	DIRTY SAND	SC		CLAYEY SAND AND CLAY-GRAVEL-SAND MIXTURES WITH MORE THAN 15% FINES.	ATTERBERG LIMITS ABOVE "A" LINE.
t		<u></u>	SILT	ML		INORGANIC SILT, VERY FINE SAND, ROCK FLOUR, AND SANDY SILT OF LOW PLASTICITY.	
		PL	ELOW "A" LINE ON MI			INORGANIC SILT OF INTERMEDIATE PLASTICITY.	
	10S	NEG	LIGIBLE ORGANIC CONTENT.	МН		INORGANIC SILT AND MICACEOUS OR DIATOMACEOUS SOIL OF HIGH PLASTICITY.	
			CLAY CL			INORGANIC CLAY OF LOW PLASTICITY, GRAVELLY, SANDY OR SILTY CLAY, 'LEAN' CLAY	SEE PLASTICITY CHART BELOW
`.	FINE-GRAINED		OVE "A" LINE ON ASTICITY CHART.	CI		INORGANIC CLAY OF INTERMEDIATE PLASTICITY. SILTY CLAY.	
Ä	E		LIGIBLE ORGANIC CONTENT.	CH.		INORGANIC CLAY OF HIGH PLASTICITY, "FAT" CLAY.	
		ORG	ANIC SILT & CLAY	ОМ		ORGANIC SILT	-
	-	BEI PI	OW "A" LINE ON ASTICITY CHART	ос		ORGANIC CLAY	
l		HIGHLY	ORGANIC SOIL	Pt	***************************************	PEAT AND OTHER HIGHLY ORGANIC SOIL	HIGH ORGANIC CONTENT AND FIBROUS TEXTURE.

GRAIN SIZE

Coarse-grained soil and silt is identified on the basis of grain size diameter as follows:

0.002 - 0.006 mm 0.006 - 0.020 mm 0.020 - 0.060 mm SILT: Fine Medium Coarse 0.06 - 0.20 mm 0.20 - 0.60 mm 0.60 - 2.00 mm SAND: Fine Medium Coarse 2.0 - 6.0 mm 6.0 - 20.0 mm 20.0 - 60.0 mm **GRAVEL:** Medium Coarse 60.0 - 200 mm COBBLES:





BOULDERS:

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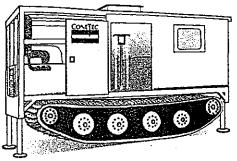
> 200 mm

Page 3 of 3



B2 - CONETEC FIELD REPORT







Geotechnical and Environmental In Situ Testing Contractors

ConeTec Field Report

Cone Penetration Test Results Mount Polley Tailings Dam

Prepared for:

Knight Piésold Ltd., Vancouver, BC

- July 31, 1996 -

PRESENTATION OF CONE PENETRATION TEST RESULTS MOUNT POLLEY TAILINGS DAM INVESTIGATION

Prepared for:

KNIGHT PIÉSOLD LTD. Vancouver, B.C.

Prepared by:

CONETEC INVESTIGATIONS LTD.

July 31, 1996

TABLE OF CONTENTS

1.0	INTRODUC	TION
2.0	FIELD EQU	IPMENT AND PROCEDURES
	2.1	Electric Cone Testing with Shear Wave Velocity Measurements
3.0	CONE PEN	ETRATION TEST DATA AND INTERPRETATION
	3.1	CPT Data
	3.2	Pore Pressure Dissipation Data
	3.3	Shear Wave Velocity Data

APPENDICES

Appendix A CPT Plots

Appendix B Tabular CPT Interpretations

Appendix C Summary of Pore Pressure Dissipation Times and

Equilibrium Pore Pressures

Appendix D Pore Pressure Dissipation Plots

Appendix E Shear Wave Velocity Data

1.0 <u>INTRODUCTION</u>

This report presents the results of a cone penetration testing (CPT) program which included pore pressure dissipation and shear wave velocity measurements. The tests were conducted on the tailings dyke at Mount Polley Mine, between July 27, 1996 and July 28, 1996. A total of 10 CPTs were performed at the site. The CPTs were performed on the embankment crest and pushed into the mine waste adjacent to the embankment. The work was carried out under the direction and supervision of Knight Piésold.

2.0 FIELD EQUIPMENT AND PROCEDURES

2.1 Electric Cone Testing

The cone penetration tests (CPT's) were carried out by *ConeTec Investigations Ltd.* of Vancouver, B.C. using an integrated electronic cone system. A 10 ton compression type cone as shown in figure 1 was used for all of the soundings. This cone has a tip area of 10 sq. cm. and friction sleeve area of 150 sq. cm. A piezometer element 6 mm thick is located immediately behind the cone tip. The compression cone is designed with an equal end area friction sleeve and a tip end area ratio of 0.85. The cone system used during the program was capable of recording the following parameters at 5 cm depth intervals:

Tip Resistance (Qc)

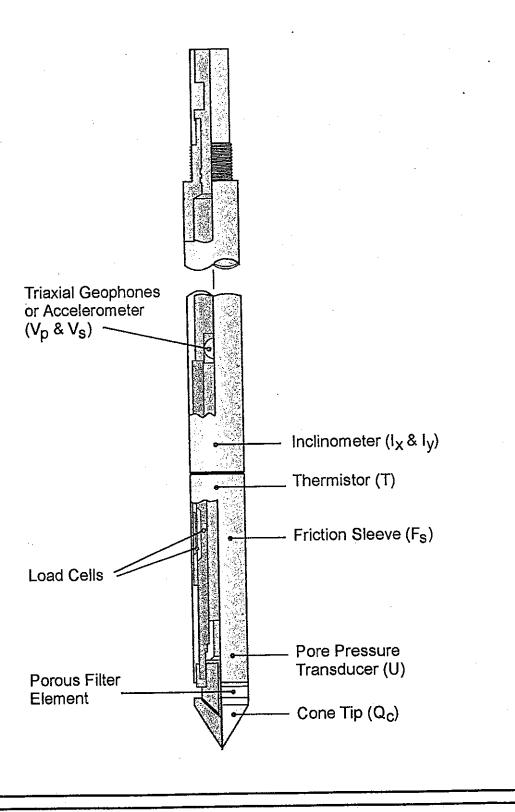
Sleeve Friction (Fs)

Dynamic Pore Pressure (Ut)

The above parameters were printed simultaneously on a printer and stored on digital media for future analysis and reference.

The porous plastic pore pressure element was located directly behind the cone tip. Each of the elements were saturated in glycerin under vacuum pressure prior to penetration. Pore pressure

Piezo Cone Penetrometer





Geofechnical and Environmental In Situ Testing Contractors

ConeTec Investigations Ltd. • 9113 Shaughnessy St. Vancouver B.C. Canada V6P 6R9 • Tel: (604) 327-4311 • Fax: (604) 327-4066

dissipations were recorded at 5 second intervals when appropriate during pauses in the penetration.

A complete set of baseline readings were taken prior to each sounding to determine temperature shifts and any zero load offsets. Establishing temperature shifts and load offsets enables corrections to be made to the cone data where necessary. These corrections are critical, especially where the load conditions are relatively low, and generally are the single largest source of error with respect to the accuracy of cone data.

2.2 Seismic Cone Testing

The equipment and procedures used in this investigation in general were as developed at UBC and reported by Rice, 1984, Laing, 1985 and Robertson et al., 1986. The procedure was incorporated within the cone penetration test (CPT) and conducted when the cone penetration test was stopped to add additional push rods. At the end of the first push and at one meter intervals afterwards shear wave velocity measurements were made. The CPT rods are one meter long, and therefore accurate depth intervals were ensured by always pushing the cone rods one meter and ending at the limit of the hydraulic ram system. Before taking wave velocity measurements the rods were decoupled from the drill rig.

Shear waves were generated by striking a steel beam which was held down by the rear stabilizers of the drill rig. A rear axle weight of ten tonnes provided excellent coupling of the beam to the ground surface. The steel beam was 2.3 meters long, 0.2 meters wide and 0.2 meters deep. The beam was equipped with steel end plates to permit striking the beam with a sledge hammer. The centre of the beam was offset horizontally from the cone rods a distance of 0.96 meters. This offset was accounted for in the velocity calculations.

A friction reducer was used to reduce the friction between the soils penetrated and the cone rods behind the cone. The particular reducer used is oversized compared to conventional reducers to facilitate the decoupling of the cone rods from the soils in an effort to produce a high quality

signal and produce a high signal to noise ratio.

The beam was struck lightly for shear wave generation using a 7.25 kg sledge hammer in a horizontal direction, parallel to the active axis of the transducer, first from one end and then the other. Each wave was inspected and the procedure was repeated if necessary. Occasionally, excessive vertical components of the striking force resulted in excessive generation of compressional waves and the procedure was repeated. A contact trigger between the beam and the hammer produced accurate triggering times and allowed for the accurate timing of S wave markers.

After each pair of wave traces was recorded, inspected and saved, the two traces were overlayed on a digital oscilloscope screen and the arrival and first crossover times were selected.

The shear waves were detected by a horizontally active geophone in the cone which was offset above the cone tip a distance of 0.2 meters. This offset is accounted for in all calculations. The geophone signal was amplified by 100 times prior to processing in the storage oscilloscope. The oscilloscope sampled at a frequency of 20 kHz (i.e., 20,000 samples per second).

The cone was pushed using a specially modified drill rig MARL 10, having a down pressure capacity of approximately 14 tons. The cone testing rig was supplied and operated by Mud Bay Drilling of Surrey, B.C.

3.0 CONE PENETRATION TEST DATA AND INTERPRETATION

3.1 CPT Data

The cone penetration test data are presented in graphical form in Appendix A. Penetration data is referenced to existing ground surface. Sections in the embankments that were augered through are labelled as drilled out on the CPT plots.

The following table summarizes the CPT test program:

Table 1 - Summary of CPT Test Programme

CPT Sounding	Test Date	Total Depth	Drill Outs	Total Dissipation
	######################################	(m)	(m)	Time (s)
CPT 96-1(1)	96/07 <i>/</i> 27	9.7	***************************************	3010
CPT 96-2 (4)	96/07/27	11.8	7.30 to 7.80	1150
CPT 96-3 (2)	96/07 <i>/</i> 27	8.25		415
CPT 96-4 (3)	96/07/27	10.25	CONTRACTOR CONTRACTOR	535
CPT 96-2 (4A)	96/07/28	15.85	0.0 to 7.6	2440
CPT 96-1 (5)	96/07/28	7.65	neren en en en en en en en en en en en en e	
CPT 96-2 (6)	96/07/28	9.6		355
CPT 96-3 (7)	96/07/28	10.85		80
CPT 96-4 (8)	96/07/28	11.50	***************************************	1055
CPT 96-5 (9)	96/07/28	10.0	***************************************	1105

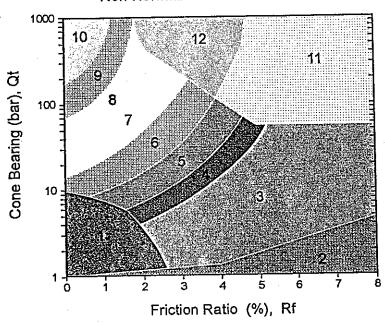
The stratigraphic interpretation is based on relationships between cone bearing Qc, sleeve friction Fs, and penetration pore pressure Ut. The friction ratio Rf (sleeve friction divided by cone bearing) is a calculated parameter which is used to infer soil behaviour type. Generally, cohesive soils have high friction ratios, low cone bearing and generate large excess pore water pressures. Cohesionless soils have lower friction ratios, high cone bearing and generate little in the way of excess pore water pressures.

The interpretation of soils encountered on this project was carried out using correlations developed by Robertson et al., 1986 (upper chart in figure 2). It should be noted that it is not always possible to clearly identify a soil type based on Qc, Fs and Ut. Occasionally soils will fall within different soil categories on the classification charts. In these situations, experience and judgment and an assessment of the pore pressure dissipation data should be used to infer the soil behaviour type. Computer tabulations of the interpreted soil types along with certain other geotechnical parameters for each cone hole is presented in Appendix B. It should be noted that the unit weights shown in the tabulated output are generated by the program based on the

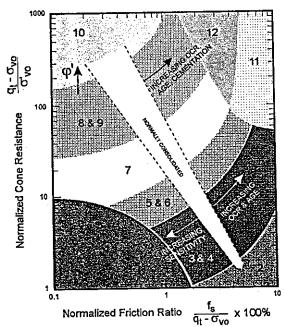
CPT Classification Chart

(after Robertson 1990)

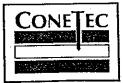
Non-Normalized Classification Chart







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Geotechnical and Environmental In Situ Testing Contractors inferred soil type. The unit weights may not be accurate given the high metal content of the tailings that were tested.

3.2 Pore Pressure Dissipation Test Results

Pore pressure dissipations were automatically recorded during pauses in penetration. The pore pressure data was recorded at 5 second intervals. The pore pressure dissipation data for each CPT is included on the data disk. A summary of dissipation times and equilibrium pore pressures is presented in Appendix C. Plots of selected dissipations are presented in Appendix D. Generally, the water pressures in the dam were fairly low, most in the range of one to three metres of water pressure. The identification of the phreatic surface is difficult due to the fine grained nature of the soils plus the large downward gradient.

3.3 Shear Wave Velocity Test Results

The variation in shear wave velocity with depth calculated for each hole (SCPT-2 and SCPT-4A) are shown in Appendix E plotted at a depth midway between the one meter test intervals. The pertinent data for each of the seismic profiles is presented in Tables 1 and 4A in Appendix E. Based on the seismic cone test data presented, the shear wave velocities in the penetrated tailings vary between about 175 m/sec and 312 m/sec.

We trust that the information presented in this report is sufficient for your purposes. If you have any questions regarding the contents of this report, please do not hesitate to contact our office.

Sincerely,

David J. Woeller, M.Eng., P.Eng. ref: a:\96-163.rpt

APPENDIX A

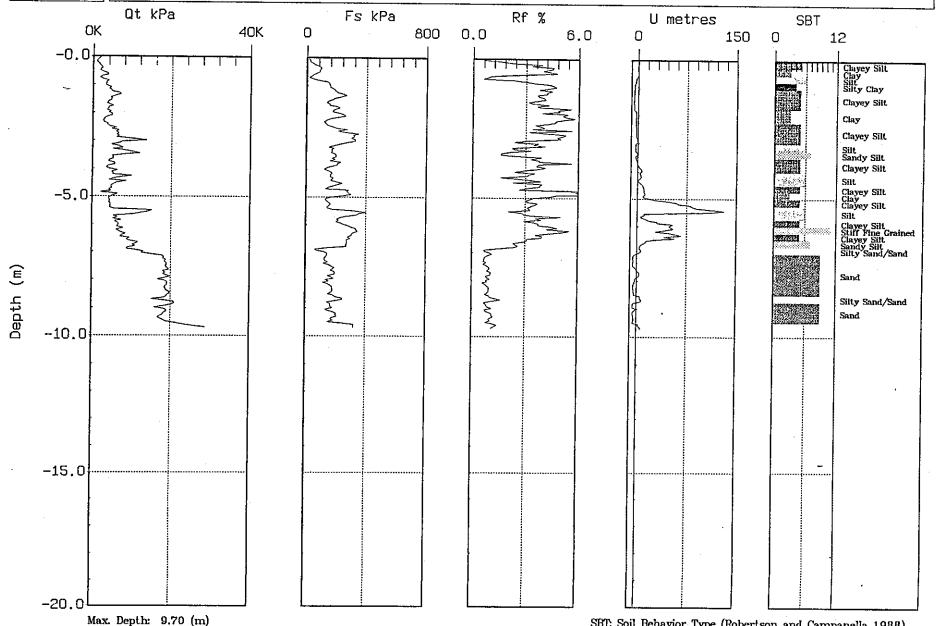
CPT PLOTS



KNIGHT PIESOLD

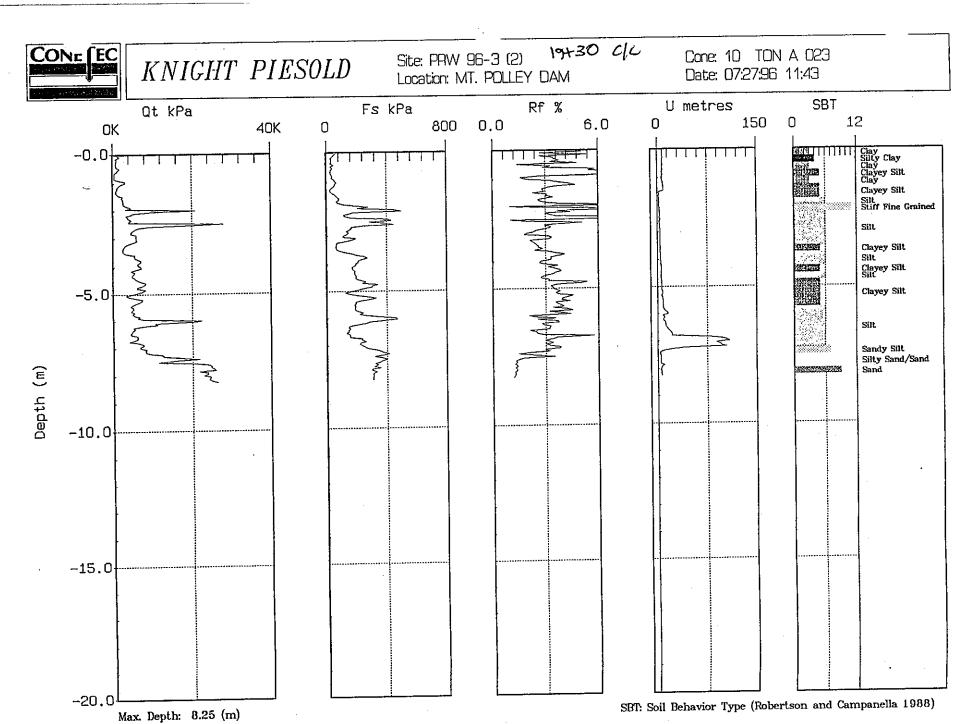
20+00 CLL Site: CPT 96-1 (1) Location: MT. POLLEY DAM

