SAMPLE: END09-07-02



Length: 78.89 mm

Diameter: 45.31 mm

Density: 2.457 g/cm³

Peak Strength: 48.5 MPa

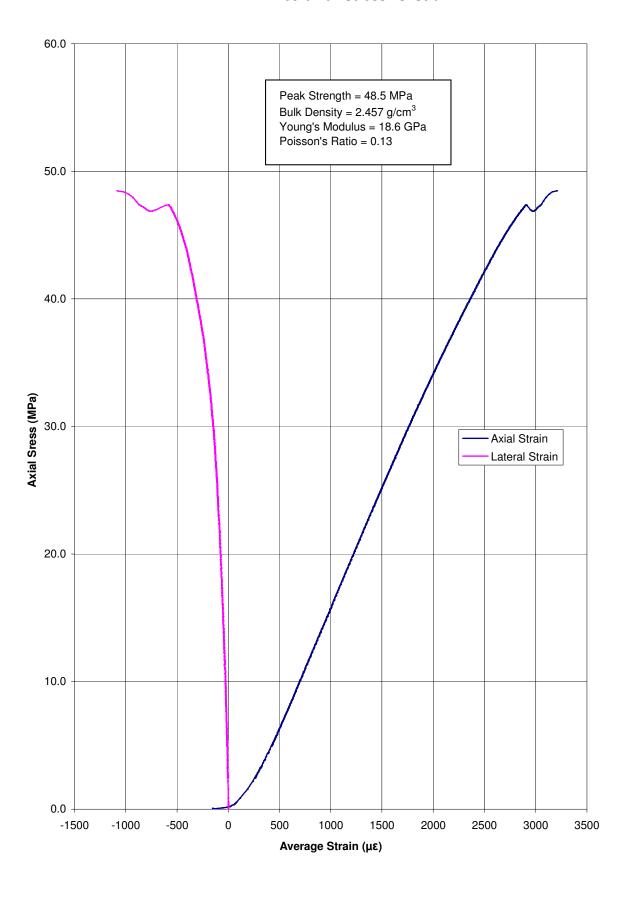
Young's Modulus: 18.6 GPa

Poisson's Ratio: 0.13

Failure Cause: Microfracture



END09-07-02 Stress vs. Strain



SAMPLE: END09-07-03



Length: 99.18 mm

Diameter: 45.08 mm Density: 2.662 g/cm³

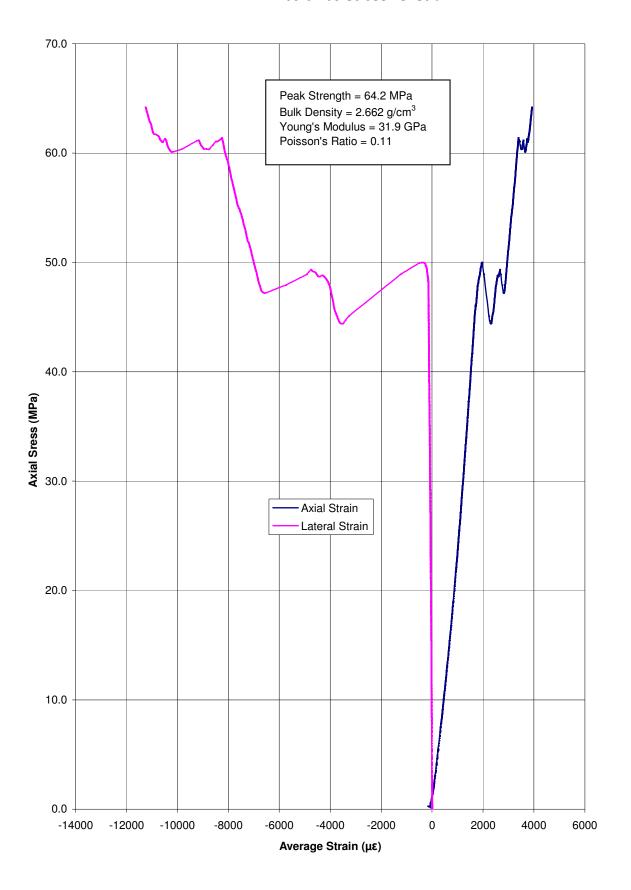
Peak Strength: 64.2 MPa

Young's Modulus: 31.9 GPa

Poisson's Ratio: 0.11



END09-07-03 Stress vs. Strain



SAMPLE: END09-11-02



Length: 100.51 mm Diameter: 47.25 mm

Density: 2.029 g/cm³

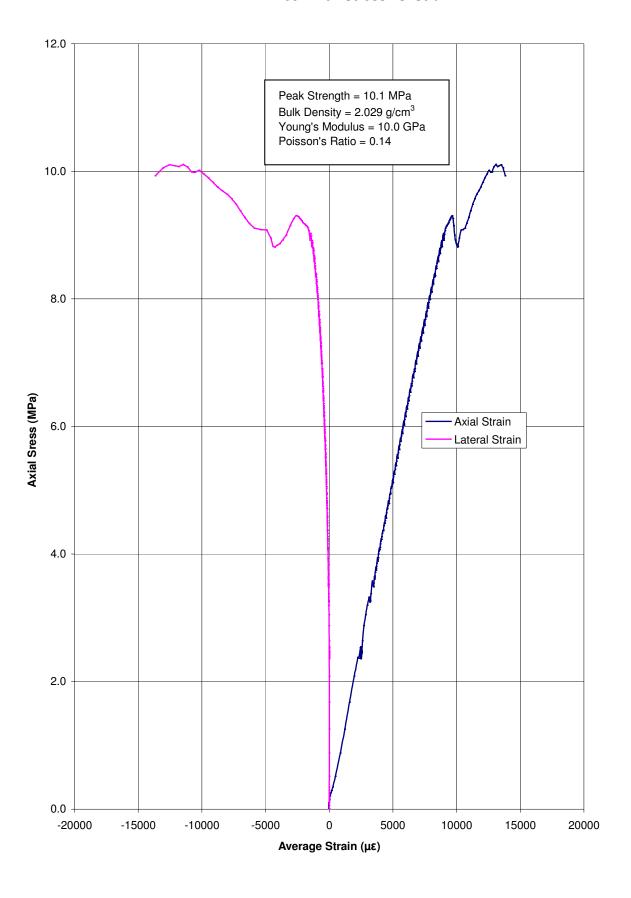
Peak Strength: 10.1 MPa

Young's Modulus: 10.0 GPa

Poisson's Ratio: 0.14



END09-11-02 Stress vs. Strain



SAMPLE: END09-11-03



Length: 112.01 mm

Diameter: 47.7 mm

Density: 2.463 g/cm³

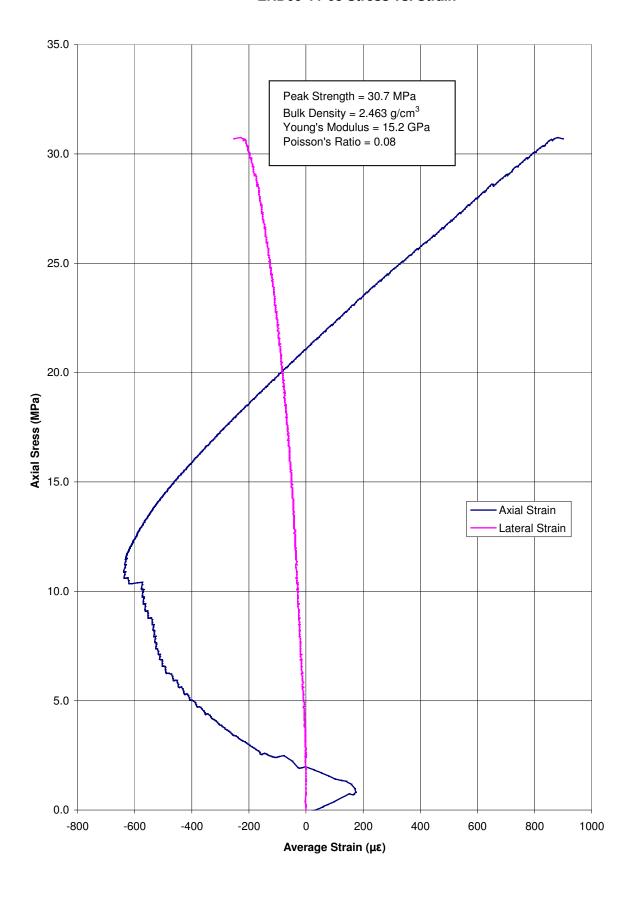
Peak Strength: 30.7 MPa

Young's Modulus: 15.2 GPa

Poisson's Ratio: 0.08



END09-11-03 Stress vs. Strain



SAMPLE: END09-11-04



Length: 113.6 mm

Diameter: 47.47 mm Density: 2.521 g/cm³

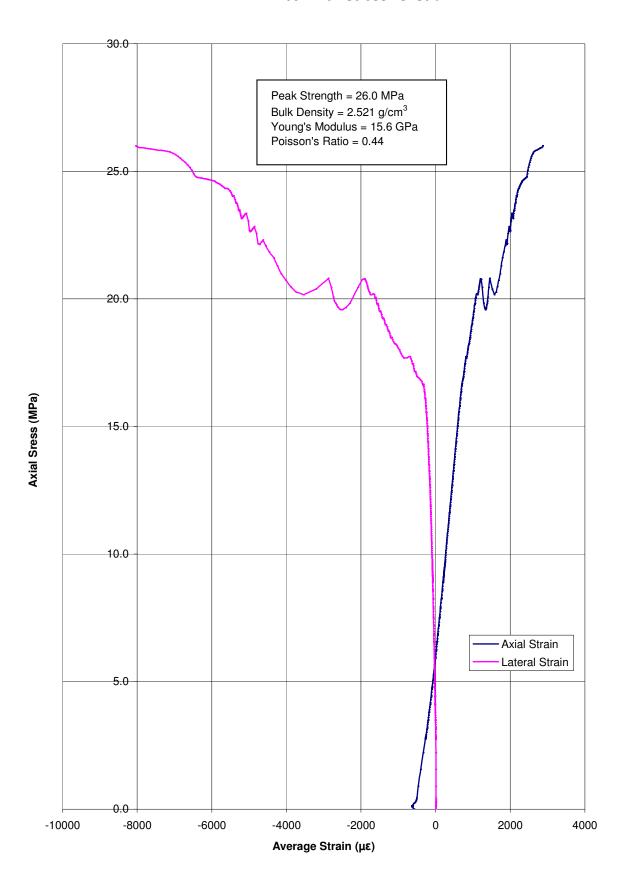
Peak Strength: 26.0 MPa

Young's Modulus: 15.6 GPa

Poisson's Ratio: 0.44



END09-11-04 Stress vs. Strain



SAMPLE: END09-11-06



Length: 112.26 mm Diameter: 47.51 mm

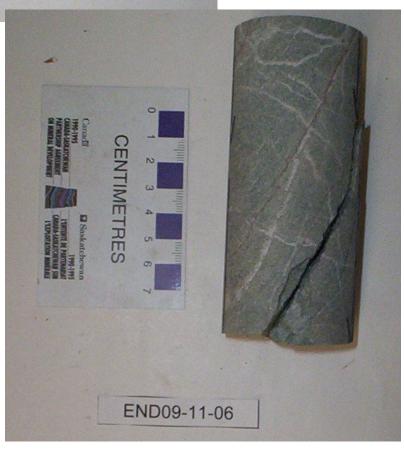
Density: 2.451 g/cm³

Peak Strength: 42.4 MPa

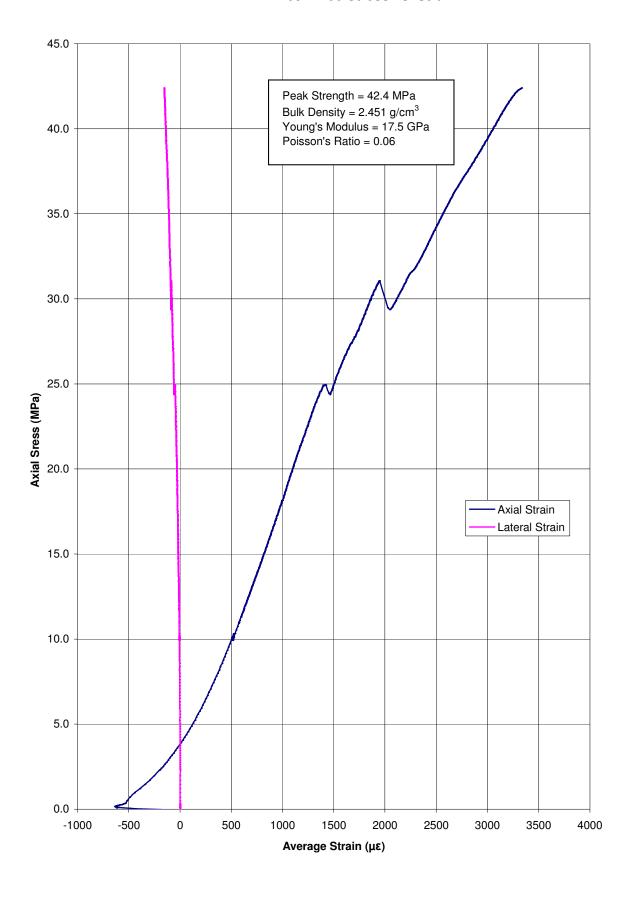
Young's Modulus: 17.5 GPa

Poisson's Ratio: 0.06

Failure Cause: Microfracture



END09-11-06 Stress vs. Strain



SAMPLE: END09-11-09



Length: 103.02 mm Diameter: 47.06 mm

Density: 2.202 g/cm³

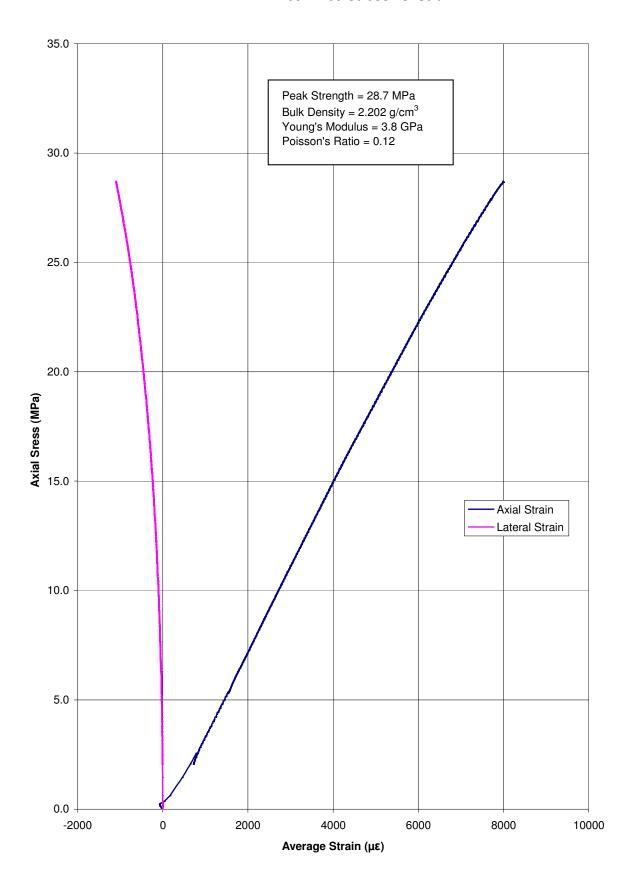
Peak Strength: 28.7 MPa

Young's Modulus: 3.8 GPa

Poisson's Ratio: 0.12



END09-11-09 Stress vs. Strain



SAMPLE: END09-12-04



Length: 94.16 mm

Diameter: 47.78 mm

Density: 2.491 g/cm³

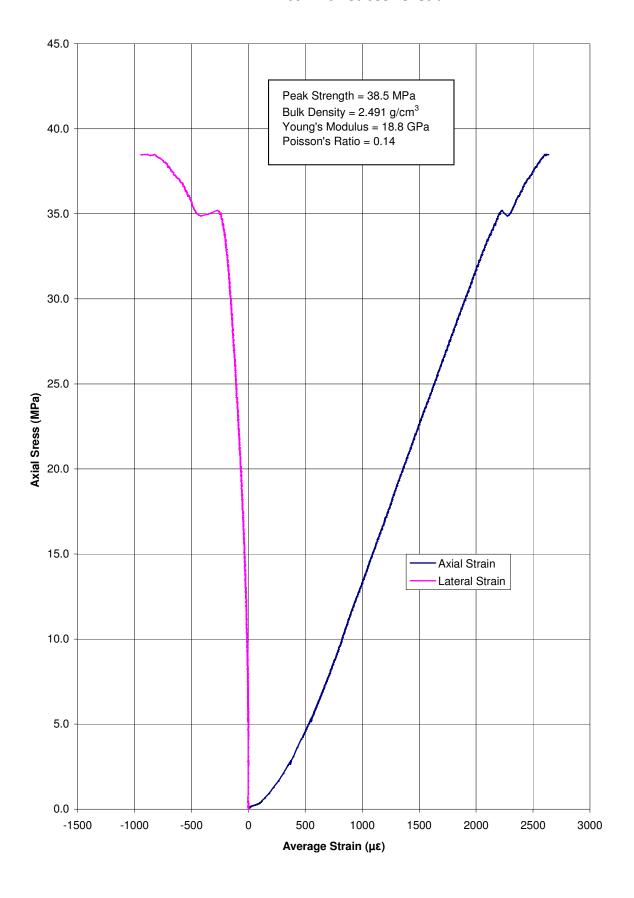
Peak Strength: 38.5 MPa

Young's Modulus: 18.8 GPa

Poisson's Ratio: 0.14



END09-12-04 Stress vs. Strain



SAMPLE: END09-12-06



Length: 97.82 mm

Diameter: 47.75 mm Density: 2.601 g/cm³

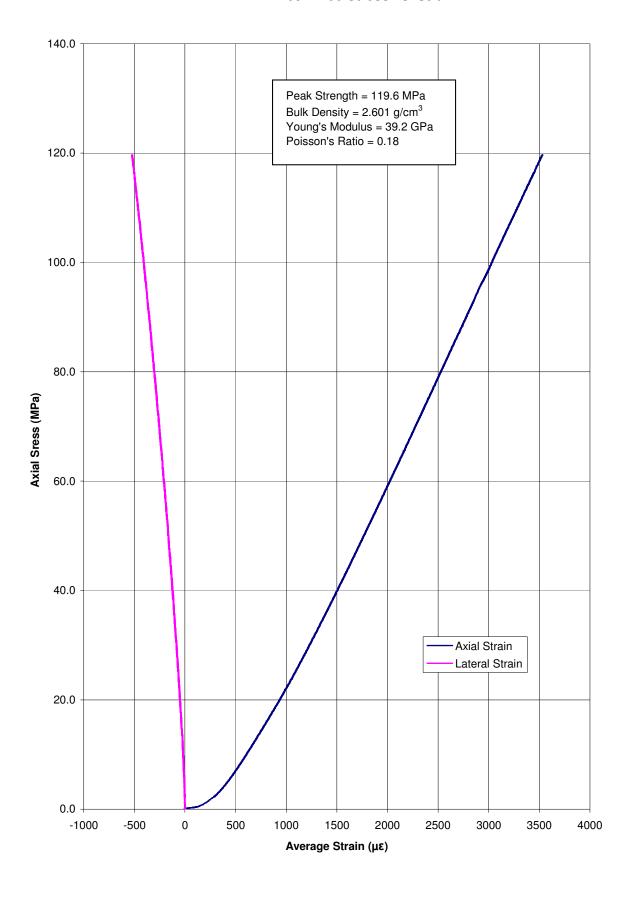
Peak Strength: 119.6 MPa

Young's Modulus: 39.2 GPa

Poisson's Ratio: 0.18



END09-12-06 Stress vs. Strain



SAMPLE: MZ09-01A-02



Length: 107.28 mm Diameter: 44.86 mm Density: 2.678 g/cm³

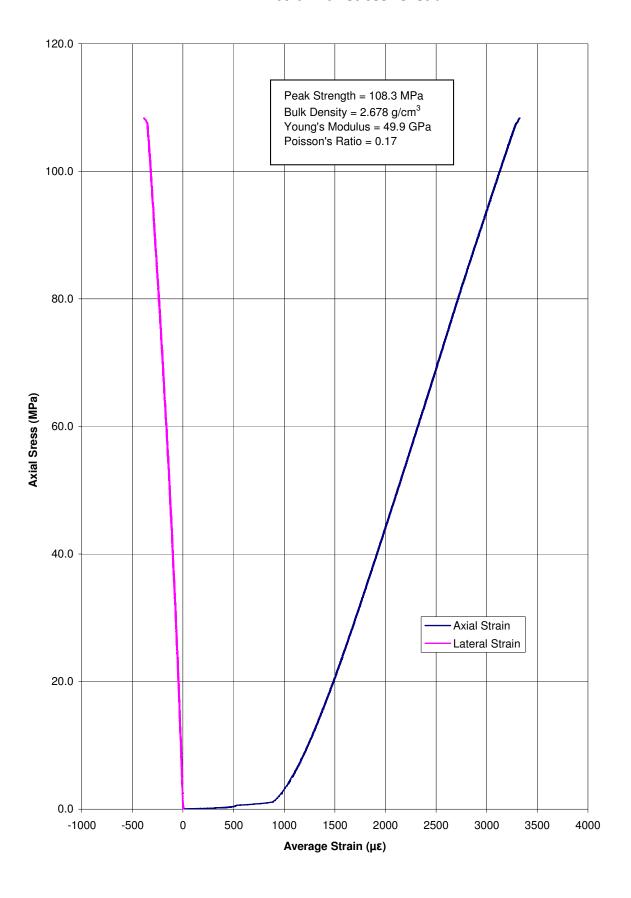
Peak Strength: 108.3 MPa

Young's Modulus: 49.9 GPa

Poisson's Ratio: 0.17



MZ09-01A-02 Stress vs. Strain



SAMPLE: MZ09-01A-03



Length: 108.29 mm

Diameter: 45.01 mm

Density: 2.713 g/cm³

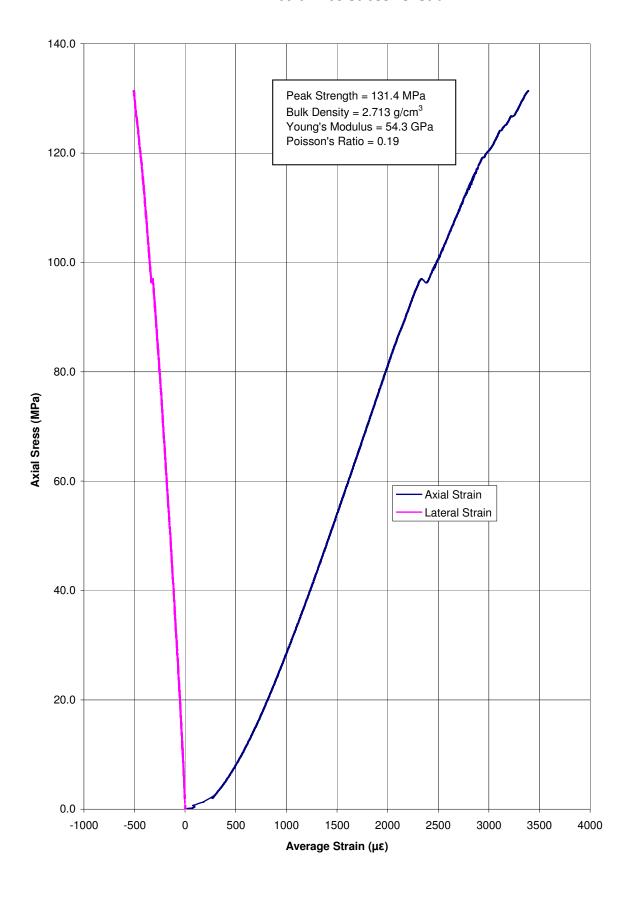
Peak Strength: 131.4 MPa

Young's Modulus: 54.3 GPa

Poisson's Ratio: 0.19



MZ09-01A-03 Stress vs. Strain



SAMPLE: MZ09-01A-05



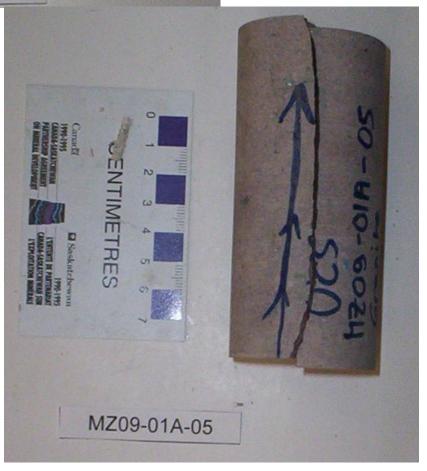
Length: 106.62 mm

Diameter: 44.61 mm Density: 2.619 g/cm³

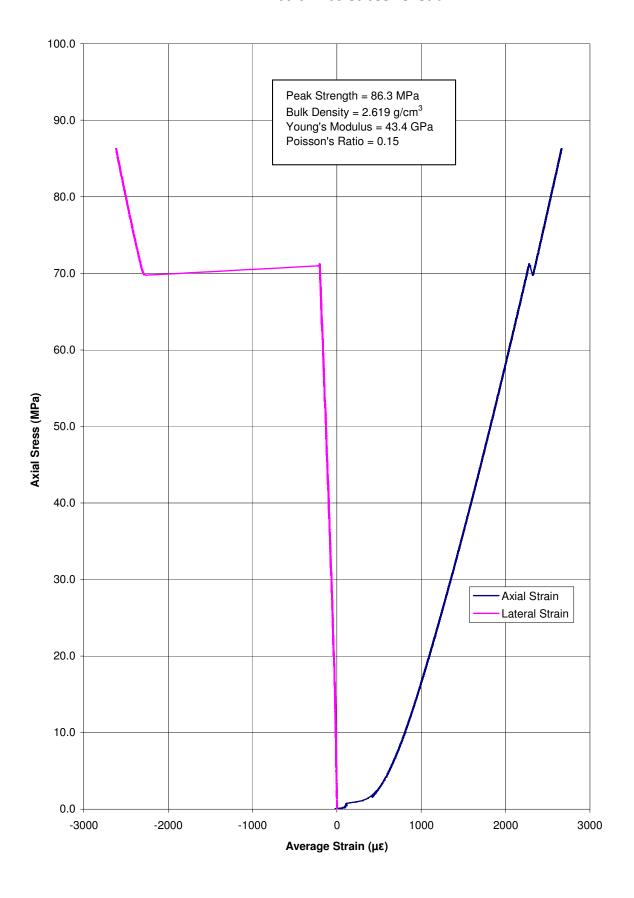
Peak Strength: 86.3 MPa

Young's Modulus: 43.4 GPa

Poisson's Ratio: 0.15



MZ09-01A-05 Stress vs. Strain



SAMPLE: MZ09-01A-06



Length: 109.53 mm

Diameter: 44.69 mm Density: 2.236 g/cm³

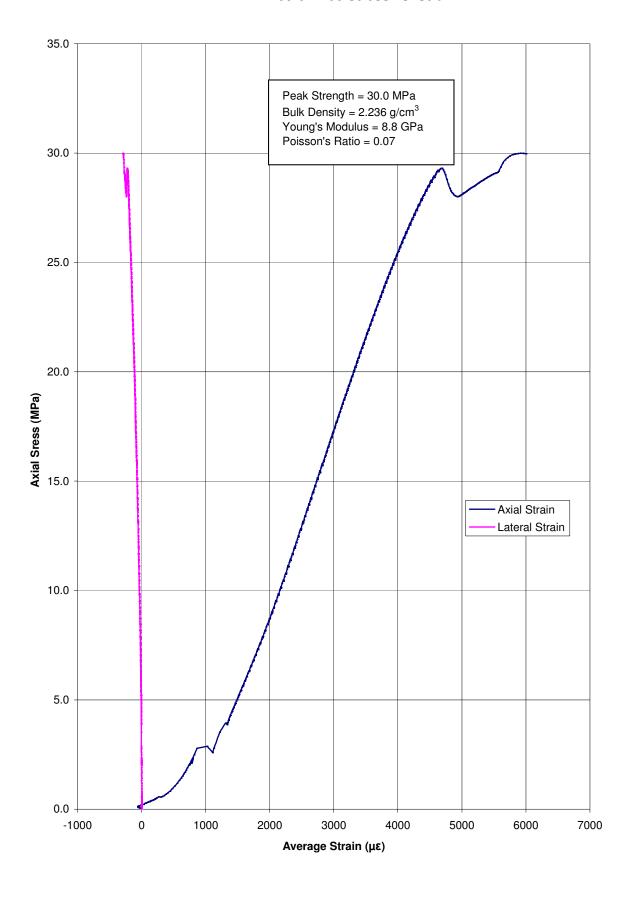
Peak Strength: 30.0 MPa

Young's Modulus: 8.8 GPa

Poisson's Ratio: 0.07



MZ09-01A-06 Stress vs. Strain



SAMPLE: MZ09-01A-07



Length: 108.02 mm

Diameter: 44.48 mm

Density: 2.360 g/cm³

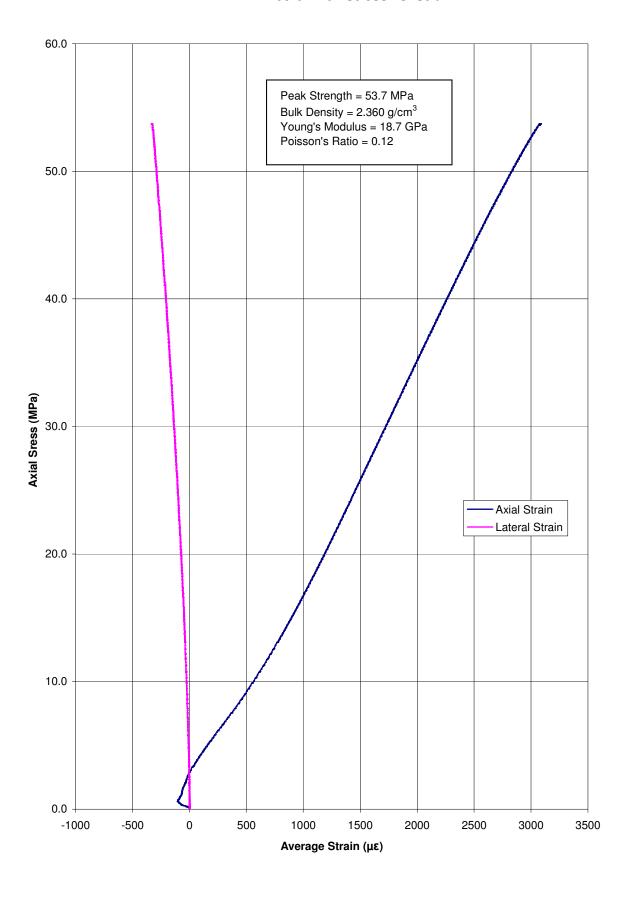
Peak Strength: 53.7 MPa