Cone: 10 TON A 023 Date: 07:27:96 08:15



Depth Inc.: 0.05 (m)

SBT: Soil Behavior Type (Robertson and Campanella 1988)



Depth Inc.: 0.05 (m)

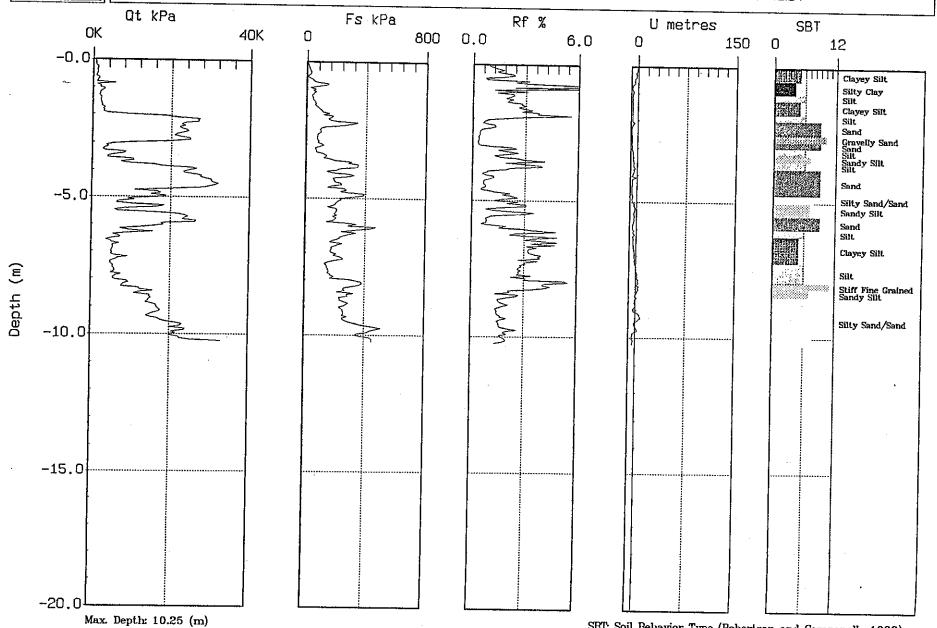


Depth Inc.: 0.05 (m)

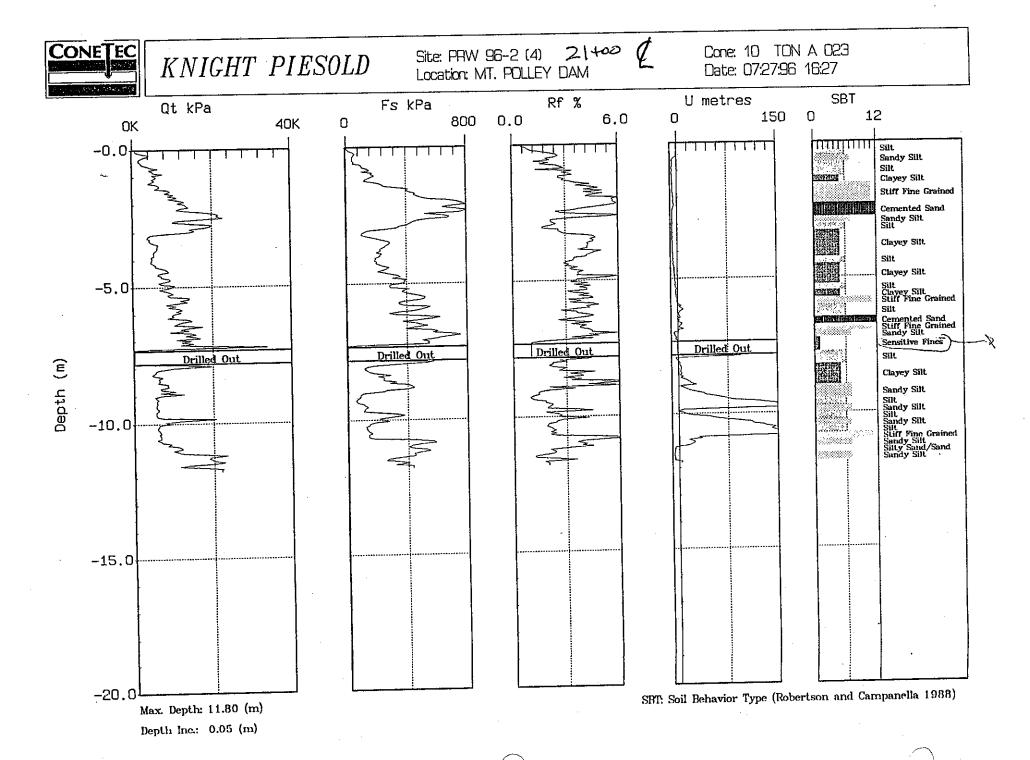
KNIGHT PIESOLD

18t00 dL Site: PAW 96-4 (3) Location: MT. POLLEY DAM

Cone: 10 TON A 023 Date: 07:27:96 13:51



SBT: Soil Behavior Type (Robertson and Campanella 1988)

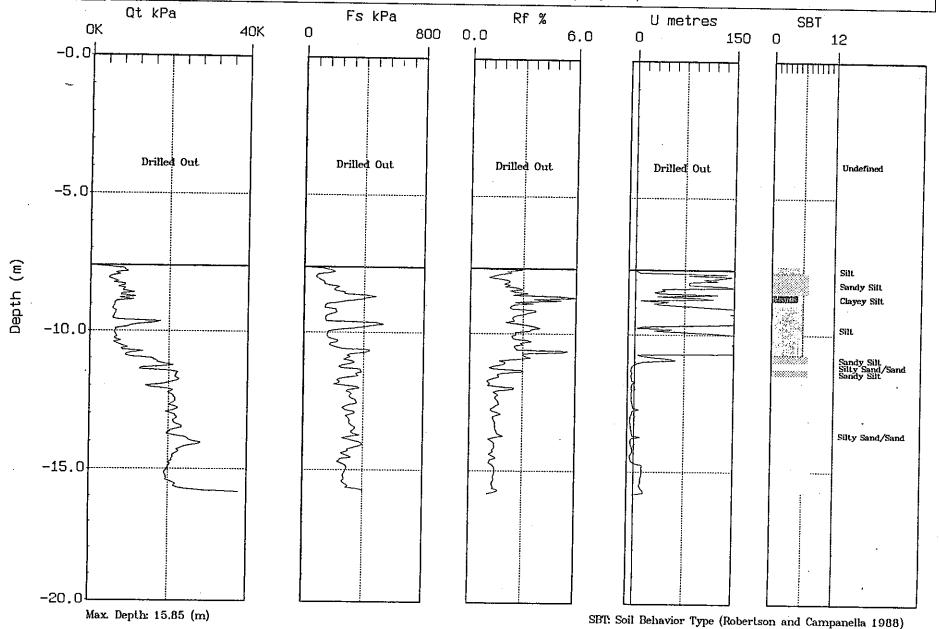


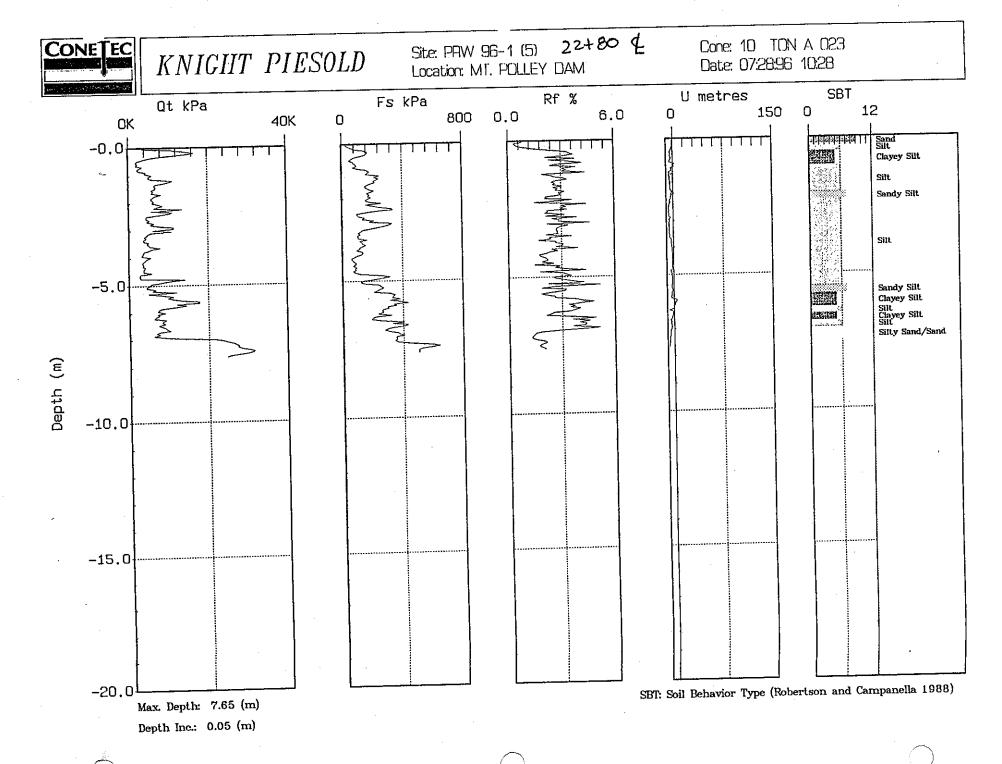


KNIGHT PIESOLD

Depth Inc.: 0.05 (m)

Site: PRW 96-2 (4A) & 21100 Cone: 10 TON A 023 Location: MT. POLLEY DAM Redrill Date: 07:28:96 08:43





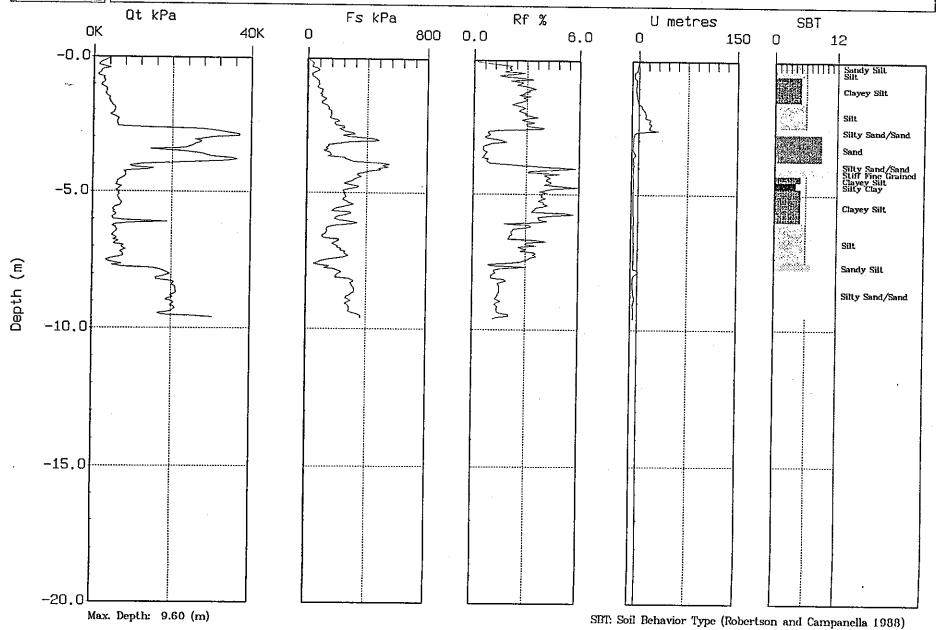


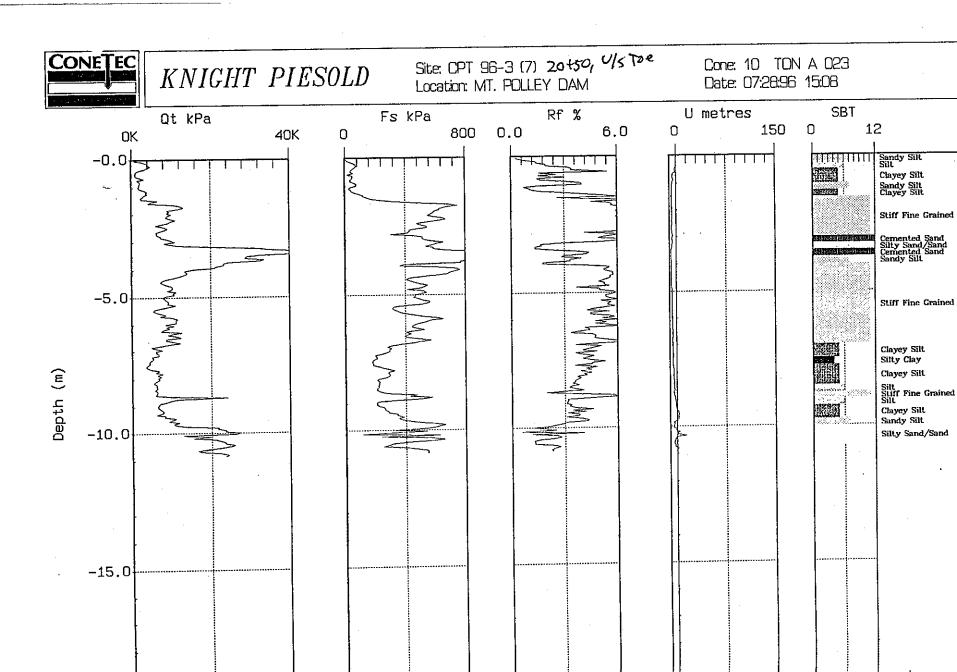
KNIGHT PIESOLD

Depth Inc.: 0.05 (m)

Site: CPT 96-2 (6) Mt50, US toe Location: MT. POLLEY DAM

Cone: 10 TON A 023 Date: 07:28:96 13:48





SBT: Soil Behavior Type (Robertson and Campanella 1988)

Max. Depth: 10.85 (m)
Depth Inc.: 0.05 (m)