Young's Modulus: 18.7 GPa

Poisson's Ratio: 0.12



MZ09-01A-07 Stress vs. Strain



SAMPLE: MZ09-02-01



Length: 89.15 mm

Diameter: 44.95 mm Density: 2.367 g/cm³

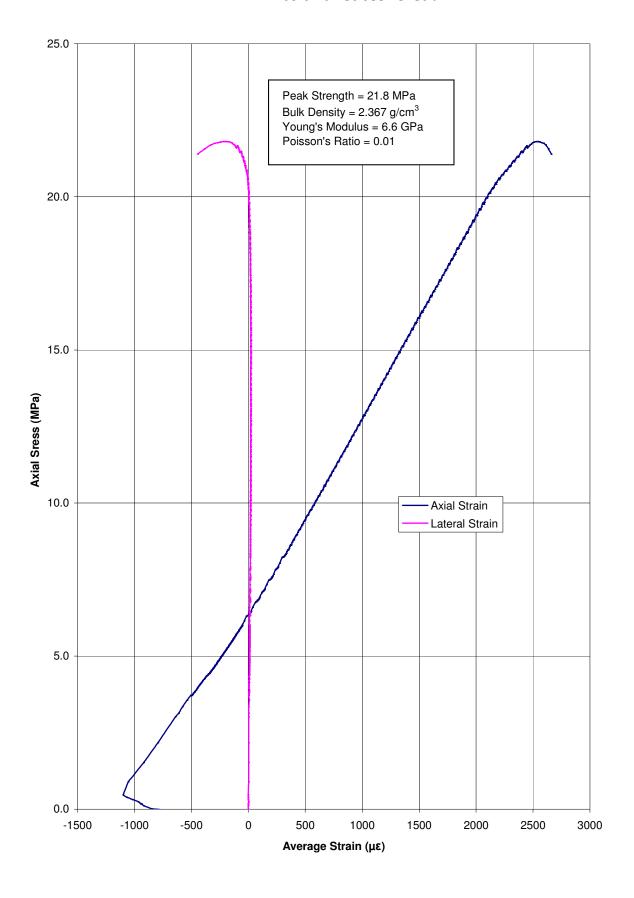
Peak Strength: 21.8 MPa

Young's Modulus: 6.6 GPa

Poisson's Ratio: 0.01



MZ09-02-01 Stress vs. Strain



SAMPLE: MZ09-02-02



Length: 108.45 mm

Diameter: 44.78 mm

Density: 2.463 g/cm³

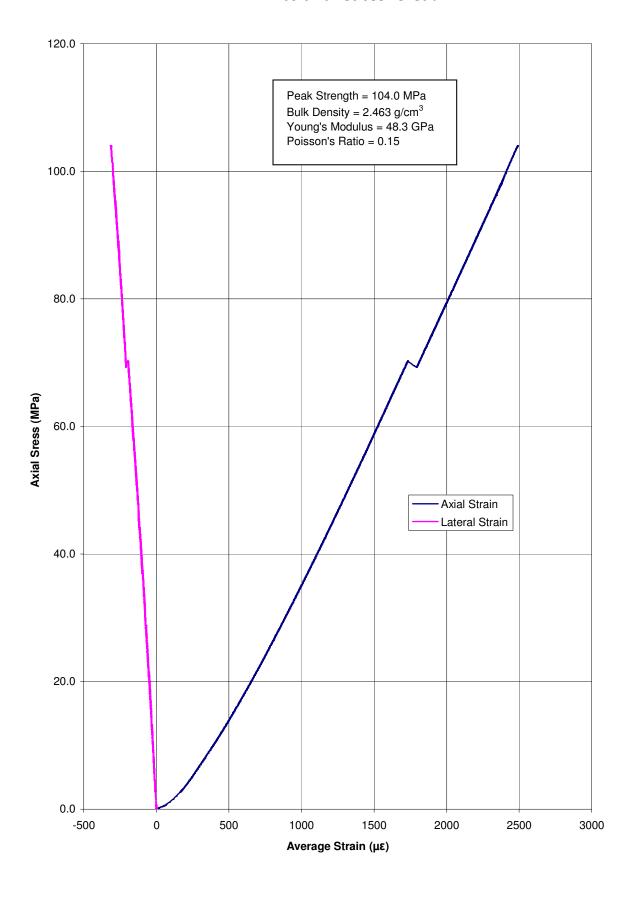
Peak Strength: 104.0 MPa

Young's Modulus: 48.3 GPa

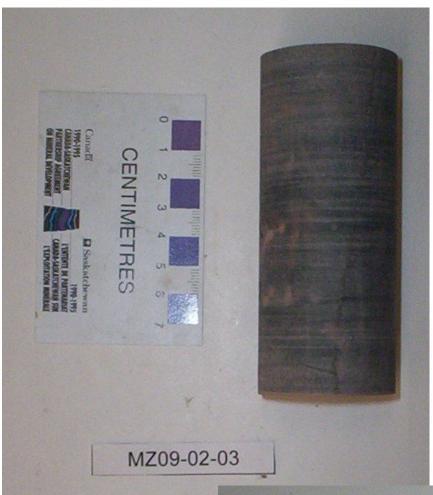
Poisson's Ratio: 0.15



MZ09-02-02 Stress vs. Strain



SAMPLE: MZ09-02-03



Length: 109.64 mm Diameter: 44.85 mm

Density: 2.733 g/cm³

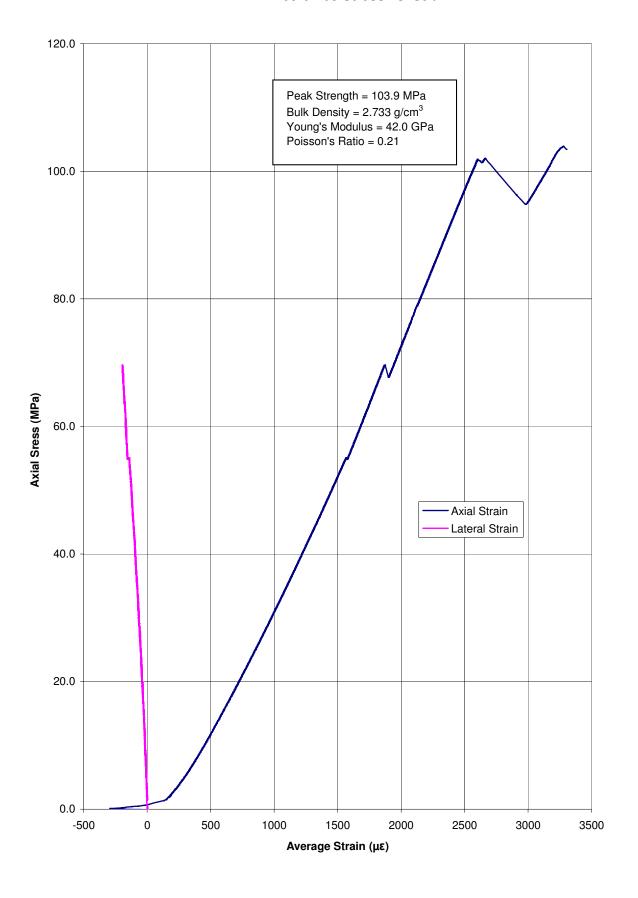
Peak Strength: 103.9 MPa

Young's Modulus: 42.0 GPa

Poisson's Ratio: 0.21



MZ09-02-03 Stress vs. Strain



SAMPLE: MZ09-02-04



Length: 106.41 mm

Diameter: 44.98 mm Density: 2.669 g/cm³

Peak Strength: 83.9 MPa

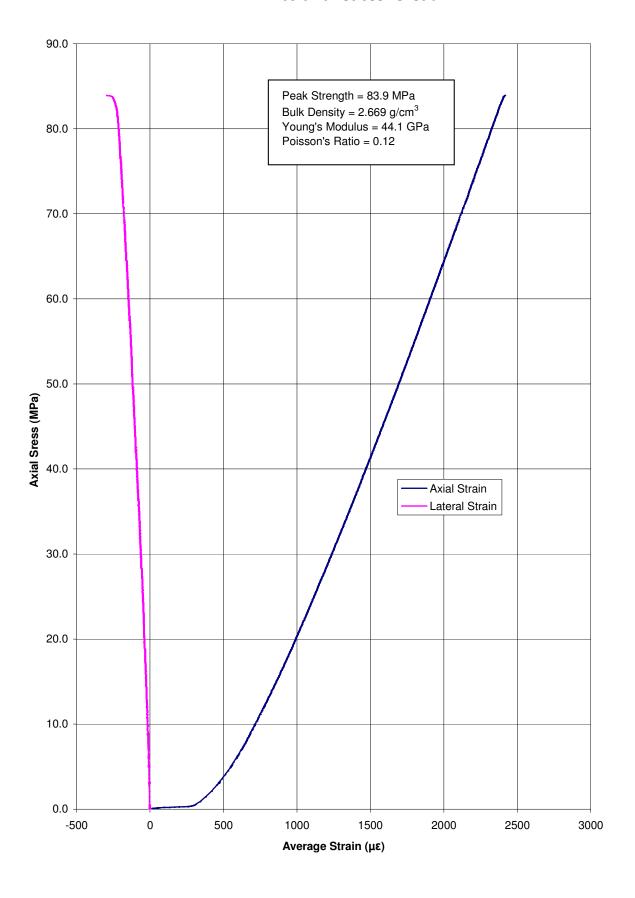
Young's Modulus: 44.1 GPa

Poisson's Ratio: 0.12

Failure Cause: Weakness Plane



MZ09-02-04 Stress vs. Strain



SAMPLE: MZ09-02-05



Length: 106.91 mm

Diameter: 44.91 mm Density: 2.721 g/cm³

Peak Strength: 152.2 MPa

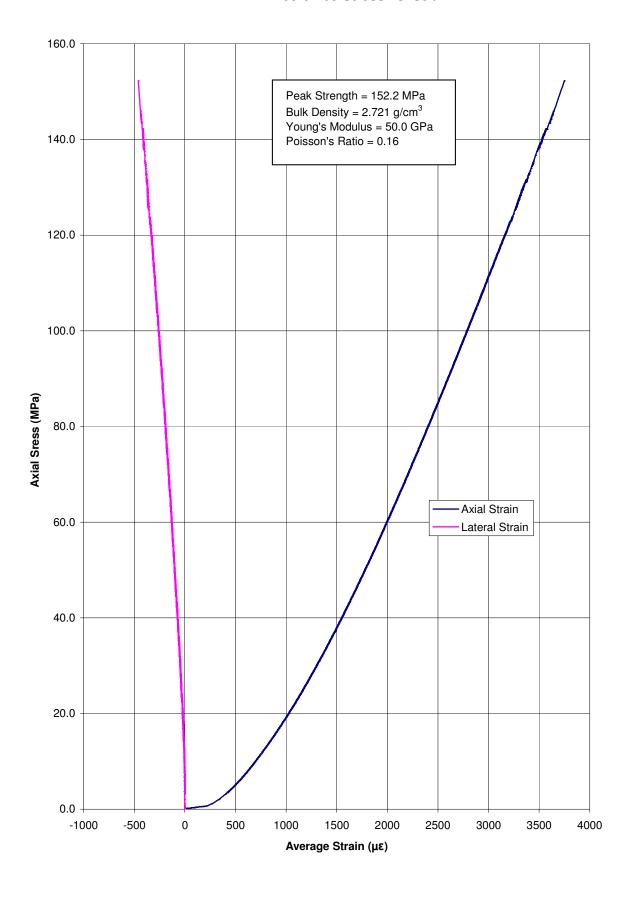
Young's Modulus: 50.0 GPa

Poisson's Ratio: 0.16

Failure Cause: Intact Rock



MZ09-02-05 Stress vs. Strain



SAMPLE: MZ09-02-06



Length: 109.13 mm Diameter: 45.14 mm Density: 2.613 g/cm³

Peak Strength: 111.1 MPa

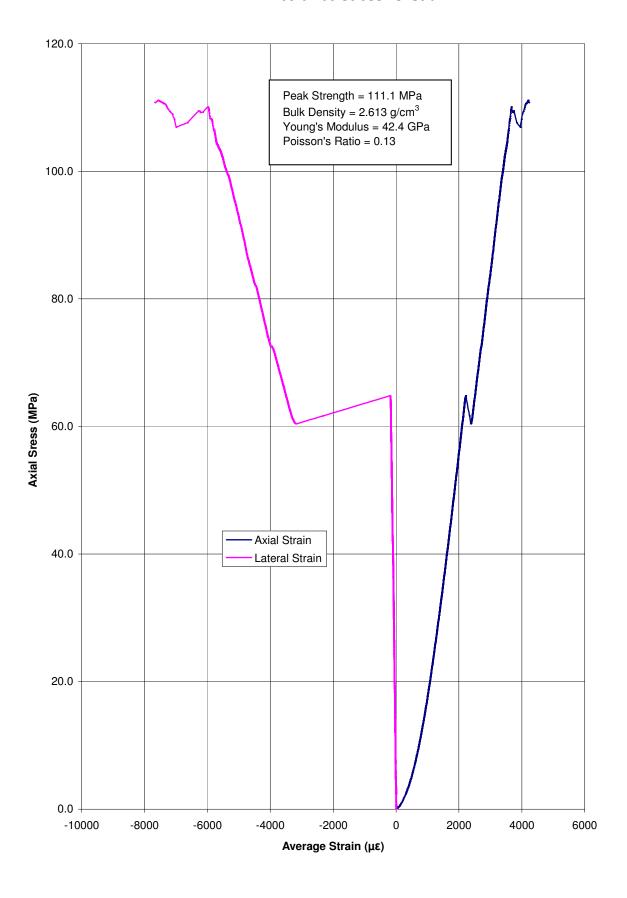
Young's Modulus: 42.4 GPa

Poisson's Ratio: 0.13

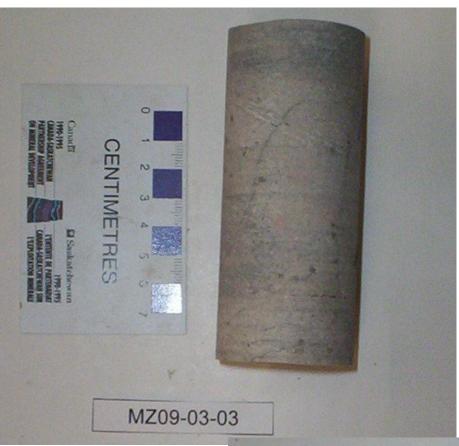
Failure Cause: Intact Rock



MZ09-02-06 Stress vs. Strain