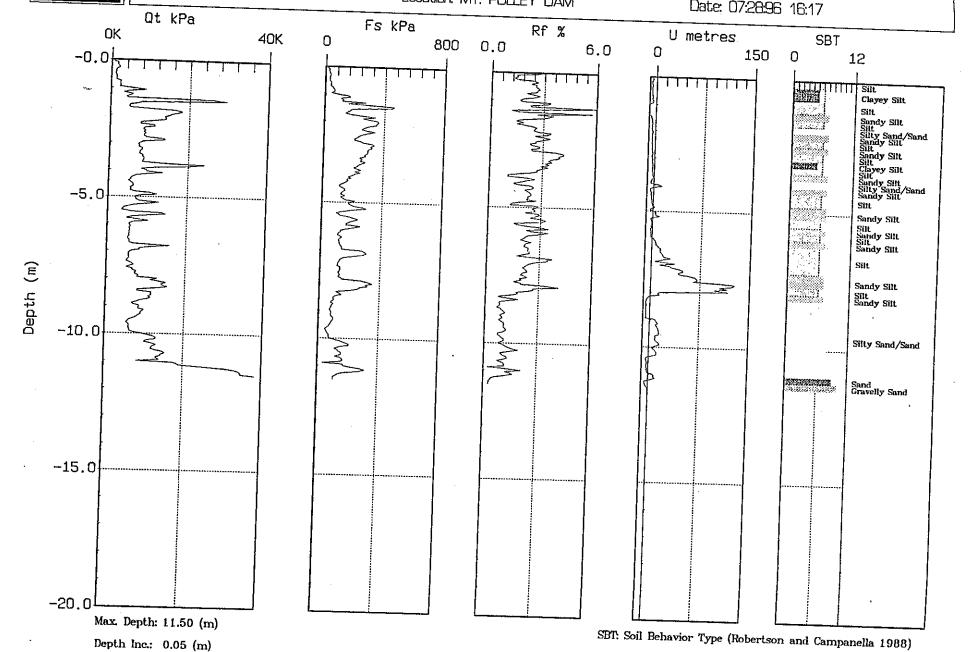
-20.0



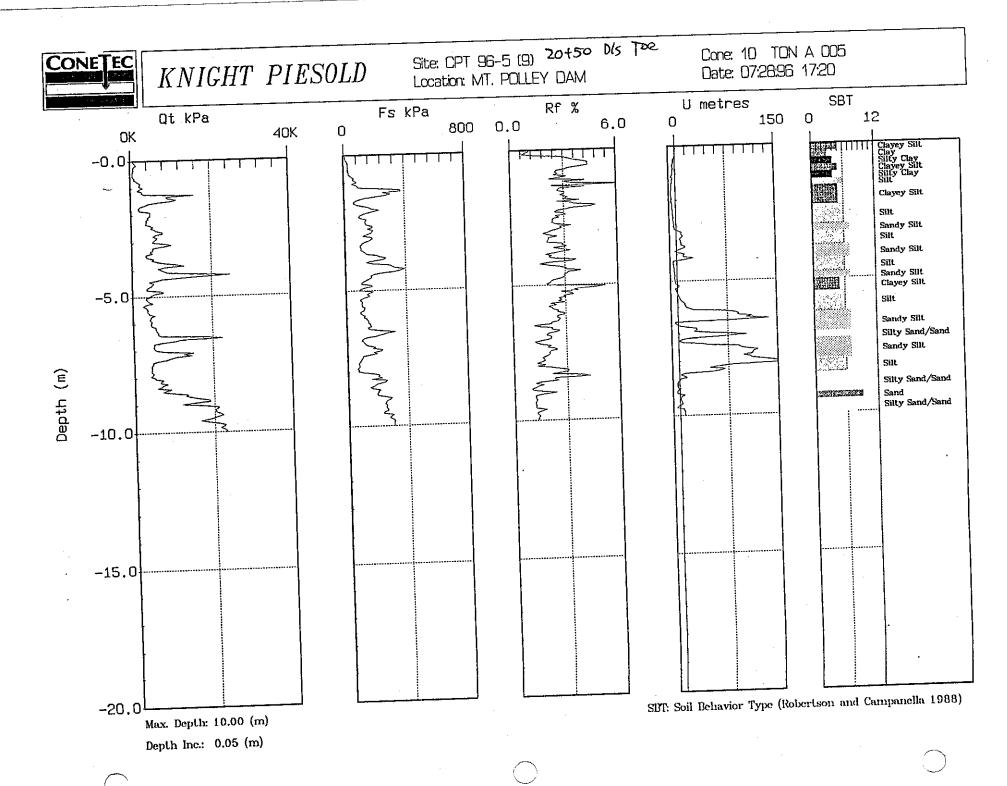
KNIGHT PIESOLD

Site: CPT 96-4 (B) 19450, D/5 Toc Location: MT. POLLEY DAM

Cone: 10 TON A 005 Date: 07:28:96 16:17



SBT: Soil Behavior Type (Robertson and Campanella 1988)



APPENDIX B

TABULAR CPT INTERPRETATIONS

20+00

ConeTec Investigations Ltd. - CPT Interpretation Client: KNIGHT PIESOLD Site: CPT 96-1 (1) Location: MT. POLLEY DAM Cone: 10 TON A 023 Cone: 07:27:96 08:15

er Table (m): _ 0.00

(ft): 0.00

≟r Su Nk	t used:	12.50									 N60 N	 160 Su	(kPa)	k(cm/s)
ауег	Depth	Qt	Fs	Rf SBT	U 1	Jt TStr	ess EStr						104.66	5.0E-06
1	0.12	1310.55	28.20			8.00 7.50					6.6 19.3	38.6	154.01	5.0E-08
ż	0.38	1931.77	84.80			8.00			6.13		11.2		222.92 260.08	5.0E-05 5.0E-07
3	0.62	2797.64 3266.65	36.80 131.80	4.03		18.00	15.62	7.04	8.58	2.00	21.8 21.8	43.6 43.6	347.22	5.0E-06
4	0.88 1.12	4360.36	181.80	4.17	5	18.00	20.12	9.09	11.04	2.00 2.00	28.7	57.3	456.54	
5 6	1.38	5731.42	219.00	3.82	5	18.00		11.14	13.49 15.94	2.00	21.4	42.7	339.33	5.0E-06
7	1.62	4270.71	156.00	3.65	5	18.00	29.12 33.56	13.18 15.17	18.39	2.00	39.7	79.4	314.99	5.0E-08
8	1.88	3970.95	194.60	4.90	3 3	17.50 17.50	37.94	17.09	20.85	2.00	38.4	76.7	303.87	
9	2.12	3836.34	200.60	5.23	5	18.00	42.38	19.08	23.30	2.00	17.0	34.0	268.39	
10	2.38	3397.24	137.20 253.60	4.04 4.26	5	18.00	46.88	21.12	25.75	2.00	29.8	59.5	472.52 683.83	
11	2.62	5953.32 8599.27	315.80	3.67	5	18.00	51.38	23.17	28.20	2.00	43.0 26.5	86.0 51.7	526.16	
12	2.88 3.12	6632.89	205.60	3.10	6	18.00	55.88	25.22	30.66	1.95 1.87	28.6	53.6	0.00) 5.0E-04
13	3.38	8587.49	178.20	2.08	7	18.50	60.44	27.33	33.11 35.56	1.80	23.7	42.7	373.30) 5.0E-06
14 15	3.62	4731.28	188.80	3.99	5	18.00	65.00 69.50	29.44 31.49	38.01	1.74	22.6	39.3	355.26	5 5 0E-06
16	3.88	4510.20	164.20	3.64	5	18.00 18.00	74.00	33.53	40.47	1.69	24.9	42.2	492.95	5 5.0E-05 4 5.0E-05
17	4.12	6235.88	160.40	2.57 3.03	6 6	18.00	78.50	35.58	42.92	1.64	24.9	40.8	491.04 376.3	
18	4.38	6216.44	188.40 189.60	3.96	5	18.00	83.00	37.63	45.37	1.60	23.9	38.2	362.4	
19	4,62	4787.51 4617.74	250.00	5.41	ž	17.50	87.44	39.61	47.82	1.55	46.2 21.5	71.8 32.6	336.0	
20	4.88 5.12	4292.24	143.00	3.33	5	18.00	91.88	41.60	50.28 52.73	1.52 1.48	27.2	40.3	536.2	
21 22	5.38		184.00	2.71	6	18.00	96.38	43.65	55.18	1.45	38.3	55.5	758.3	1 5.0E-05
23	5.62		316.40	3.30	6	18.00	100.88 105.38	45.69 47.74	57.63	1.42	31.3	44.4	492.9	
24	5.88	6267.72		3.52	5	18.00 20.50	110.19	50.10	60.09		19.0	26.3	0.0	
25	6.12	6663.82			11 5	18.00	115.00	52.46	62.54	1.35	40.0			0 5.0E-06 00 5.0E-04
26			298.00 7 269.6		7	18.50		54.57	64.9					00 5.0E-04 00 5.0E-03
27			4 131.8		8	19.00	124.25	56.81	67.4	4 1.30				00 5.0E-02
28			0 133.4		9	19.50	129.06	59.17	69.9	0 1.27 5 1.25		_	_	00 5.0E-02
29 30				0.76	9	19.50		61.59	72.3 74.8				1 0.	.00 5.0E-02
20 71			8 183.0	0.97	9	19.50		64.01 66.43				•	1 0.	.00 5.0E-02
	7.88		8 164-0		9	19.50 19.50						6 44-		.00 5.0E-02
	8.12	18809.			9 9	19.50				6 1.1				.00 5.0E-02
- 34					8	19.00			84.6	51 1.1				.00 5.0E-03 .00 5.0E-02
35	8.6				9	19.50	163.06	76.00	87.0		2 38.		• •	.00 5.0E-02
36					ģ	19.5	167.94							00 5.0E-02
37					9			80.84	91.9	yr 1.U	12 27		• •	

ConeTec Investigations Ltd. - CPT Interpretation Client: KNIGHT PIESOLD Site: PRW 96-3 (2)
Location: MT. POLLEY DAM Cone: 10 TON A 023

(ft): 0.00

Table (m): 0.00

19130 4

ConeTec Investigations Ltd. - CPT Interpretation Client: KNIGHT PIESOLD Site: PRW 96-4 (3) Location: MT. POLLEY DAM Cone: 10 TON A 023 07:27:96 13:51

(ft): 5.74 or Table (m): 1.75

ou Nk	t used:	12.50												
Layer	Depth	Qt	Fs	Rf S	вт	u Wt TS	tress ESt	ress H	Pres	Cn	N60 N		(kPa) k	
1	0.12	752.38	10.00	1.33	5	18.00		2.25			3_8,	7.5 (.0E-06 5.0E-06
, ,	0.38	1086.87	19.00	1.75	5	18.00	6.75	6.75	0.00	2.00	5.4 7.8	15.6		5.0E-07
2 3 4	0.62	1170.65	35.80	3.06	4	18.00		11.25	0.00	2.00	14.9	29.8	177.78	5.0E-07
4	0.88	2238.00	86.00	3.84	4	18.00		15.75	0.00 0.00	2.00	9.0	17.9	177.59	5.0E-05
5	1.12	2240.13	44.00	1.96	6	18.00		20.25 24.75	0.00	1.97	9.6	18.9	151.70	5.0E-06
6	1.38	1921.05	45.20	2.35	5	18.00	24.75 29.25	29.25	0.00	1.81	11.9			5.0E-06
7	1.62	2371.95	69.80	2.94	5_	18.00 18.00	33.75	32.52	1.23	1.72	17.1	29.3	339.12	5.0E-05
8	1.88	4272.75	128.40	3.01	, 6		38.44	34.76			48.6	80.6	0.00	5.0E-02
9	2.12	24284.02		0,84			43.31	37.18			45.6	73.2	0.00	5.0E-02
10	2.38	22803.97		0.36	10		48.25	39.67	8.58	1.55	37.9	58.9	0.00	5.0E+00
11	2.62	22755.32		0.39	9	19.50	53.19	42.15	11.04	1.51	40.8	61.5	0.00	5.0E-02
12	2.88	20393.23	76.80	2.11	6	18.00	57.88	44.39	13.49	1.47	14.6			5.0E-05
13	3.12	3648.38 5726.08	109.40	1.91	7		62.44	46.50	15.94	1.44	19.1	27.4	0.00	5.0E-04
14	3.38 3.62	8123.18	256.80	3.16	. 6	*=:	67.00	48.61	18.39	1.40	32.5	45.6	644.49	5.0E-05
15	3.88	22003.86				9 19.50		50.84		1.37	44.0		0.00	5.0E-02
16	4.12	25626.96				9 19.50	76.56	53.26		1.34			0.00	5.0E-02
17 18	4.38	30330.53				9 19.50		55.69					0.00	5.0E-02 5.0E-02
19	4.62	23951.85				9 19.50		58.11					0.00	5.0E-02
20	4.88	17053.27	320.80		8	8 19.0		60.47	30.66				0.00	5.0E-04
21	5.12	8637.88	205.00	2.37	7		95.81	62.70	33.11	1.24	28.8	35.6 45.1	0.00	
22	5.38	11146.26	191.80	1.7		7 18.5								
23	5.62	22028.03	195.40	0.8		9 19.5		67.17					0.00	
24	5.88	22278.29				9 19.5	110.06						955.06	
25	6.12	12052.97	7 365.60			6 18.0					33.8	38.5	531.25	5.0E-06
26	6.38	6759.84	245.00	3.62		18.00	119.25	73.88	45.37 47.82	1.12	25.9	29.1	404.54	5.0E-06
27	6.62	5180.53	199.80	3.86	5	18.00	123.75	75.93 77.97	50.28	1.11	25.9	28.7	404.73	5.0E-06
28	6.88	5187.37	171.60	3.31	. 5	18.00	128.25	80.02	52.73		31.8	34.7	497.46	5.0E-06
29	7.12	6351.05	223.40	3.52				82.07	55.18		20.8	22.5	404.96	5.0E-05
30	7.38	5199.19	149.60	2.88				84.12	57.63		22.5	24.0	438.34	5.0E-05
31	7.62	5621.04	150.80	2.68				86.16	60.09		23.6	24.9	460.23	5.0E-05
	7.88	5899.14	196.40	3.33				88.52			23.2	24.2	0.00	0.0E+00
	8.12	8130.81		4.39 2.2		7 18.5				9 1.03		7 40.8		
24	8.38	11923.84				8 19.0								
35	8.62	14088.78 15319.8				8 19.0				0 1.0				
36	8.88					8 19 0			8 72.3					
37	9.12 9.38					8 19.0		3 100.0	7 74.8					
38	9.58					8 19 0			7 77.2					
39 40	9.88					8 19.0		3 104.6					0.00	
40	10.12					8 19.0		2 106.9	7 82.1	6 0.9	5 61.	8 58.5	0.00	J 5.UE-U3
41	10.12	5412101	~ ~~~.~		. –	, •								

18100

ConeTec Investigations Ltd. - CPT Interpretation Client: KNIGHT PIESOLD Site: PRW 96-2 (4) Location: MT. POLLEY DAM Cone: 10 TON A 023 Pate: 07:27:96 16:27