SAMPLE: MZ09-03-03



Length: 108.46 mm

Diameter: 45.09 mm Density: 2.697 g/cm³

Peak Strength: 101.7 MPa

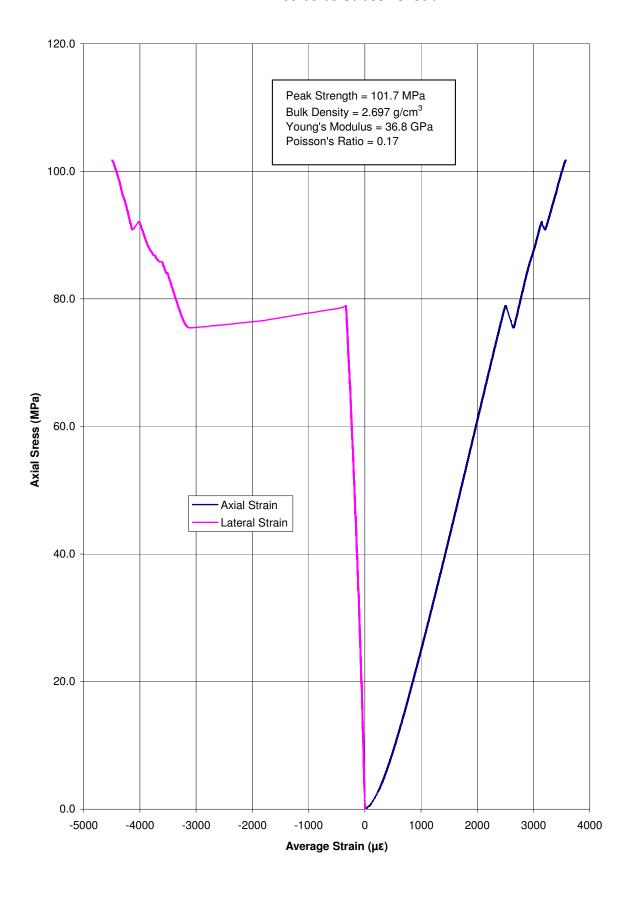
Young's Modulus: 36.8 GPa

Poisson's Ratio: 0.17

Failure Cause: Intact Rock



MZ09-03-03 Stress vs. Strain



SAMPLE: MZ09-03-04



Length: 109.51 mm

Diameter: 45.03mm

Density: 2.614 g/cm³

Peak Strength: 99.5 MPa

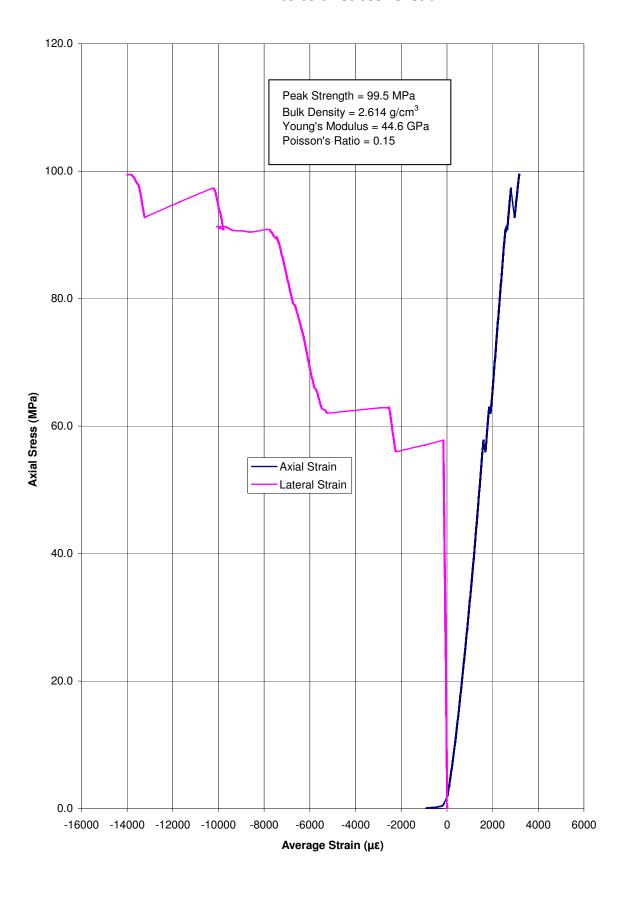
Young's Modulus: 44.6GPa

Poisson's Ratio: 0.15

Failure Cause: Intact Rock



MZ09-03-04 Stress vs. Strain



SAMPLE: MZ09-03-05



Length: 108.63 mm

Diameter: 45.03 mm

Density: 2.691 g/cm³

Peak Strength: 119.5 MPa

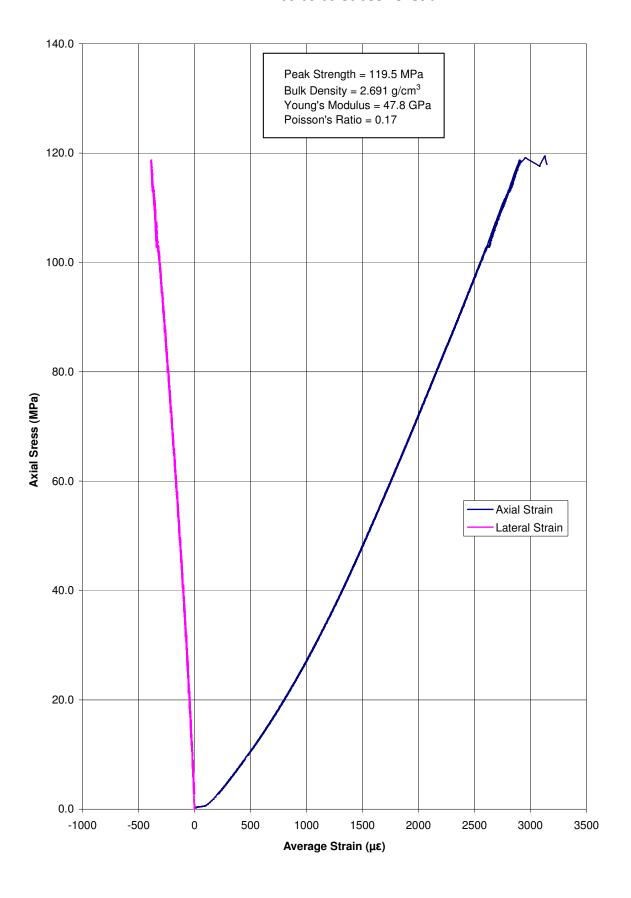
Young's Modulus: 47.8 GPa

Poisson's Ratio: 0.17

Failure Cause: Intact Rock



MZ09-03-05 Stress vs. Strain



SAMPLE: CZ09-01-02



Length: N/A

Diameter: N/A

Density: N/A

Peak Strength: N/A

Young's Modulus: N/A

Poisson's Ratio: N/A

Failure Cause N/A



SAMPLE: END09-05-02



Length: N/A

Diameter: N/A

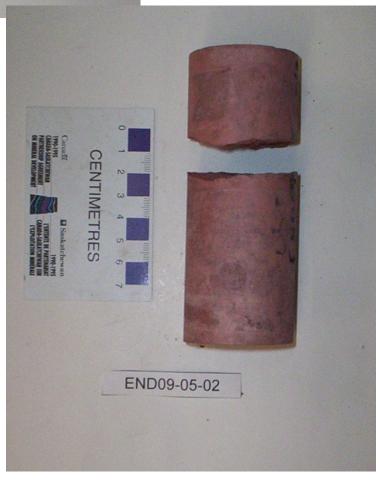
Density: N/A

Peak Strength: N/A

Young's Modulus: N/A

Poisson's Ratio: N/A

Failure Cause N/A



APPENDIX B RADIOACTIVE SAMPLES

SAMPLE: END09-02_210.24-210.42



Length: 94.451 mm

Diameter: 45.02 mm

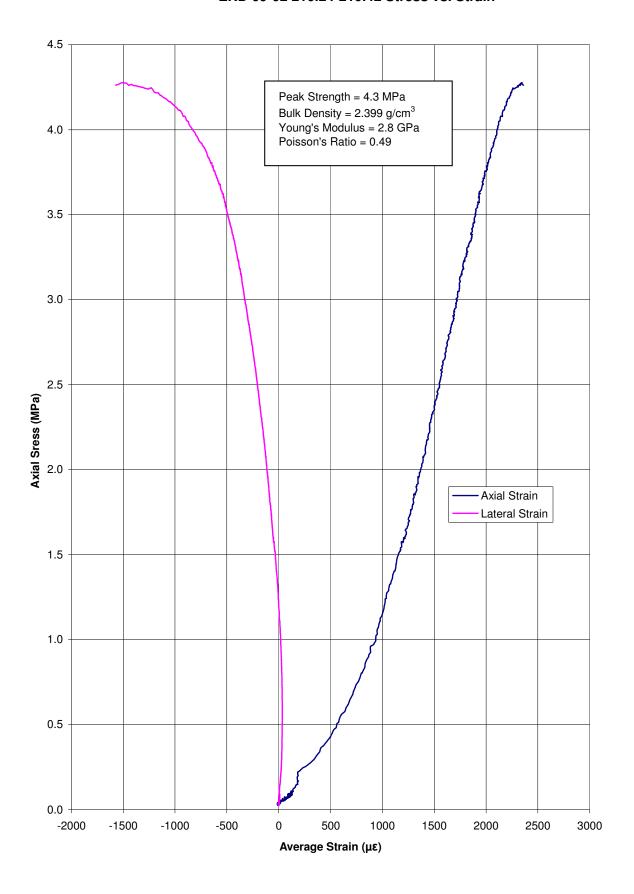
Density: 2.399 g/cm³

Peak Strength: 4.3 MPa

Young's Modulus: 2.8 GPa



END 09-02 210.24-210.42 Stress vs. Strain



SAMPLE: END09-02_215.48-215.67



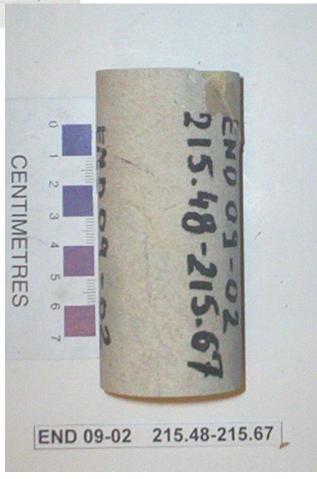
Length: 99.48 mm

Diameter: 45.05 mm

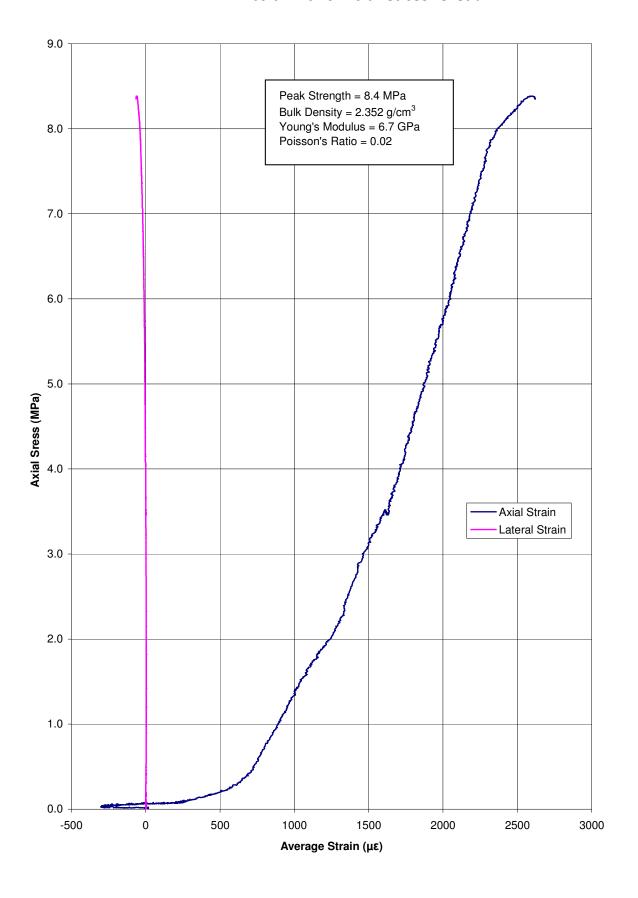
Density: 2.352 g/cm³

Peak Strength: 8.4 MPa

Young's Modulus: 6.7 GPa



END 09-02 215.48-215.67 Stress vs. Strain



SAMPLE: END09-02_234.67-234.89



Length: 98.07 mm

Diameter: 45.12 mm

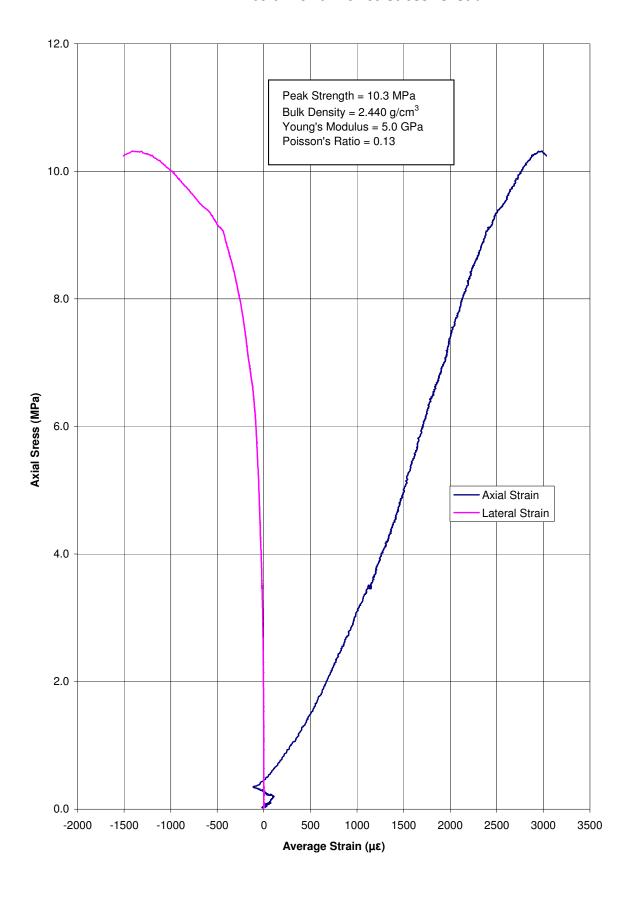
Density: 2.440 g/cm³

Peak Strength: 10.3 MPa

Young's Modulus: 5.0 GPa



END 09-02 234.67-234.89 Stress vs. Strain



SAMPLE: END09-02_246.92-247.04



Length: 102.86 mm

Diameter: 45.05 mm

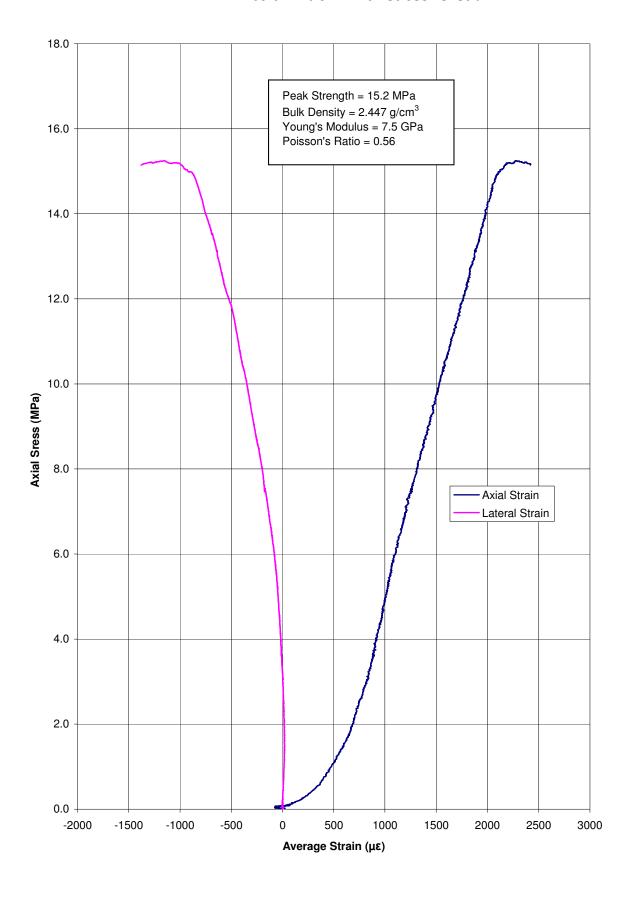
Density: 2.447 g/cm³

Peak Strength: 15.2 MPa

Young's Modulus: 7.5 GPa



END 09-02 246.92-247.04 Stress vs. Strain



SAMPLE: END09-02_272.81-272.95



Length: 100.68 mm

Diameter: 45.03 mm

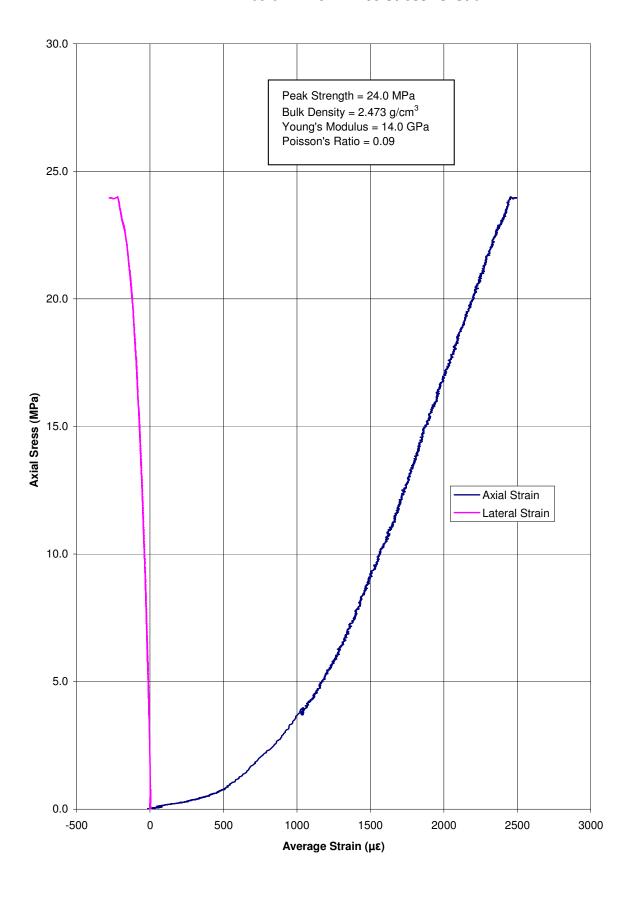
Density: 2.473 g/cm³

Peak Strength: 24.0 MPa

Young's Modulus: 14.0 GPa



END 09-02 272.81-272.95 Stress vs. Strain



SAMPLE: END09-02_284.00-284.30



END 09-02 284.00-284.30

Length: 98.16 mm

Diameter: 44.96 mm

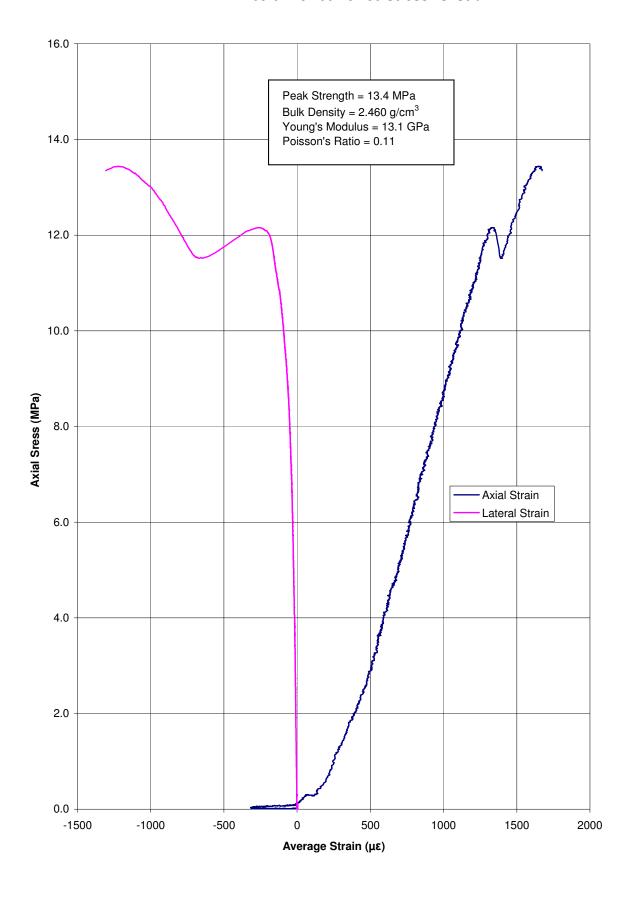
Density: 2.460 g/cm³

Peak Strength: 13.4 MPa

Young's Modulus: 13.1 GPa



END 09-02 284.00-284.30 Stress vs. Strain



SAMPLE: END09-02_302.10-302.43



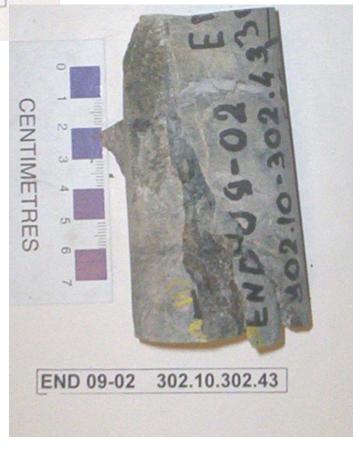
Length: 101.08 mm

Diameter: 45.08 mm

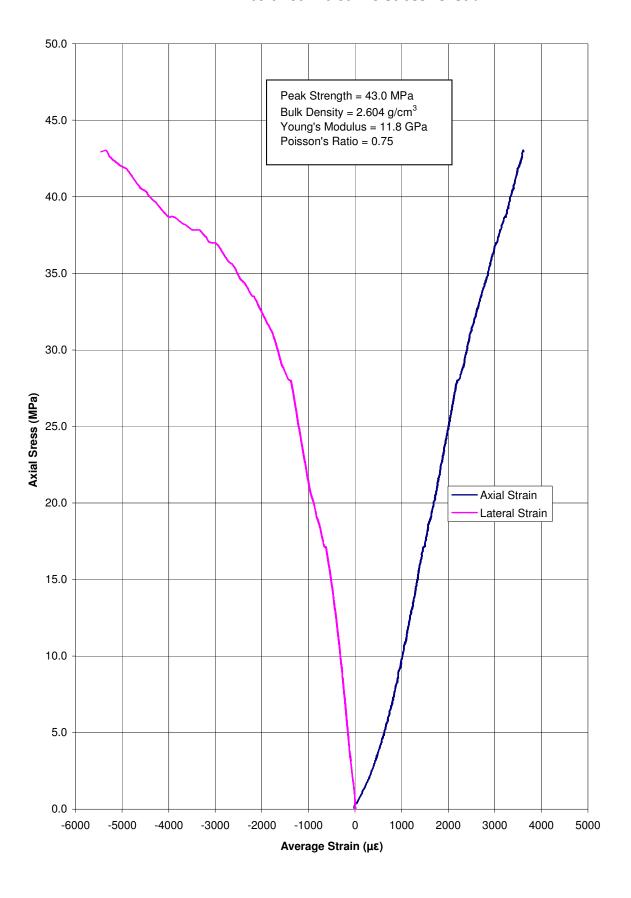
Density: 2.604 g/cm³

Peak Strength: 43.0 MPa

Young's Modulus: 11.8 GPa



END 09-02 302.10-302.43 Stress vs. Strain



SAMPLE: END09-02_312.35-312.55



Length: 102.08 mm

Diameter: 44.95 mm

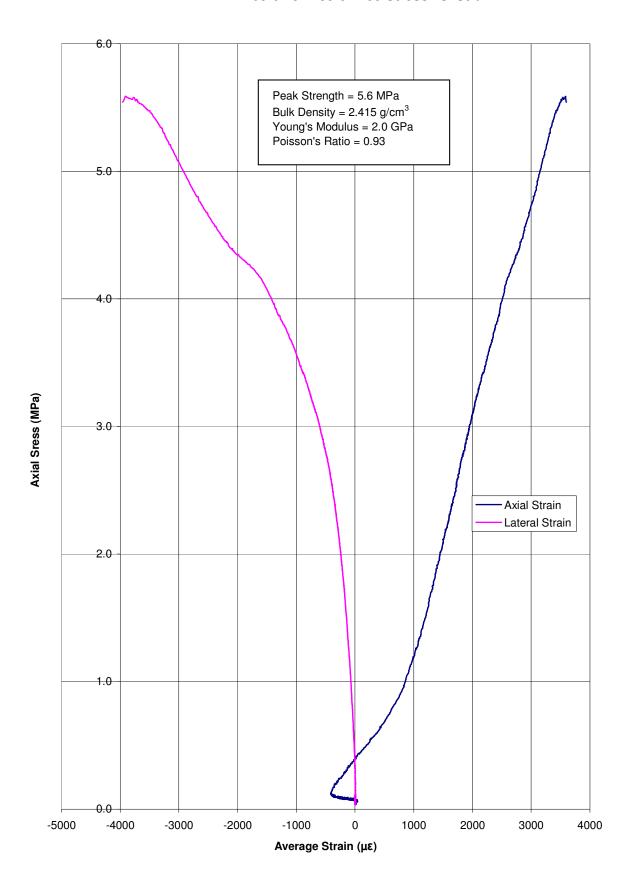
Density: 2.415 g/cm³

Peak Strength: 5.6 MPa

Young's Modulus: 2.0 GPa



END 09-02 312.35-312.55 Stress vs. Strain



SAMPLE: END09-02_313.95-314.12

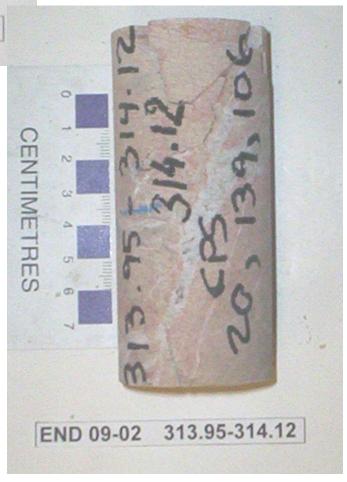


Length: 101.57 mm

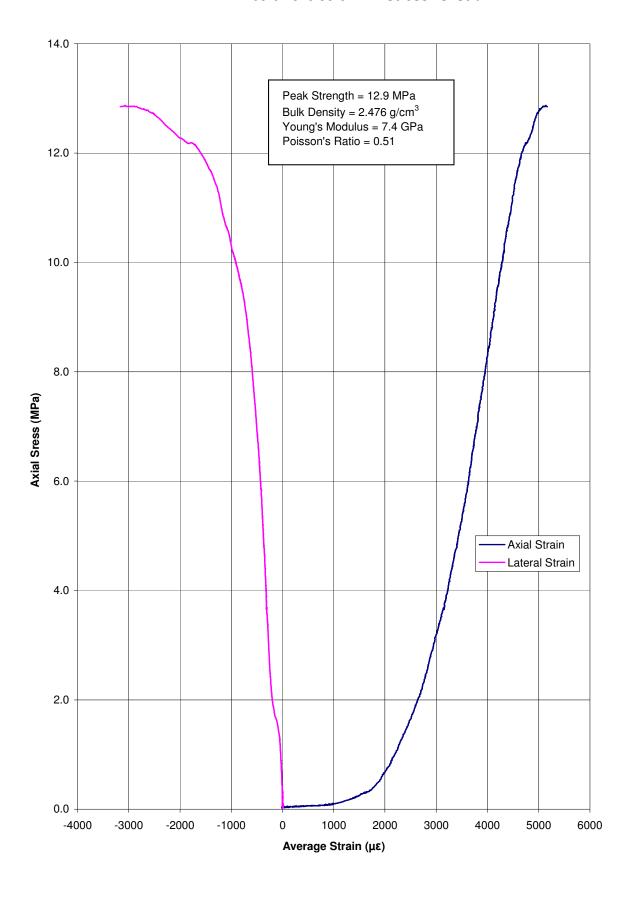
Diameter: 44.97 mm Density: 2.476 g/cm³

Peak Strength: 12.9 MPa