(ft): 0.00

2400 4

Table (m): 0.00

-ù aK	t used:	12.50			,										
ayer	Depth	Qt	Fs	Rf SB	T U	Wt T	Stress ES	tress	HPres	Cn	N60	N160 s	u (kPa)	k(cm/s)	
1	0.12	2489.82	28.20	1.13	6	18.00	2.25	1.02	1.23	2.00	10.0	19.9	400 04		
2		1979.33	45.00	2.27	6	18.00	6.75	3.07	3.68	2.00	7.9	15.8	199.01	5.0E-05	
3	0.62	4303.89	80.40	1.87	7	18.50	11.31	5.18	6.13	2.00	14.3	28.7	157.81	5.0E-05	
4	0.88	5826.88	148.00	2.54		18.00	15.88	7.29	8.58	2.00	23.3	46.6	0.00	5.0E-04	
5	1.12	5043.68	149.80	2.97	6	18.00		9.34	11.04	2.00	20.2	40.3	464.88 401.86		
6	1.38	6890.27	242.60	3.52		18.00	24.88	11.39	13.49	2.00	34.5	68.9	549.23		
7	1.62	9616.74	428.00	4.45	11	20.50	29.69	13.75	15.94		27.5	55.0	0.00	5.0E-06 0.0E+00	
8	1.88	11674.06	590.00	5.05		20.50		16.42	18.39	2.00	33.4	66.7	0.00	0.0E+00 0.0E+00	
9		12861.01		5.64	11	20.50	39.94	19.09		2.00	36.7	7 73.5			
10	2.38	20914.98		3.34		19.00	44.88		23.30	2.00	104.	209.1	0.00	0.0E+00	
11	2.62	14319.71	561.00	3.92	12	19.00		23.87	25.75	2.00	71.6	143.2			
. 12	2.88	16307.56	351.20	2.15	7	18,50	54.31	26.11	28.20	1.92	54.4	104.1		5.0E-04	
13	3.12	4958.75	117.40	2.37	6	18.00	58.88	28.22	30.66	1.84	19.8	36.5	391.99	5.0E-05	
14	3.38	5279.07	236.40	4.48	5	18.00	63.38	30.27	33.11	1.78	26.4	47.0	417.26	5.0E-05	
15	3.62		248.60	4.32	5	18.00	67.88	32.31	35.56	1.72	28.8	49.5	454.62		
16	3.88	4993.30	219.80	4.40	5	18.00	72.38	34.36	38.01	1.67	25.0	41.7			
17		5189.90	193.20	3.72	5	18.00	76.88	36.41	40.47	1.62	25.9	42.1	409.04	5.0E-06	
18		7986.63	251.20	3.15	6	18.00	81.38	38.46	42.92	1.58	31.9	50.4	632.42	5.0E-05	
19			288.80	3.64	5	18.00	85.88	40.50	45.37	1.54	39.6	61.0	627.36	5.0E-06	
20	4.88	8550.85	365.40	4.27	5	18.00	90.38	42.55	47.82	1.50	42.8	64.1	676.84	5.0E-06	
21	5.12	8669.96	332.60	3.84	5	18.00 18.00	94.88	44.60	50.28	1.47	43.3 46.7	63.5	686.01	5.0E-06	
22	5.38	11677.58	443.20	3.80	6	18.00	99.38	46.65	52.73	1.43	46.7	66.9	926.26	5.0E-05	
23	5.62	10557.94	414.20	3.92	5	18.00	103.88	48.69	55.18		52.8	74.0	836.33		
24	5.88	11457.12	484.60	4.23	11			51.05	57.63	1.37	32.7	44.8	0.00		
25	6.12	11538.37	416.40	3.61	6	18.00	113.50	53.41		1.34	46.2	61.8	913.99		
26	6.38	13159.21	472.60	3.59	6	18.00	118.00	55.46	62.54	1.31	52.6	69.2	1043.3		;
27	6.62	13169.97	511.00	3.88	12	19.00	122.62	57.63		1.29	65.8	84.9	0.00		,
28	6.88	14263.35	672.40	4.71	11	20.50	127.56	60.12	67.44	1.26	40.8	51 4	ስ ሰና	0.05+00	
29	7.12	18233.88	545.60	2.99	. 7	18.50	132.44	62.54	69.90	1.24	60.8	75.2	0.00	5.0E-04 1.0E-07 1.0E-07 5.0E-05	
30		198.53	2.00 1	.01 1	1 12 1 12	2.50 1	36.31 6 39.44 6	3.96	72.35	1.22	1.0	1.2	4.98	1.05-07	~ 1
31		198.53	2.00 1	.01_ 1	1 12	2,50 1	39.44 <i>6</i>	54.64	74.80	1.22	1.0	1.2	4.73	1.0E-07 <	13. 7
C >			285.60	3.24	D	18.00	143.25	66.00	77.25	1.20	35.3	42.5	694.26	5.0E-05	19:5
(./	0.12	5870.77	134.60	2.29	6	18.00	147.75	68.04	79.71	1.19	23.5	27.9	457.84	5.0E-05	
35				3.39	5 5 5	18.00	152.25	70.09	82.16	1.17	31.4	36.7	490.06	5.0E-06	
36			297.60	3.89	5	18.00	156.75	72.14	84.61	1.15	38.2	44.1	599.40	5.0E-06	
37	8.88	8679.92	353.40	4.07	_5	18.00	161.25	74.19	87.06	1.14	43.4	49.3	681.49	5.0E-06 ·	
38	9.12	5619.57	0J.4U	1.48	<u> </u>	8.50	165.81 170.44	76.30	89.52		18.7	21.0		5.0E-04	
39		5707.20 5477.40	77.80	1.36	<i>(</i> 1	8.50	170.44	78.47	91.97	1.10	19.0	21.0	0.00	5.0E-04	
40	7.02	5677.40	137.00	2.38	6_	18.00	175.00 179.56	80.58	94.42	1.09	22.7	24.8	440.19	5.0E-05	
		11321.03	407.00	2.54	7	18.50	179.56	82.69	96.87	1.08	37.7	40.6	0.00	5.0E-04	
		6076.15	174.00	_2.8]	_ 6	-18-00	184.12	84.80	99.33	_1.06_	24.3	25.8	471.36	5.0E-05	
43	10.30	6500.81 7043.41	100.40	2.10	7	18.50	188.69		101.78	1.05	21.7	22.7	0.00	5.0E-04	
44	10.88	1043.41	4U2.8U	2.92	6	18.00	193.25	89.02	104.23	1.04	28.2 30.3	29.2	548.01	5.0E-05	
		10602.73	494.00	4.66	11	20.50	198.06	91.38	106.68	1.02	30.3	31.0	0.00		
		16470.34		2.94	7	18.50	202.94 207.62	93.80		1.01	54.9 50.7	55.5	0.00	5.0E-04	
		20282.48 17529.17	351.40	1.73	ర	19.00	207.62	96.04		1.00	50.7	50.6	0.00		
71	11.02	11267.11	202.20	2.20	1	18.50	212.31	98.27	114.04	0.99	58.4	57.7	0.00		

ConeTec Investigations Ltd. - CPT Interpretation
Client: KNIGHT PIESOLD
Site: PRW 96-2 (4A)
Location: MT. POLLEY DAM
Cone: 10 TON A 023
Tate: 07:28:96 08:43

ete: r Table (m): 0.00 Su Nkt used: 12.50 (ft): 0.00

SUNK	t usec:	:2.50														
Layer	Depth	Qt	Fs	Rf S	ВТ	U Wt	TStress	EStre	ss HF	res	Cn	N60	N160 St	(kPa)	k(cm/s)	
1 2 3 4 5	0.12 0.38 0.62 0.88 1.12 1.38	10.00 10.00 10.00 10.00	0.00 0.00 0.00 0.00	0.00 0.00 0.00 0.00 0.00 0.00	0 0 0 0 0	19.50 19.50 19.50 19.50 19.50	7.31 12.19 17.06	3. 6. 8.	63 3 06 6 48 8 90 1	3.68 5.13 3.58 1.04 3.49	2.00 2.00 2.00 2.00 2.00 2.00	0.1 0.1 0.1 0.1 0.1 0.1	0.2 0.2 0.2 0.2 0.2 0.2	0.00 0.00 0.00 0.00 0.00	0.0E+00 0.0E+00 0.0E+00 0.0E+00 0.0E+00 0.0E+00	
7 8 9 10	1.62 1.88 2.12 2.38	10.00 10.00 10.00	0.00 0.00 0.00	0.00 0.00 0.00 0.00	0 0 0	19.50 19.50 19.50 19.50	36.56 41.44 46.31	18. 20. 23.	.17 18 .59 29 .01 23	5.94 8.39 0.85 3.30	2.00 2.00 2.00 2.00	0.1 0.1 0.1	0.2 0.2 0.2 0.2	0.00 0.00 0.00	0.0E+00 0.0E+00 0.0E+00 0.0E+00	
11 12 13 14	2.62 2.88 3.12 3.38	10.00 10.00 10.00 10.00 10.00	0.00 0.00 0.00	0.00 0.00 0.00 0.00 0.00	0 0 0	19.50 19.50 19.50 19.50 19.50	56.06 60.94 65.81	27. 30.	.86 2: .28 3 .70 3	5.75 8.20 0.66 3.11 5.56	1.94 1.85 1.78 1.71 1.65	0.1 0.1 0.1 0.1 0.1	0.2 0.2 0.2 0.2 0.2	0.00 0.00 0.00 0.00	0.0E+00 0.0E+00 0.0E+00 0.0E+00 0.0E+00	Ś
15 16 17 18 19	3.62 3.88 4.12 4.38 4.62	10.00 10.00 10.00 10.00	0.00 0.00 0.00 0.00	0.00 0.00 0.00 0.00	0 0 0	19.50 19.50 19.50 19.50	75.58 80.44 85.3 90.19	37 4 39 1 42 9 44	.55 3 .97 4 .39 4 .82 4	8.01 0.47 2.92 5.37	1.60 1.55 1.50 1.46	0.1 0.1 0.1 0.1	0.2 0.2 0.2 0.1	0.00 0.00 0.00 0.00	0.0E+00 0.0E+00 0.0E+00 0.0E+00	<u>(</u>
20 21 22 23 24	4.88 5.12 5.38 5.62 5.88	10.00 10.00 10.00 10.00 10.00	0.00 0.00 0.00	0.00 0.00 0.00 0.00	0 0 0 0	19.50 19.50 19.50 19.50	99.94 104.8 109.6	4 49 1 52 9 54	.66 5 .08 5 .51 5 .93 5	7.82 0.28 2.73 5.18 7.63	1.42 1.39 1.36 1.33 1.30	0.1 0.1 0.1 0.1 0.1	0.1 0.1 0.1 0.1 0.1	0.00 0.00 0.00 0.00	0.0E+00 0.0E+00 0.0E+00 0.0E+00	
25 26 27 28 29	6.12 6.38 6.62 6.88 7.12	10.00 10.00 10.00 10.00 10.00	0.00 0.00 0.00 0.00	0.00 0.00 0.00 0.00	0 0 0	19.50 19.50 19.50 19.50	119.4 124.3 129.1 134.0 138.9	4 59 1 61 9 64 6 66 4 69	.77 6 .20 6 .62 6	0.09 2.54 4.99 7.44 9.90	1.27 1.25 1.22 1.20 1.18	0.1 0.1 0.1 0.1	0.1 0.1 0.1 0.1	0.00 0.00 0.00 0.00 0.00	0.0E+00 0.0E+00 0.0E+00 0.0E+00 0.0E+00 0.0E+00	j
30 31 34	7.38 7.62 7.88 8.12 8.38	10.00 2554.19 7509.14 5613.22 8301.00	0.00 59.60 123.00 118.20 204.20	0.00 2.33 1.64 2.11 2.46	, 7	7 18. 7 18. 7 18.	00 148. .50 153 .50 157 .50 162	50 7 .06 .69 .31	3.70 75.81 77.98 80.15	2.35 74.80 77.25 79.71 82.16	1.11	18. 27.	0 28.1 7 20.7 7 30.2	192.46 0.0 0.0 0.0 728.0	5.0E-05 0 5.0E-0 0 5.0E-0 0 5.0E-0	i)4)4)4
35 36 37 38 39	8.62 8.88 9.12 9.38 9.62	9266.89 7465.28 5977.13 6100.89 11840.43		3.97 3.21 3.05 2.63 0 2.9	6 6 6	18 18 18 6 18	.00 171 .00 175 .00 180 3.00 18	.38 .88 .38 34.88	82.26 84.31 86.36 88.41 90.45	84.61 87.06 89.52 91.97	5 1.07 2 1.05 7 1.04 42 1.0	29. 23. 24. 3 47	9 31.8 9 25.2 4 25.4 .4 48.	583.5 464.1 473.6 7 932.	1 5.0E-0 0 5.0E-0 64 5.0E-0 .44 5.0E	05 05 05 -05
40 41 42 43 44	9.88 10.12 10.38 10.62 10.88	7665.37 6338.70 6774.77 9483.44 10910.60	165.20 194.00 349.60	2.61 2.86 3.69	6	5 18 5 18 5 18	.00 193 .00 198 .00 202	.88 .38 .88	92.50 94.55 96.60 98.64 100.75	96.87 99.33 101.78 104.23 106.6	3 1.01 8 1.00 3 0.99 68 0.9	25. 27. 37. 8 36	4 25.5 1 27.0 9 37.4 .4 35.	491.5 526.1 742.4 5 0	59 5.0E- 11 5.0E- 15 5.0E- .00 5.0E	05 05 05 -04
45 46 47 48 49	11.12 11.38 11.62 11.88 12.12	18099.96 17105.63 21928.24 18886.70 19129.71	292.44 347.44 290.04 262.04	0 1.6 0 2.0 0 1.3 0 1.3	52)3 52 59	8 1 7 1 8 1 8 1	9.00 21 8.50 21 9.00 22 9.00 22 9.00 23	12.12 16.81 21.50 26.25 31.00	102.99 105.22 107.46 109.76 112.05	111.! 114.! 116 118.!	59 0.9 04 0.9 49 0.9 95 0.9	95 57 94 54 93 47 92 47	.2 43. .0 54. .8 51. .2 44.	4 0 8 0 1 0 2 0	.00 5.0E .00 5.0E .00 5.0E .00 5.0E	-04 -03 -03 -03
50 51 52 53 54	12.38	21249.07 20972.90 21011.12 21631.48 22406.21	323.8 309.4 307.8 297.4	0 1.5 0 1.4 0 1.4 0 1.3	18 16 37	8 1 8 1 8 1	9.00 24 9.00 24 9.00 25	35.75 40.50 45.25 50.00 54.75	114.35 116.65 118.95 121.24 123.54	123. 126. 128.	85 0.9 30 0.9 76 0.9	91 52 90 52 89 54	3.1 48. 2.4 47. 2.5 47. 3.1 48. 3.0 49.	5 0 1 0 1 0 3 0	.00 5.0E .00 5.0E .00 5.0E .00 5.0E	-03 -03 -03 -03
55 56 57 58	13.62 13.88 14.12 14.38	20394.74 24320.19 25718.23 21931.04	337.0 313.0 357.2 281.2	0 1.6 0 1.3 0 1.3	55 29 39 28	8 1 8 1 8 1	9.00 20 9.00 20 9.00 20 9.00 2	59.50 64.25 69.00 73.75 78.50	125.84 128.14 130.43 132.73 135.03	133. 136. 138. 141.	11 0. 57 0. 02 0.	86 64 86 64 85 54	1.0 44. 0.8 52. 4.3 55. 4.8 46. 1.7 43.	6 0 1 0 6 0	.00 5.0E .00 5.0E .00 5.0E .00 5.0E	-03 -03 -03
59 60 61 62 63	14.62 14.88 15.12 15.38 15.62	20694.76 20029.00 19246.34 19898.37 22273.60	284.8 272.2 268.4	0 1.4 0 1.4 0 1.3	42 41 35	8 1 8 1 8 1	9.00 2 9.00 2 9.00 2	78.30 83.25 88.00 92.75 97.50	137.33 139.62 141.92 144.22	145. 148. 150.	92 0. 38 0. 83 0.	84 51 83 41 82 41	9.7 40. 5.7 45.	8 0 9 0	.00 5.00 1.00 5.00 1.00 5.00	-03 -03

ConeTec Investigations Ltd. - CPT Interpretation

Location: Cone:

6.88

7.12

30

11301.36

25891.01

29759.61

`te:

KNIGHT PIESOLD PRW 96-1 (5) MT. POLLEY DAM 10 TON A 023

07:28:96 10:28

Table (m): 0.00 (Su nkt used: 12.50

(ft): 0.00

4.16

3.07

1.37

1.87

347.00 353.80

557.80

18.00

18.00

19.00

19.00

6

119.88

124.38

129.00

133.75

54.88

56.93

59.10

61.40

64.99

67.44 69.90

72.35

43.0

45.2 64.7

74.4

1.30 1.27

56.8

58.6

82.4

92.9

678.65

894.16

0.00

0.00

Layer Depth Qt Fs Rf SBT U Wt TStress EStress HPres Cn N60 N160 Su (kPa) k(cm/s) Age. nehrer at to pri ont one loricoo raticoo mico on uno utoo on fulab virinkob 0.12 11951.69 72.40 0.61 9 19.50 0.38 5533.44 135.40 2.45 6 18.00 0.62 2433.28 69.80 2.87 5 18.00 0.88 3867.24 122.20 3.16 5 18.00 1.21 1.23 2.00 23.9 3.45 3.68 2.00 22.1 5.49 6.13 2.00 12.2 7.54 8.58 2.00 19.3 9.59 11.04 2.00 29.7 11.64 13.49 2.00 28.4 2.44 7.12 11.62 5533.44 135.40 5533.44 135.40 2433.28 69.80 3867.24 122.20 7420.22 205.40 7092.17 200.20 6540.42 177.00 6038.54 174.80 47.8 44.3 24.3 0.00 5.0E-02 22.1 12.2 442.11 5.0E-05 3.16 2.77 193.73 5.0E-06 18.00 16.12 19.3 29.7 28.4 26.2 1.12 38.7 5.0E-06 5.0E-05 308.09 6 18.00 20.62 1.38 2.82 59.4 591.97 25.12 29.62 6 18.00 13.49 15.94 1.62 2.71 2.89 565.36 56.7 5.0E-05 5.0E-05 6 18.00 13.68 2.00 89 1.88 52.3 520.86 18.00 18.50 6 7 34.12 15.73 18.39 480.35 2.12 8552.01 189.00 24.2 48.3 2.21 5.0E-05 38.69 20.85 23.30 17.84 2.00 10 2.38 7803.56 258.00 28.5 57.0 3.31 6 0.00 5.0E-04 18.00 43.25 47.75 52.25 19.95 4976.94 8240.75 11 2.62 2.88 31.2 62.4 39.8 116.20 620.82 2.33 6 5.0E-05 22.00 24.05 18.00 12 25.75 2.00 19.9 262.20 3.18 394.34 5.0E-05 18.00 5825.66 4227.59 28.20 2.00 1.92 13 655.08 3.12 156.60 2.69 33.0 65.8 56.75 61.25 65.75 5.0E-05 6 18.00 26.09 30.66 14 3.38 118.00 132.40 23.3 461.51 44.6 2.79 18.00 5.0E-05 33.11 35.56 28.14 15 1.84 3.62 5763.61 18.00 18.00 18.00 16.9 31.2 2.30 6 333.31 5.0E-05 41.1 27.1 25.8 30.19 1.78 16 3.88 3937.28 23.1 116.80 2.97 70.25 74.75 455.83 5.0E-05 32.24 38.01 84.40 3860.89 1.72 15.7 4.12 309.36 2.19 6 34.28 36.33 38.38 5.0E-05 3793.95 40.47 15.4 15.2 1.67 4.38 82.40 5.0E-05 2.17 302.89 79.25 83.75 6 18.00 42.92 45.37 1.62 4.62 3386.36 90.40 5.0E-05 5.0E-05 24.6 297.18 2.67 6 18.00 8647.71 231.00 6723.94 204.60 9048.37 269.20 1.58 20 4.88 2.67 13.5 21.4 264.21 88.25 92.75 97.25 6 18.00 40.43 47.82 1.54 684.76 530.50 5.12 34.6 53.2 3.04 5.0E-05 18.00 50.28 52.73 22 42.47 40.4 53.1 71.9 1.50 5.38 26.9 2.98 15058.41 345.40 8564.16 344.20 7321.29 269.60 18.00 5.0E-05 23 44.52 1.47 5.62 36.2 716.09 2.29 18.50 101.81 5.0E-05 46.63 55.18 57.63 5.88 1.43 50.2 18.00 106.38 4.02 0.00 5.0E-04 48.74 1.40 25 6.12 42.8 676.62 60.0 5.0E-06 3.68 50.79 18.00 110.88 60.09 1.37 6.38 36.6 8272.92 254.40 50.3 576.83 5.0E-06 3.08 18.00 115.38 52.84 1.35 27 62.54 6.62 8603.04 357.80 33.1 44.6 652.60

22+80

5.0E-05

5.0E-06

5.0E-05 5.0E-03

5.0E-03

19+50 Us toe

ConeTec Investigations Ltd. - CPT Interpretation
Client: KNIGHT PIESOLD
Site: CPT 96-2 (6)
Location: MT. POLLEY DAM
Cone: 10 TON A 023
Pate: 07:28:96 13:48

ı	Nkt	used:	7	2	•	5	Ų	

Cone:		10 10 07:28:	96 13:4	8										
er, su Nk	Table (t used:	m): 0.00 12.50	(f	t): 0.0						 Cn	 и60	 N160 Su	(kPa)	 k(cm/s)
Layer	Depth	Qt F	s	Rf SBT	U	Wt TSt	ess EStr							5.0E-04
Layer 1 2 3 4 5 6 7 8 9 11 12 13 14 5 6 7 8 9 11 12 13 14 5 6 7 8 9 21 22 23 22 5 26 7 28 9 3 3 1 4 3 5	0.12 0.38 0.62 0.88 1.38 1.68 2.12 2.88 2.12 2.88 3.38 4.62 4.38 4.62 4.38 5.62 5.62 6.88 7.13 7.62 7.88 8.12	2554.30 2585.48 1658.62 2537.35 2772.81 3398.94 4205.90 5264.96 5018.57 6140.30 17298.53 34518.41 26889.72 21606.39 29328.63 25634.70 11217.15 7993.18 6192.68 6565.03 6434.01 6989.60 5300.37 5377.40 9421.03 5837.81 6509.17 7551.41 5927.42 27754.07 17728.99 27772.23 20576.17	55.80 56.20 70.20 84.60 98.80 106.80 1154.00 1192.80 237.00 3312.40 1190.80 424.80 424.80 424.80 424.80 225.80 2275.80 2212.20 2292.00 130.20 149.40 191.60 93.80	1.01 2.14 2.18 2.77 3.05 2.91 2.54 2.76 3.07 3.14 1.37 0.89 0.65 0.65 0.65 4.19 4.19 4.10 3.88 3.95 2.23 4.19 2.23 3.23 1.63 7.70 1.47 2.18 2.23 1.47 2.23 1.47 2.23 1.47 2.23 1.47 2.23 1.47 2.23 1.47 2.23 1.47 2.23 1.47 2.23 1.47 2.23 1.47 2.23 1.47 2.23 2.23 2.23 2.23 2.23 2.23 2.23 2.2	111111 7655555 6666 8999981 1545555566666667 8888	8.50 8.00 8.00 8.00 18.00 18.00 18.00 19.50 19.50 19.50 19.50 19.50 18.00 18.00 18.00 18.00 18.00 18.00 18.00 18.00 18.00 18.00 18.00	2.31 6.88 11.38 15.88 20.38 24.88 29.38 33.88 33.88 47.50 52.31 57.19 62.06 66.94 71.75 76.69 90.50 99.50 99.50 104.00 117.50 122.00 117.50 144.75 0 154.25 0 159.00	7.29 7.29 7.29 9.34 13.43 17.53 19.58 21.75 24.13 26.53 33.74 36.22 38.58 40.63 42.68 44.72 46.72 46.82 52.91 54.96 67.5 67.5 67.5 67.5 74.3	1.23 3.68 6.13 8.104 13.49 15.94 18.39 20.35 25.75 28.20 30.61 35.56 38.01 42.92 45.37 47.82 50.28 52.73 55.18 57.63 60.99 67.35 74.80 77.27 99.884.0	2.00 2.00 2.00 2.00 2.00 2.00 2.00 2.00	8.5 10.3 12.7 13.9 17.0 16.8 20.1 20.1 24.6 43. 40.3 24.6 32. 24.6 32. 24.6 32. 25. 26. 27. 28. 29. 29. 29. 29. 29. 29. 29. 29. 29. 29	17.0 20.7 16.6 27.7 34.0 33.6 40.1 49.1 20.8 49.1 49.1 20.8 40.1 49.1 20.8 40.1 40.1 40.1 40.1 40.1 40.1 40.1 40.1	0.00 206.29 131.78 201.72 220.19 269.92 334.12 418.49 398.42 487.79 0.0 0.0 0.0 0.0 632.93 488.53 517.96 507.11 551.2 415.7 421.5 744.6 457.6 420.2 510.6 463.3 0.0 8	5.0E-05 5.0E-05 0.0E-02 0.0E-02 0.0E-02 0.0E-02 0.0E-02 0.0E-03 0.0E-03 0.0E-06 5.0E-06 5.0E-06 5.0E-06 1.5.0E-06 1.5.0E-06 1.5.0E-05 5.0E-05 5.0E-05 5.0E-05 5.0E-05 5.0E-05 5.0E-05 5.0E-05 5.0E-03 0.0E-03 0.0E-03
36 37 38	9.1	2 20419.2	276.8	30 1.36			0 168.59	78.9 81.5				1.0 56 7.0 51		.00 5.0E-03 .00 5.0E-03

ConeTec Investigations Ltd. - CPT Interpretation Client: KNIGHT PIESOLD Site: CPT 96-3 (7)
Location: MT. POLLEY DAM Cone: 10 TON A 023
Pate: 07:28:96 15:08

uls toe 20150

r Table (m): 0.25 su Nkt used: 12.50

(ft): 0.82

30 M	Kt useu:	12.50							•					•
Layer	Depth	Qt	Fs	Rf SE	3T L	JWt TS	tress ES1	ress	HPres	Cn	N60	N160 St	ı (kPa)	k(cm/s)
1	0.12	3278.85	32.80	1.00	7	18.50	2.31	2.31	0.00	2.00	10.9	21.9	0.00	5.0E-04
2	0.38	2168.61	52.60	2.43	6	18.00	6.88	5.65	1.23	2.00	8.7	17.3	172.94	5.0E-05
3	0.62	1706.08	42.40	2.49	5	18.00	11.38	7.70	3.68	2.00	8.5	17.1	135.58	5.0E-06
4	0.88	1966.77	54.20	2.76		18.00	15.88	9.74	6.13	2.00	9.8	19.7	156.07	5.0E-06
5	1.12	3334.83	40.40	1.21		18.50	20.44	11.85	8.58	2.00	11.1	22.2	0.00	5.0E-04
6	1.38	3462.91	128.80	3.72	5	18.00	25.00	13.96	11.04	2.00	17.3	34.6	275.03	5.0E-06
7	1.62	9049.95	494.00	5.46	11	20.50	29.81	16.32	13.49	2.00	25.9	51.7	0.00	0.0E+00
8	1.88	9855.25	622.80	6.32	11	20.50	34.94	19.00	15.94	2.00	28.2	56.3	0.00	0.0E+00
9	2.12	9638.12	606.20	6.29	11	20.50	40.06	21.67	18.39	2.00	27.5	55.1	0.00	0.0E+00
10	2.38	7939.74	529.80	6.67	11	20.50	45.19	24.34	20.85	1.98	22.7	45.0	0.00	0.0E+00
11	2.62	6919.19	457.00	6.60	11	20.50	50.31	27.01	23.30	1.88	19.8	37.2	0.00	0.0E+00
12	2.88	8819.36	462.80	5.25	11	20.50	55.44	29.69	25.75	1.80	25.2	45.3	0.00	0.0E+00
13 14	3.12 3.38	16909.29 40957.82		3.50	12	19.00	60.38	32.17	28.20	1.73	84.5	145.9	0.00	
15	3.62	29676.91	1080.80	1.94 3.64	. 8 . 12	19.00	65.12	34.47	30.66	1.67	102.4		0.00	
16	3.88	20847.34	580.80	2.7 9						1 <u>.6</u> 1	148.	4 239.5		
17	4.12	13243.68		5.46	7 11	18.50 20.50		39.00	35.56	1.57	69.5		0.00	
18	4.38	8419.58	474.60	5.64	11	20.50	79.44 84.56	41.42	38.01	1.52	37.8		0.00	
19	4.62	9666.03	506.60	5.24	11	20.50	89.69	44.10 46.77	40.47	1.47	24.1	35.5	0.00	0.0E+00
20	4.88	9777.71	487.80	4.99	11	20.50	94.81	49.44	42.92 45.37	1.43	27.6	39.5	0.00	0.0E+00
21	5.12	8865.86	498.80	5,63	11	20.50	99.94	52.11	47.82	1.39	27.9	38.9	0.00	0.0E+00
22	5.38	7696.41	420.20	5.46	11	20.50	105.06	54.79	50.28	1.36 1.32	25.3	34.3	0.00	0.0E+00
23	5.62	7928.42	377.60	4.76	11	20.50	110.19	57.46	52.73	1.29	22.0 22.7	29.1 29.2	0.00	0.0E+00
24	5.88	10656.51	576.80	5.41	11		115.31	60.13	55.18	1.26	30.4	38.4	0.00	0.0E+00
25	6.12	8821.33	470.20	5.33	11	20.50	120.44	62.80	57.63	1.23	25.2	31.1	0.00	
26	6.38	11217.29	446.40	3.98	11	20.50	125.56	65.48	60.09	1.21	32.0	38.8	0.00	0.0E+00
27	6.62	9895.79	475.40	4.80	11	20.50	130.69	68.15	62.54	1.19	28.3	33.5	0.00	0.0E+00 0.0E+00
28	6.88	6450.34	339.00	5.26	11	20.50	135.81	70.82	64.99	1.16	18.4	21.4	0.00	0.0E+00
29	7.12	5975.12	257.00	4.30	5	18.00	140.62	73.18	67.44	1.14	29.9	34.2	466.76	5.0E-06
30	7.38	4423.93	191.00	4.32	5	18.00	145.12	75.23	69.90	1.13	22.1	25.0	342.30	5.0E-06
31	7.62	4999.82	223.20	4.46	4 5	18.00	149.62	77.28	72.35	1.11	33.3	37.1	388.02	5.0E-07
	7.88	6908.56	293.40	4.25	5	18.00	154.12	79.32	74.80	1.10	34.5	38.0	540.35	5.0E-06
	8.12		227.40	3.84	5	18.00	158.62	81.37	77.25	1.08	29.6	32.1	460.93	5.0E-06
34	8.38		214.40	3.48	5	18.00	163.12	83.42	79.71	1.07	30.8	33.0	479.14	5.0E-06
35 36	8.62	11578.52		2.87	6			85.47	82.16	1.06	46.3	49.0	912,87	5.0E-05
36 37	8.88		361.60	4.73	11		172.44	87.83	84.61	1.04	21.9	22.8	0.00	0.0E+00
38	9.12 9.38		238.80	3.30	6		177.25	90.19	87.06	1.03	29.0	29.9	565.45	5.0E-05
39	9.62	11250.18	328.00 437.80	3.78	5_	18.00	181.75	92.23	89.52	1.02	43.4	44.2	679.66	5.0E-06
40		23541.62	544.20	3.89	5	18.00	186.25	94.28	91.97	1.01	56.3	56.7	885.11	5.0E-06
41	10.12	19005.78	366.20	2.31 1.93	7	18.50	190.81	96.39	94.42	1.00	78.5	78.2	0.00	5.0E-04
42		23971.17	505.40	2.11	8 8	19.00 19.00	195.50	98.63	96.87	0.99	47.5	46.8	0.00	5.0E-03
43		19587.39	393.60	2.01	8	19.00	200.25	100.92	99.33	0.97	59.9	58.4	0.00	5.0E-03
-10	.0104	17301137	373.00	2.01	٥	17.00	205.00	103.22	101.78	0.96	49.0	47.2	0.00	5.0E-03

ConeTec Investigations Ltd. - CPT Interpretation
Client: KNIGHT PIESOLD
Site: CPT 96-4 (8)
Location: MT. POLLEY DAM
Cone: 10 TON A 005
Date: 07:28:96 16:17

nate:

ter Table (m): 0.00 Su Nkt used: 12.50

0.00 (ft):