Young's Modulus: 7.4 GPa



END 09-02 313.95-314.12 Stress vs. Strain



SAMPLE: END09-02_329.30-329.50



Length: 103.53 mm

Diameter: 44.97 mm

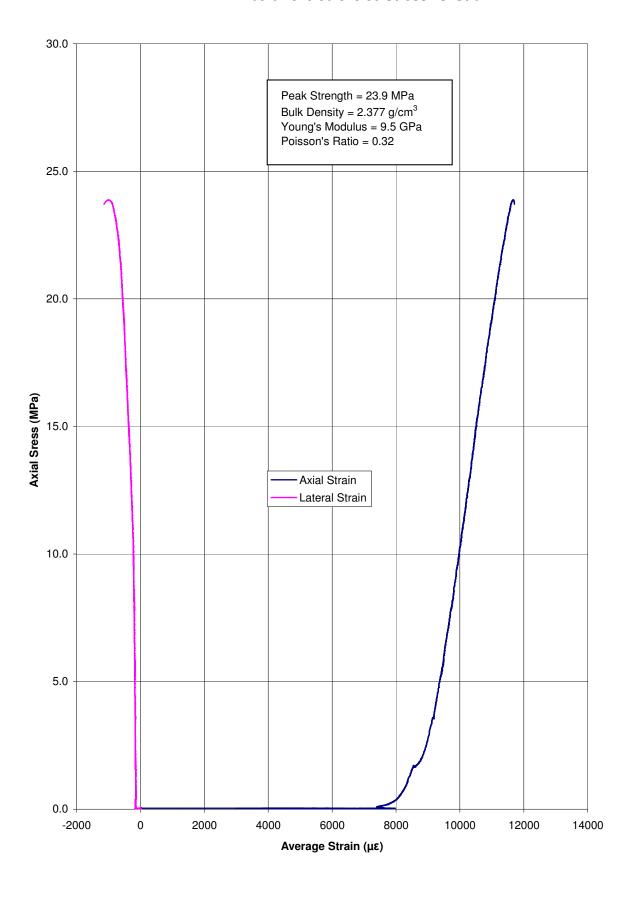
Density: 2.377 g/cm³

Peak Strength: 23.9 MPa

Young's Modulus: 9.5 GPa



END 09-02 329.30-329.50 Stress vs. Strain



SAMPLE: END09-02_333.25-333.42



END 09-02 333.25-333.42

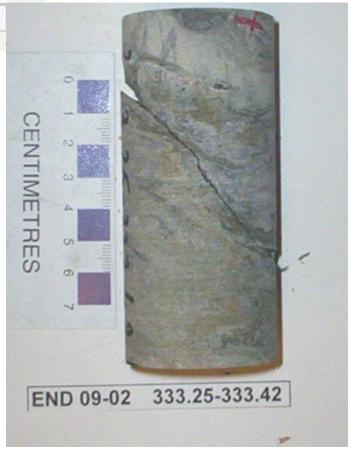
Length: 101.03 mm

Diameter: 45.13 mm

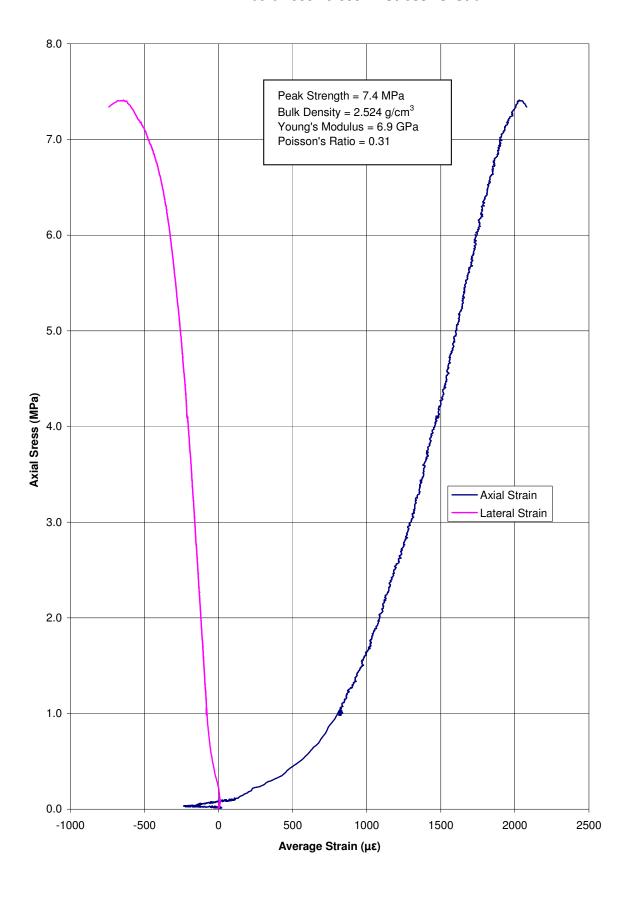
Density: 2.524 g/cm³

Peak Strength: 7.4 MPa

Young's Modulus: 6.9 GPa



END 09-02 333.25-333.42 Stress vs. Strain



SAMPLE: END09-02_341.20-341.40



END 09-02 341.20-341.40

Length: 100.13 mm

Diameter: 45.05 mm

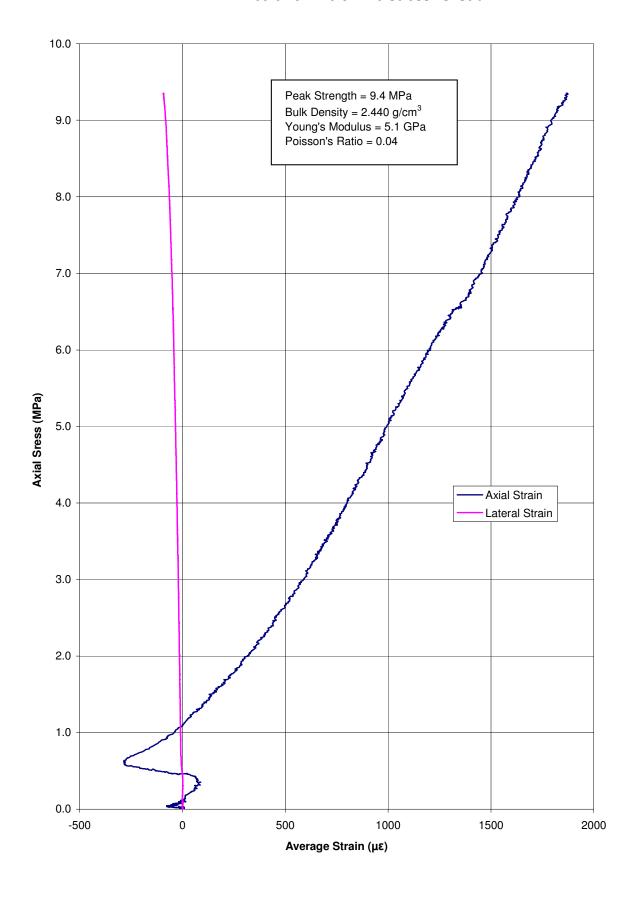
Density: 2.440 g/cm³

Peak Strength: 9.4 MPa

Young's Modulus: 5.1 GPa



END 09-02 341.20-341.40 Stress vs. Strain



SAMPLE: END09-02_291.40-291.50



Length: N/A

Diameter: N/A

Density: N/A

Peak Strength: N/A

Young's Modulus: N/A

Poisson's Ratio: N/A

Failure Cause N/A

SAMPLE: END09-02_319.50-319.60



Length: N/A

Diameter: N/A

Density: N/A

Peak Strength: N/A

Young's Modulus: N/A

Poisson's Ratio: N/A

Failure Cause N/A



APPENDIX B

Rock Mass Classification





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APPENDIX B - ROCK MASS CLASSIFICATION

1.0 INTRODUCTION

This appendix presents the results of the rock mass classification carried out on the 2009 geotechnical borehole data for Andrew Lake (AL). The data analysed in the rock mass classification work has been discussed in the previous appendices and the 2009 data report (Golder, 2009). In total, there was approximately 1000 m of core drilled at AL in the 2009 season.

The following appendix discusses the rock mass classification procedures, and outlines the rock mass classification results for the AL site, including interpretations for slope design.

2.0 ROCK MASS CLASSIFICATION PROCEDURE

For classification of the rock mass, the 1976 Rock Mass Rating (RMR) system was used (Bieniawski 1976). For validation of the RMR classification scheme, the Norwegian Geotechnical Institute Q-System (Barton et al. 1974) for rock mass classification was also used; however these results have not been presented herein.

Due to the variability in the rock mass quality for any particular rock lithology, the RMR was carried out at the logging interval level, meaning a separate classification was carried out on intervals ranging from 0.10 m to a maximum of once per drill run (i.e., typically 3 m of core or less). Using this approach, the RMR distribution in the borehole delineates intercepted geotechnical features, or zones in the rock mass which share similar geotechnical characteristics such as strength, fracture spacing, or alteration.

The RMR value per drilling interval is calculated as follows (on a scale of 100):

$$RMR = P_1 + P_2 + P_3 + P_4 + P_5$$

Where P₁ through P₅ represent values assigned based on the following rock mass parameters:

- Strength of Intact Rock (P_1) ; assessed on a per run basis according to either the UCS data if available, the PLT $I_{s(50)}$ data if available, or the field assessment of intact rock strength (R) based on the ISRM field classification. These strength parameters are discussed in Appendix A.