au Ni	ct used:	12.50												
Layer	Depth	Qt	Fs	Rf	SBT	U W t TS	tress ESt	ress }	lPres	Cn !	160 N	160 Su	(kPa) k	(cm/s)
1 2	0.12 0.38	1489.64 1140.63	22.20 28.00	1.49	6 5	18.00 18.00	2.25 6.75	1.02	1.23 3.68 6.13	2.00 2.00 2.00	5.7	11.4	90.71	5.0E-05 5.0E-06 5.0E-06
3 4	0.62 0.88	1519.61 2941.49	37.00 61.00	2.43	5 6	18.00 18.00	11.25 15.75	5.12 7.17 9.21	8.58 11.04	2.00				5.0E-05 5.0E-05
5 6	1.12 1.38	5447.08 14293.81		2.41	9	18.00 7 18.50 18.00	20.25 24.81 29.38	11.32	13.49 15.94		47.6 25.4	95.3 50.8	0.00	5.0E-04 5.0E-05
7 8	1.62 1.88	6345.66 16438.79	164.00 296.40	2.58	:0	19.00 19.00 7 18.50	34.00	15.61 17.84	18.39 20.85	2.00	41.1	82.2 80.7	0.00	5.0E-03 5.0E-04
9 10	2.12	12106.73 9422.01	282.80	2.6 3.00 2.53	6 (18.00	43.25 47.81	19.95 22.06	23.30 25.75	2.00	37.7 29.2	75.4 58.5	750.30 0.00	5.0E-05 5.0E-04
11 12	2.62 2.88 3.12	8770.52 10576.06 7701.14	308.40 289.00		2	6 18.00		24.17 26.22	28.20 30.66	1.99	42.3 38.5	84.2 73.6	841.89 611.54	5.0E-05 5.0E-06
13 14 15	3.38 3.62	7908.17 9511.50	257.80 228.00	3.26	6	18.00	61.38	28.27 30.38	33.11 35.56	1.84 1.78	31.6 31.7	58.2 56.3	627.74 0.00	5.0E-05 5.0E-04
16 17	3.88 4.12	14686.47 9118.98	224.00 166.60		i3	8 19.00		32.61 34.85	38.01 40.47	1.71 1.66	36.7 30.4	62.9 50.4	0.00	5.0E-04
18 19	4.38 4.62	5518.55 5799.24	125.80 135.20	2.28	3 6 3 6	18.00 18.00	79.88 84.38	36.96 39.00	42.92 45.37	1.61 1.57	22.1 23.2	35.5 36.4	435.09 457.19	5.0E-05 5.0E-05
20 21	4.88 5.12	8651.35 9005.23	161.00 226.20	1.86 2.5	5 7 1 7	18.50 18.50	88.94 93.56	41.11 43.29	47.82 50.28	1.53 1.49	28.8 30.0	44.0 44.7	0.00	5.0E-04 5.0E-04
22 23	5.38 5.62	4629.19 10147.48		2.89	3	7 18.5	0 102.69	45.40 47.51	52.73 55.18 57.63	1.45 1.42 1.39	18.5 33.8 27.5	26.9 48.0 38.2	362.49 0.00 540.70	5.0E-05 5.0E-04 5.0E-05
24 25	5.88 6.12	6866.04 5607.55	193.20 118.80	2.8 ² 2.17 2.2	2 7	18.50	. 111.81	49.62 51.73 53.84	60.09 62.54	1.36	18.7 21.8	25.4 29.0	0.00	5.0E-04 5.0E-05
26 27	6.38 6.62 6.88	5440.44 5757.40 9883.47	122.20 132.40 273.60	2.3) 6	18.00	120.88	55.88 57.93	64.99 67.44	1.31	23.0 39.5	30.2 50.8	450.92 780.65	5.0E-05 5.0E-05
28 29 30	7.12 7.38	5544.45 5757.49	129.40 123.20	2.3	3 6 4 7	18.00 18.50	129.88	59.98 62.09	69.90 72.35	1,26 1,24	22.2 19.2	28.0 23.8	433.17 0.00	5.0E-05 5.0E-04
31	7.62 7.88	6297.84 8657.99	123.00 244.00	1.95 2.85	5 7 2 <i>6</i>	18.50 18.00	139.06 143.62	64.26 66.37	74.80 77.25	1.22 1.20	21.0 34.6	25.6 41.6	0.00 681.15	5.0E-04 5.0E-05
ز 54	8.12 8.38	13558.92 11141.01	115.80	1.1	04	7 18.5 8 19.0	0 152.88	70.72	82.16	5 1.16	45.2 27.9 27.8	53.5 32.4 31.8		
35 36	8.62 8.88	11115.29 7788.25	76.00	0.78 0.98 0.93	8		157.62 162.38 167.12	73.01 75.31 77.61	84.61 87.06 89.52	1.15 1.13 1.11	19.5 22.2	22.0	0.00	5.0E-03 5.0E-03
37 38 39	9.12 9.38 9.62	8897.54 7601.65 5803.99	82.40 83.00 53.00	1.09	8 8 8	19.00	171.88 176.62	79.91 82.20	91.97 94.42	1.09 1.08	19.0 14.5	20.8 15.7	0.00 0.00	5.0E-03 5.0E-03
40 41	9.88 10.12	6980.76 11896.91	66.40	0.95	8		181.38	84.50 86.80	96.87 99.3	1.06 3 1.05	17.5 29.7	18.6 31.2	0.00	5.0E-03 5.0E-03
42 43	10.38	11820.44 13920.64	133.20) 1.) 1.	13 03	8 19.0 8 19.0	0 195.62	91.39	104.2	3 1.02	34.8	35.6	0.00	5.0E-03
44 45	10.88 11.12	12496.92	248.8	3 1.	12	8 19.0 9 19.5	0 205.19	96.0	5 109.1	4 1.00	44.4	44.4	0.00	5.0E-02
46	11.38	34996.00	114.0	0.	` دد	10 20.0	0 210.12	98.5	+ 111.5	7 0.77	30.3		0.00	, 2102.00

ConeTec Investigations Ltd. - CPT Interpretation
Client: KNIGHT PIESOLD
Site: CPT 96-5 (9)
Location: MT. POLLEY DAM
Cone: 10 TON A 005
Pate: 07:28:96 17:20

r Table (m): 0.00 su Nkt used: 12.50

(ft): 0.00

					. 									•
Layer	Depth	Qt	Fs	Rf S	BT	U Wt	TStress ES	Stress	HPres	Cn	N60	พ160	Su (kPa)	k(cm/s)
1	0.12	1018.61	12,80	1.26	5	18.00	2.25	1.02	4 27	2 00				
Ž	0.38	648.52	25.20	3.89	3	17.50	6.69	3.01	1.23	2.00	5.1	10.2	_81.31	5.0E-06
3	0.62	919.58	30.00	3.26	4	18.00	11.12	4.99	3.68 6.13	2.00	6.5	13.0	51.35	5.0E-08
4	0.88	2398.97		2.71	5	18.00	15.62	7.04	8.58	2.00	6.1	12.3	72.68	5.0E-07
Ś	1.12	2608.19	110.40	4.23	4	18.0		9.09	11.04	2.00	12.0	24.0	190.67	5_0E-06
6	1.38	11168.50		2.84		6 18.				2.00	17.4			5.0E-07
7	1.62	3822.25	115.80	3.03	5			13.18						
8	1.88	3669.27	120.20	3.28	5	18.0		15.23	15.94 18.39	2.00	19.1	38.2		5.0E-06
9	2.12	4111.14	154.80	3.77	5			17.28		2.00	18.3	36.7		5.0E-06
10	2,38	4413.42	121.80	2.76	6	18.0		19.33	20.85	2.00	20.6	41.1		5.0E-06
11	2.62	6640.82	157.60	2.37	6		0 47.12	21.37	23.30	2.00	17.7	35.3		5.0E-05
12	2.88	5762.16	159.40	2.77	6	18.0		23.42	25.75	2.00	26.6	53.1		5.0E-05
13	3.12	7388.25	153.60	2.08	7			25.53	28.20	2.00	23.0	46.1		5.0E-05
14	3.38	4405.62	102,60	2.33	6	18.0		27.64	30.66	1.94	24.6	47.7		5.0E-04
15	3.62	5649.82	163.60	2.90	6	18.0		29.69	33.11	1.86	17.6	32.8		5.0E-05
16	3.88	9208.24	202.20	2.20	7			31.80	35.56	1.80	22.6	40.6		5.0E-05
17	4.12	14828.99	336.40	2.27	• • •	7 18.			38.01	1.74	30.7	53.3		5.0E-04
18	4.38	6998.16	213.80	3.06	6	18.0		36.08	7 40.47 42.92					
19	4.62	4225.43	124.60	2.95	6	18.0		38.13	45.37	1.63	28.0	45.6		5.0E-05
20	4.88	5894.66	127.80	2.17	7	18.5		40.24	47.82	1.58	16.9	26.8		5.0E-05
21	5.12	3951.36	153.20	3.88	5	18.0		42.35		1.54	19.6	30.3		5.0E-04
22	5.38	3287.90	99.80	3.04	5	18.00		44.40	50.28 52.73	1.50	19.8	29.7		_5.0E-06
23	5.62	4359.59	111.20	2.55	6	18.0	0 101.62	46.44	55.18	1.47	16.4	24.1	255.26	5.0E-06
24	5.88	4438.42	103.40	2.33	6	18.0	0 106.12	48.49	57.63	1.44	17.4	25.0		5.0E-05
25	6.12	5236.42	141.00	2.69	6	18.0	0 110.62	50.54		1.41	17.8	25.0		5.0E-05
26	6.38	8829.14	165.20	1.87	7	18.5		52.65	60.09 62.54	1.38	20.9	28.8		5.0E-05
27	6.62	11331.10	222.40		7	18.	50 119.81			1.35	29.4	39.7		5.0E-04
28	6.88	5223.85	107.00	2.05	7	18.5		56.99	67.44					
29	7.12	12072.20	187.40	1.55	ė			59.23	69.90	1.30	17.4	22.6		
30	7.38	6512.86	141.00	2.16	7		0 133.81	61.46	72.35		30.2			
31	7.62	4366.52	80.00	1.83	7		138.44	63.64	74.80	1.25	21.7	27.1		5.0E-04
	7.88	4577.85	69.80	1.52	7	18.50	143.06	65.81	77.25	1.23 1.21	14.6	17.9	0.00	5.0E-04
-(8.12	6739.68	183.00	2.72	6	18.0		67.92	79.71		15.3	18.4	0.00	5.0E-04
34	8.38	7307.73	209.80	2.87	6	18.0		69.97	82.16	1.19	27.0	32.0		5.0E-05
35	8.62	11593.67	178.20	1.54	8		00 156.75	72.14	84.61	1.17	29.2	34.2		5.0E-05
36	8.88	16164.79	220.80		8	19.			87.06		29.0			
37	9.12	20040.56	253.40		8	19.		76.73			40.4			
38	9.38	21070.97			ğ	19.		79.09			50.1			
39	9.62	19505.44	272.40		8	19.0	00 175.88	81.45			42.1			
40		22503.41	300.00		8			83.75			48.8			
		,	200.00		0	17.	00 100.02	03./5	96.87	1.07	56.3	60.2	2 0.00	5.08-03

20150

APPENDIX C

SUMMARY OF PORE PRESSURE DISSIPATION TIMES AND EQUILIBRIUM PORE PRESSURES

Mount Polly Tailings Dam - Likely, BC Dissipation Times and Equilibrium Pore Pressures

Mount Polly Tailings Dam - Likely, BC Dissipation Times and Equilibrium Pore Pressures

Sounding	Depth (m)	Duration (s)	Equilibrium Pore Pressure (m)
CPT 96-1(1)	4.8	945	
	5.8	275	
	6.7	200	8.6
	7.5	375	8.3
	7.7	140	9.1
	8.1	245	9.0
	8.6	200	10,0
	9.05	395	10.6
T 1 1 5	9.5	235	11.0
Total Dissipation Time (s)		3010	

Sounding	Depth (m)	Duration (s)	Equilibrium Pore Pressure (m)
CPT 96-2 (4)	7.25	930	7.4
	11.75	220	11.0
CPT 92-2(4A)	8.75	255	
	9.75	135	
	10.75	205	12.5
	11.75	195	11.25
	12.75	155	12.5
	14.75	210	14.25
	15.85	135	16.00
CPT96-2(6)	7.75	355	7.8
Total Dissipation Time (s)		2795	<u>. </u>

Knight Piésold Ltd.

Sounding	Depth (m)	Duration (s)	Equilibrium Pore Pressure (m)
CPT 96-3(2)	8.25	415	9.0
Total Dissipation Time (s)		415	

Sounding	Depth (m)	Duration (s)	Equilibrium Pore Pressure (m)
CPT 96-4 (3)	10.25	535	8.0
Total Dissipation Time (s)		535	

Sounding	Depth (m)	Duration (s)	Equilibrium Pore Pressure (m)
CPT 96-3 (7)	10.55	80	10.3
CPT 96-4 (8)	6.95	470	
	8.95	425	9.3
	9.95	160	10.5
Total Dissipation Time (s)		1135	

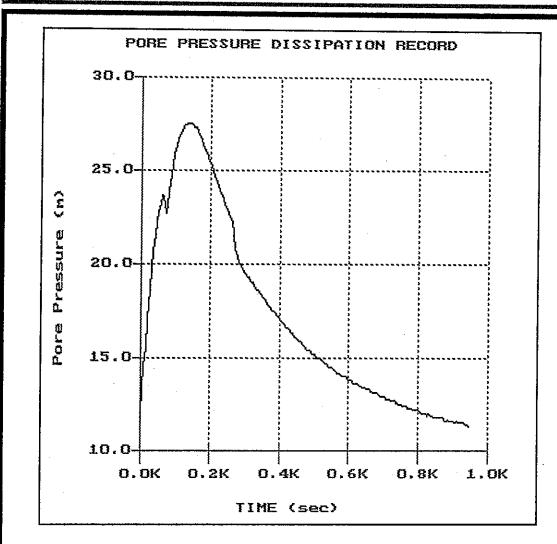
Knight Piésold Ltd.

Sounding	Depth (m)	Duration (s)	Equilibrium Pore Pressure (m)
CPT 96-5 (9)	5.10	595	6.0
	8.40	235	
	10.00	275	11.35
Total Dissipation Time (s)		1105	

APPENDIX D PORE PRESSURE DISSIPATION PLOTS

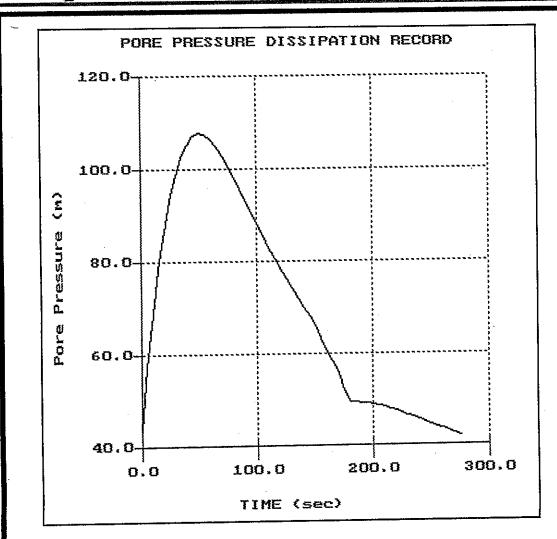
Hole:CPT 96-1 (1) Location:MT. POLLEY DAM

Cone:10 TON A 023 Date:07:27:96 08:15



File: 163CPO1.PPD Depth (m): 4.80 (ft): 15.75 Duration: 945.0s

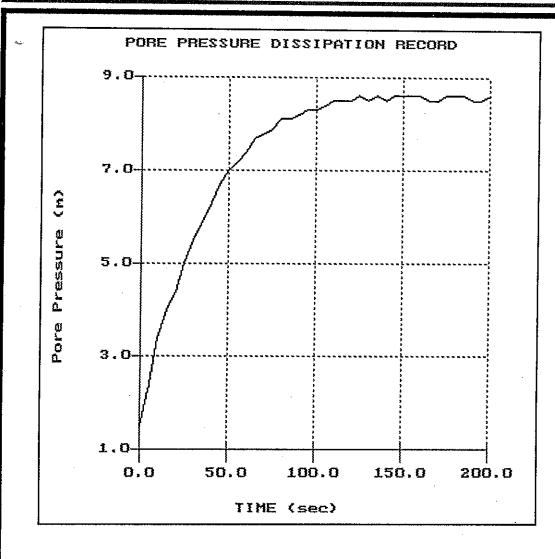
Hole:CPT 96-1 (1) Location:MT. POLLEY DAM Cone:10 TON A 023 Date: 07:27:96 08:15



File: 163CPO1.PPD Depth (m): 5.80 (ft): 19.03 Duration: 275.0s

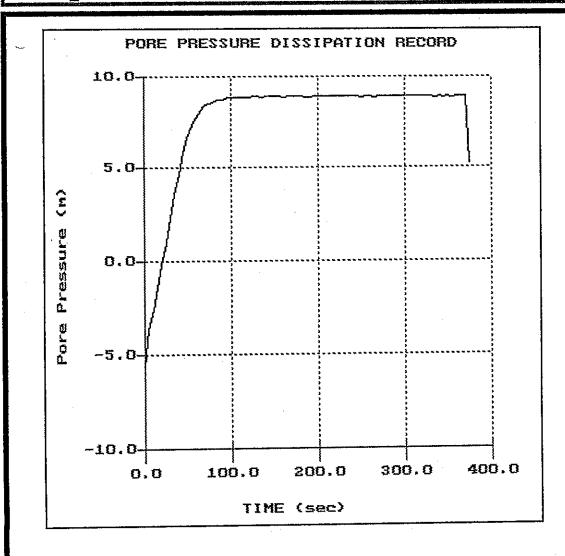
Hole:CPT 96-1 (1) Location:MT. POLLEY DAM

Cone:10 TON A 023 Date:07:27:96 08:15



File: 163CPO1.PPD Depth (m): 6.70 (ft): 21.98 Duration: 200.0s

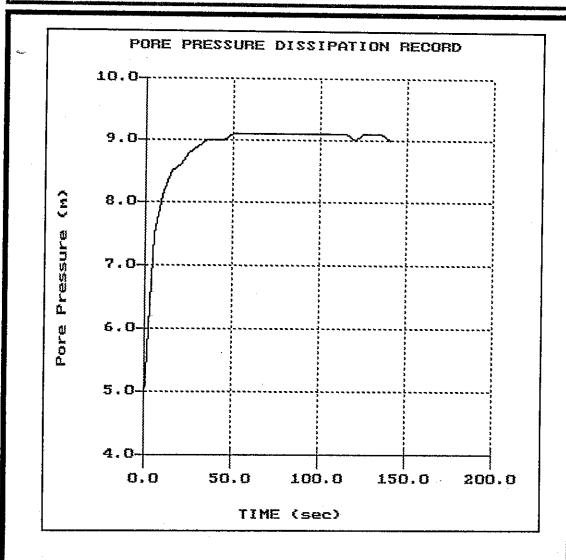
Hole:CPT 96-1 (1) Location:MT. POLLEY DAM Cone:10 TON A 023 Date:07:27:96 08:15



File: 163CPO1.PPD Depth (m): 7.50 (ft): 24.61 Duration: 375.0s

Hole:CPT 96-1 (1) Location:MT. POLLEY DAM

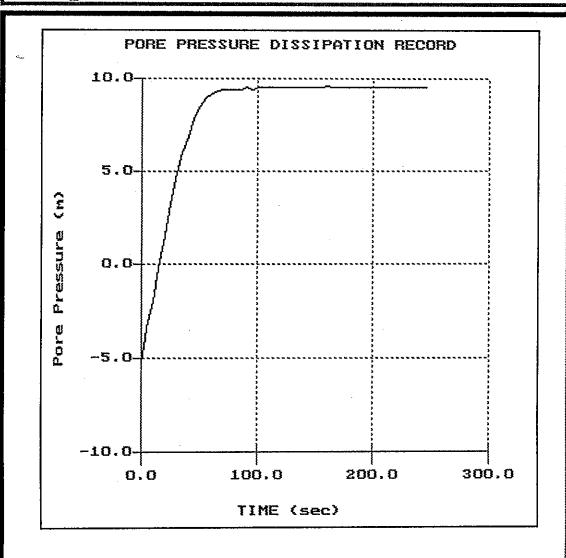
Cone:10 TON A 023 Date:07:27:96 08:15



File: 163CP01.PPD Depth (m): 7.70 (ft): 25.26 Duration: 140.0s

Hole:CPT 96-1 (1) Location:MT. POLLEY DAM

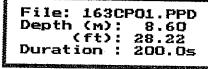
Cone:10 TON A 023 Date:07:27:96 08:15

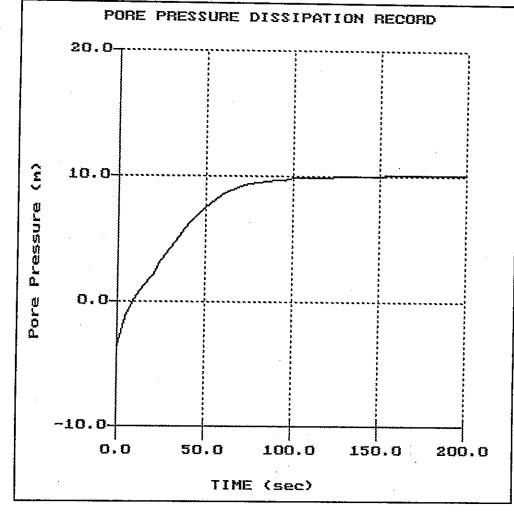


File: 163CPO1.PPD Depth (m): 8.10 (ft): 26.57 Duration: 245.0s Knight Piesold 10.0-

Hole:CPT 96-1 (1) Location: MT. POLLEY DAM

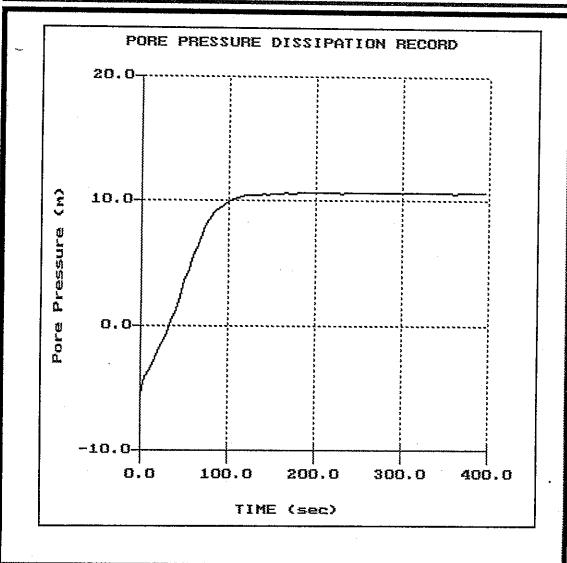
Cone:10 TON A 023 Date:07:27:96 08:15





Hole:CPT 96-1 (1) Location:MT. POLLEY DAM

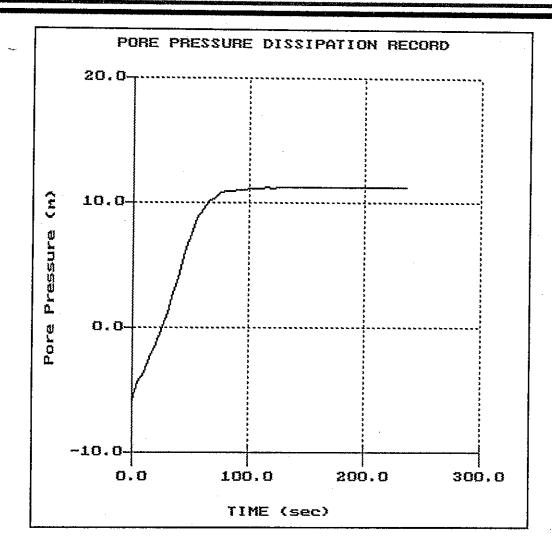
Cone:10 TON A 023 Date:07:27:96 08:15



File: 163CP01.PPD Depth (m): 9.05 (ft): 29.69 Duration: 395.0s

Hole:CPT 96-1 (1) Location:MT. POLLEY DAM

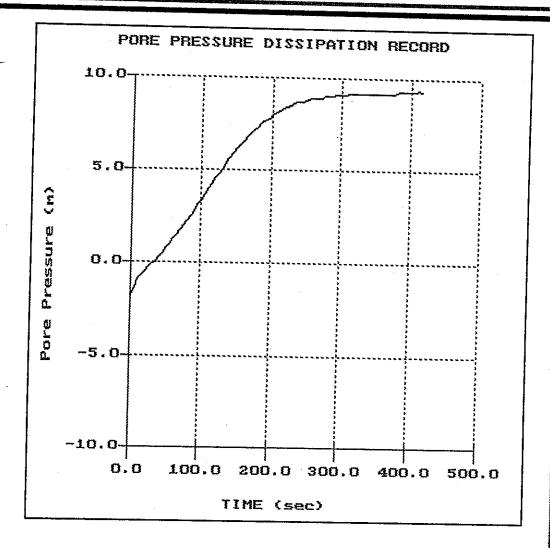
Cone:10 TON A 023 Date:07:27:96 08:15



File: 163CPO1.PPD Depth (m): 9.50 (ft): 31.17 Duration: 235.0s

Hole:PRW 96-3 (2) Location:MT. POLLEY DAM

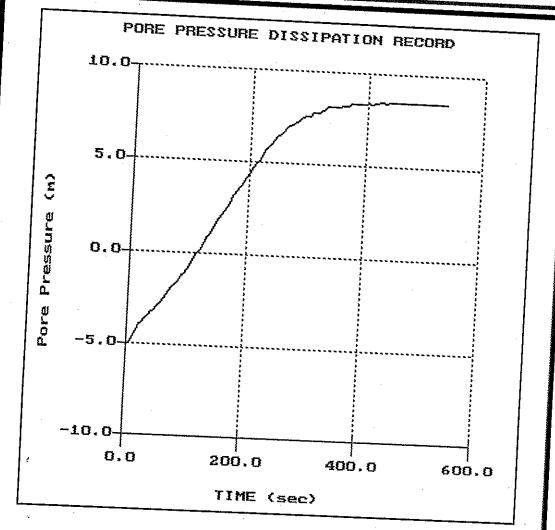
Cone:10 TON A 023 Date:07:27:96 11:43



File: 163CPO2.PPD Depth (m): 8.25 (ft): 27.07 Duration: 415.0s

Hole:PRW 96-4 (3) Location:MT. POLLEY DAM

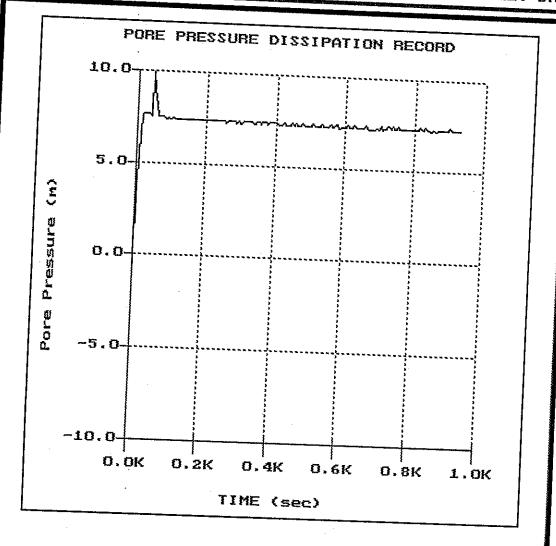
Cone:10 TON A 023 Date:07:27:96 13:51



File: 163CPO3.PPD Depth (m): 10.25 (ft): 33.63 Duration: 535.0s

Hole:PRW 96-2 (4) Location:MT. POLLEY DAM

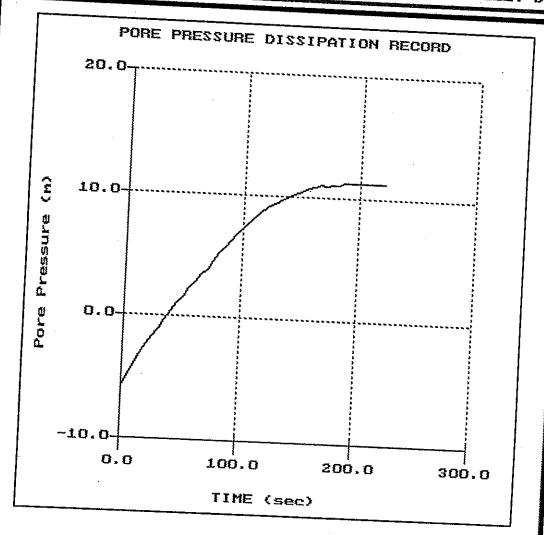
Cone:10 TON A 023 Date:07:27:96 16:27



File: 163CPO4.PPD Depth (m): 7.25 (ft): 23.79 Duration: 930.0s

Hole:PRW 96-2 (4) Location:MT. POLLEY DAM

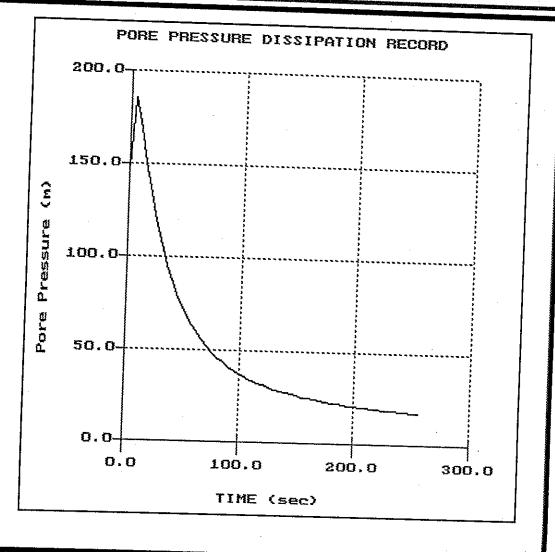
Cone:10 TON A 023 Date:07:27:96 16:27



File: 163CPO4.PPD Depth (m): 11.75 (ft): 38.55 Duration: 220.0s

Hole:PRW 96-2 (4A) Location:MT. POLLEY DAM

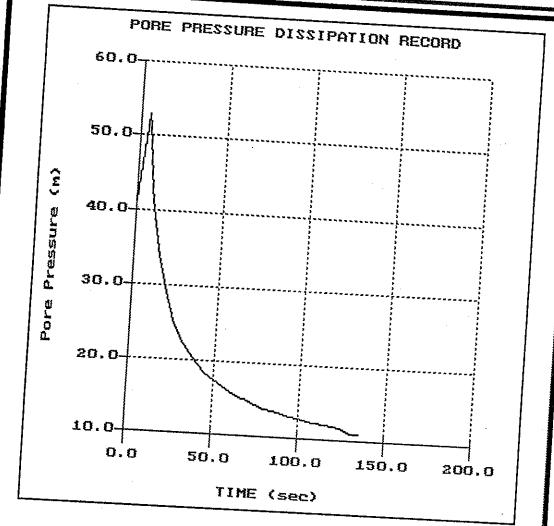
Cone:10 TON A 023 Date:07:28:96 08:43



File: 163CPO4A.PPD Depth (m): 8.75 (ft): 28.71 Duration: 255.0s

Hole:PRW 96-2 (4A) Location:MT. POLLEY DAM

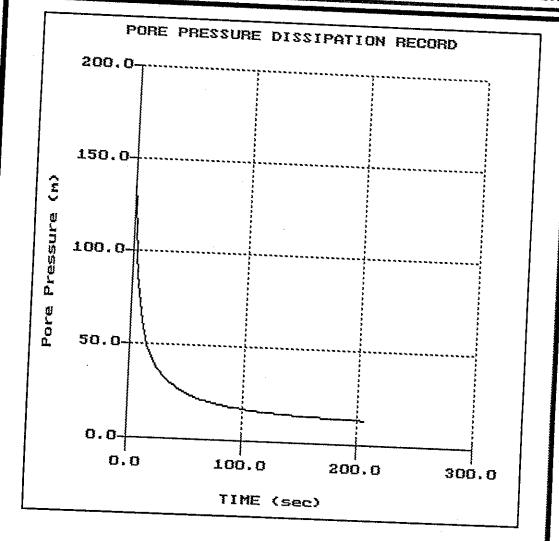
Cone:10 TON A 023 Date:07:28:96 08:43



File: 163CPO4A.PPD Depth (m): 9.75 (ft): 31.99 Duration: 135.0s

Hole:PRW 96-2 (4A) Location:MT. POLLEY DAM

Cone:10 TON A 023 Date:07:28:96 08:43

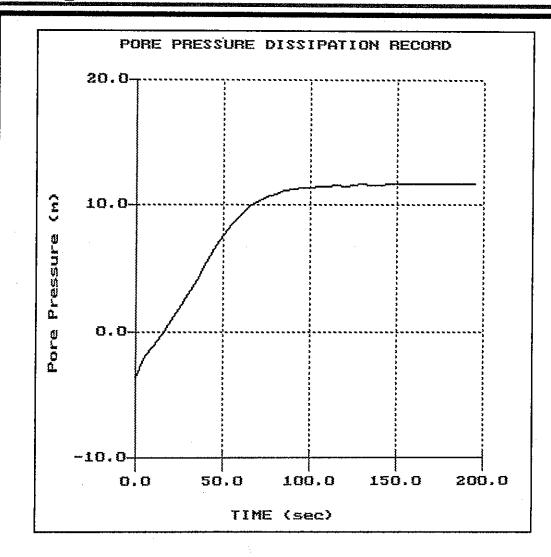


File: 163CPO4A.PPD Depth (m): 10.75 (ft): 35.27

Duration: 205.0s

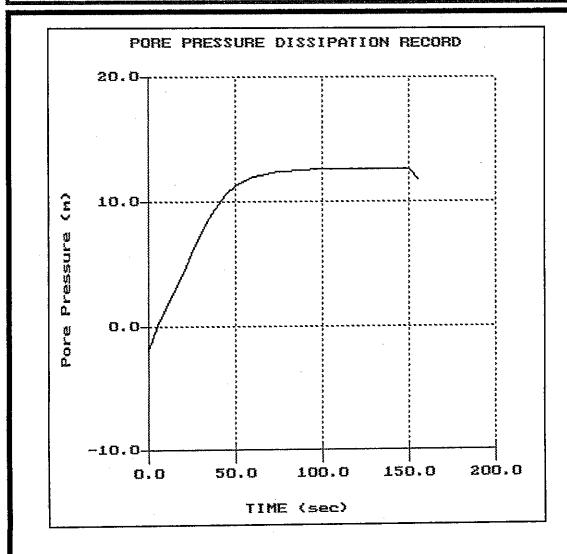
Hole:PRW 96-2 (4A) Location: MT. POLLEY DAM

Cone:10 TON A 023 Date:07:28:96 08:43



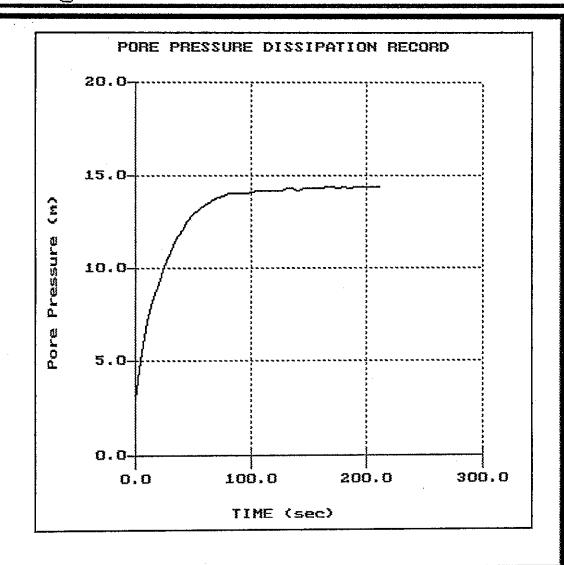
File: 163CPO4A.PPD Depth (m): 11.75 (ft): 38.55 Duration: 195.0s

Hole:PRW 96-2 (4A) Location: MT. POLLEY DAM Cone:10 TON A 023 Date:07:28:96 08:43



File: 163CPO4A.PPD Depth (m): 12.75 (ft): 41.83 Duration: 155.0s

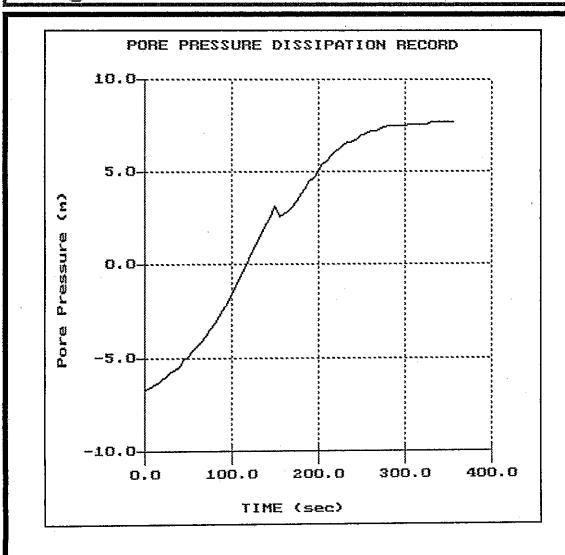
Hole:PRW 96-2 (4A) Location:MT. POLLEY DAM Cone:10 TON A 023 Date:07:28:96 08:43



File: 163CPO4A.PPD Depth (m): 14.75 (ft): 48.39

Duration: 210.0s

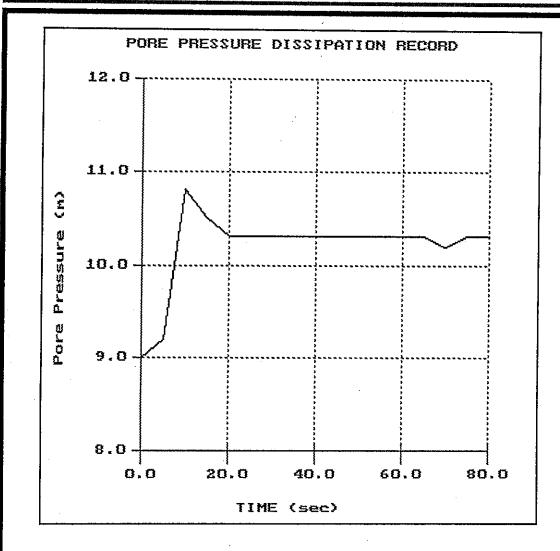
Hole:CPT 96-2 (6) Location:MT. POLLEY DAM Cone:10 TON A 023 Date:07:28:96 13:48



File: 163CPO6.PPD Depth (m): 7.75 (ft): 25.43 Duration: 355.0s

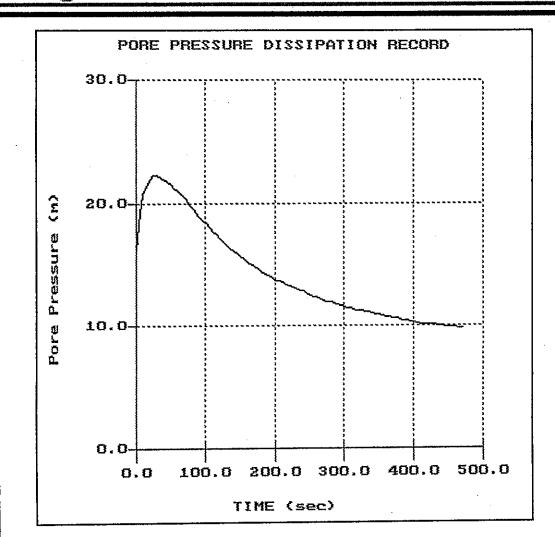
Hole:CPT 96-3 (7) Location:MT. POLLEY DAM

Cone:10 TON A 023 Date:07:28:96 15:08



File: 163CPO7.PPD Depth (m): 10.55 (ft): 34.61 Duration: 80.0s

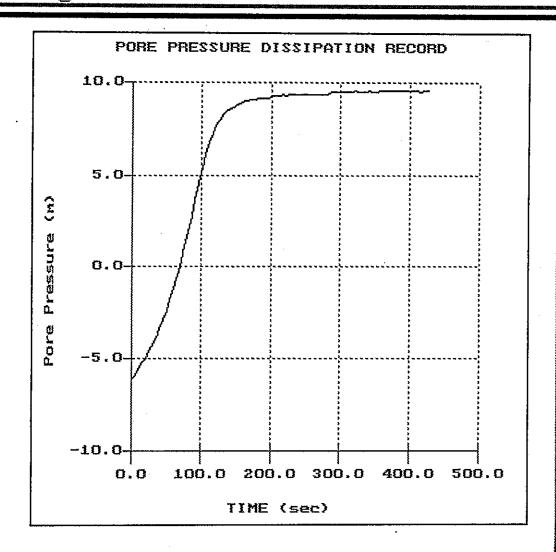
Hole:CPT 96-4 (8) Location: MT. POLLEY DAM Cone:10 TON A 005 Date:07:28:96 16:17



File: 163CPO8.PPD Depth (m): 6.95 (ft): 22.80 Duration: 470.0s

Hole:CPT 96-4 (8) Location:MT. POLLEY DAM

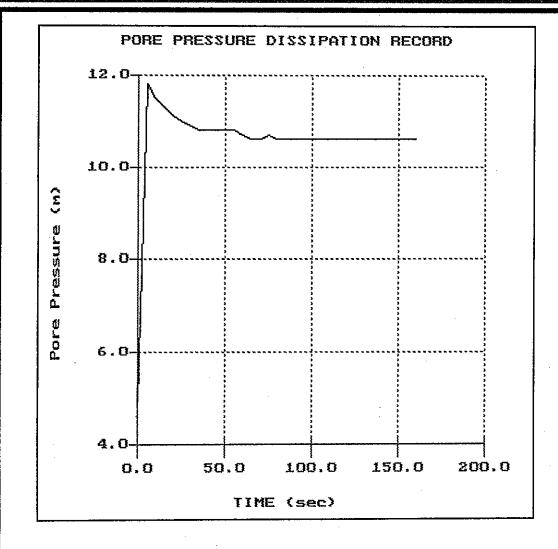
Cone:10 TON A 005 Date:07:28:96 16:17



File: 163CPO8.PPD Depth (m): 8.95

(ft): 29.36 Duration: 425.0s

Hole:CPT 96-4 (8) Location:MT. POLLEY DAM Cone:10 TON A 005 Date:07:28:96 16:17



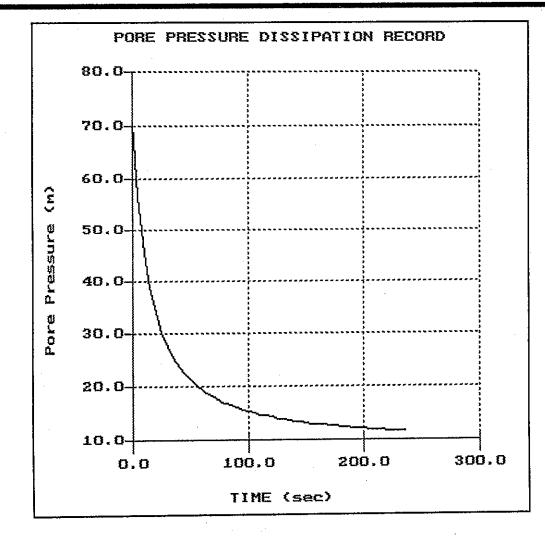
File: 163CPO8.PPD Depth (m): 9.95 (ft): 32.64 Duration: 160.0s Knight Piesold Hole:CPT 96-5 (9) Location: MT. POLLEY DAM PORE PRESSURE DISSIPATION RECORD 5.0-2.0--1.0--4.0-0.0 200.0 400.0 600.0 TIME (sec)

Cone:10 TON A 005 Date:07:28:96 17:20

File: 163CPO9.PPD Depth (m): 5.10

(ft): 16.73 Duration: 595.0s

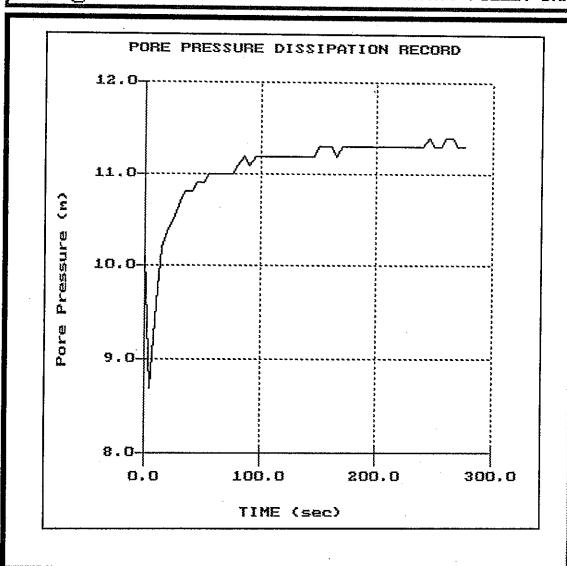
Hole:CPT 96-5 (9) Location:MT. POLLEY DAM Cone:10 TON A 005 Date:07:28:96 17:20



File: 163CP09.PPD Depth (m): 8.40 (ft): 27.56 Duration: 235.0s

Hole:CPT 96-5 (9) Location:MT. POLLEY DAM

Cone:10 TON A 005 Date:07:28:96 17:20



File: 163CP09.PPD Depth (m): 10.00 (ft): 32.81 Duration: 275.0s

APPENDIX E SHEAR WAVE VELOCITY DATA

				tìp	•		Vel	Pro	(m)
				above					
				er 0.20m	Average	Depth	of	sensor	Œ
	۲-	LEY MINE		Shear Source Offset 0.6m / Seismometer 0.20m above tip			Corrected of	Distance CrossoverCrossover sensor	(msec) (msec) (m)
ions Ltd.	Reduction	MOUNT POI	•	ot 0.6m /	Beam			Crossover	(msec)
Investiga	Cone Data	PEISOLD		urce Offse	Source Hammer and Beam	1996		Distance	(E)
ConeTec Investigations Ltd.	Seismic	KNIGHT &	SCPT-4A	Shear So	Source H	July 27, 1996	Geophone	Depth	(E)

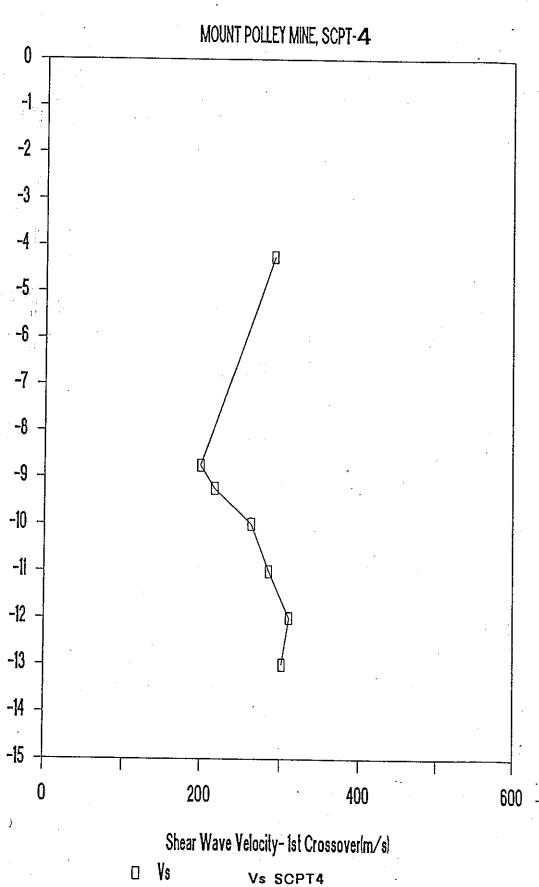
Velocity Profile (m/sec)	292.8 199.8 217.1 285.2 312.0 302.6 277.5
of sensor (m)	4.8.9. 6.25.25. 7.1.1.1.4.1.
Corrected Distance Crossover(msec)	29.1 29.02777 31.6 31.53001 33.9 33.83258 37.7 37.63859 41.2 41.14403 44.4 44.34893 47.7 47.65295 51.3 51.25613
Distance (m)	8.521150 9.019977 9.518928 10.51712 11.51564 12.51439 13.51332 14.51240
cy 27, ophone oth n)	8 <u>001151877</u> 201151875

ConeTec Investigations Ltd.
Seismic Cone Data Reduction
KNIGHT & PEISOLD MOUNT POLLEY MINE
SCPT-1
Shear Source Offset 0.6m / Seismometer 0.20m above tip
Source Hammer and Beam
July 27, 1996
Geophone
Geophone
Depth
Distance CrossoverCrossover sensor
(m) (msec) (msec) (m)

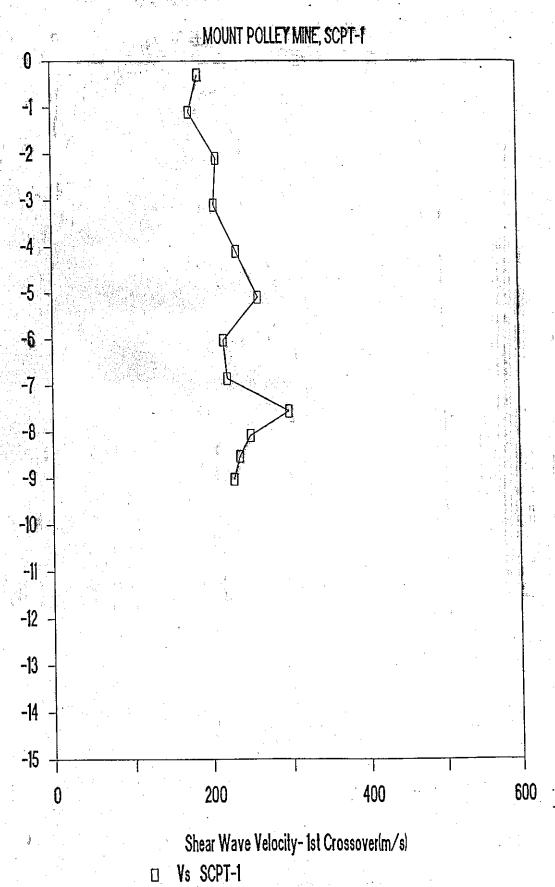
Velocity Profile (m/sec)

187.5 175.6 209.9 206.6 233.7 259.8 216.5 221.1 298.1 248.9 236.0
-0.3 -2.1 -2.1 -6.025 -6.025 -7.55 -8.075 -9.025
8.5 6.010407 12.5 11.70411 16.9 16.46721 21.6 21.30610 25.8 25.58329 29.6 29.43155 33.5 33.35599 37.1 36.97360 39.1 38.98628 40.9 40.79355 45.2 45.10622
0.6 0.848528 1.6 1.708800 2.6 2.668332 3.6 3.649657 4.6 4.638965 5.6 5.632051 6.45 6.477846 7.25 7.274785 7.85 7.274785 7.85 7.274785 8.75 8.770547 9.3 9.319334





SHEAR WAVE VELOCITY VS DEPTH



Depth Below Grade (metres)