- Rock Quality Designation, RQD (P₂); this parameter was assessed in the field for every logging interval, as discussed in the data report (Golder, 2009).
- Fracture Spacing (P_3) ; this parameter was calculated based on the fracture frequency which was assessed in the field for every drill run of core as discussed in the data report (Golder, 2009).
- Condition of Joints (P_4); this parameter was taken as the average joint condition value (J_{con}) per drill run of core for the individual discontinuities assigned in the field (Golder, 2009).
- Groundwater (P_5); for the calculation of the rock mass RMR, the ground was assumed dry ($P_5 = 10$). For stability assessment considerations, the appropriate value will have to be subtracted from the calculated RMR values to reflect in situ seepage conditions, or in the case of numerical modelling, considered separately, such as a water table.





The RMR classification parameters are shown on Table B1.

Table B1: 1976 Rock Mass Rating (RMR) classification parameters (after Bieniawski, 1974).

A. CLASSIFICATION PARAMETERS AND THEIR RATINGS

	PARAM	ETER	RANGES OF VALUES							
	Strength of	Point load strength index	>8 MPa	4-8 MPa	2-4 MPa	I-2 MPa	- uniax	is low ial comp st is pre	pres- i	
	intact rock material	Uniaxial compressive strength	> 200 MPa	100 - 200 MPa	50 - 100 MPa	25 - 50 MPo	10-25 MPa	3-10 MPa	I-3 MPa	
L	Ro	ting	15	12	7	4	2	1	0	
2	Drill core qu	uality RQD	90%-100%	75% -90%	50%-75%	25%-50%		〈 25 %	,]	
2	Ra	ting	20	17	13	8	3			
,	Spacing	of joints	>3 m	. 1-3m	0.3-lm	50 - 300 mm (5		< 50 m	iO mm	
3	Ro	ting Ì	30	25 .	20	. 10		5		
4.	Condition of joints		Very rough surfaces Not continuous No separation Hard joint wall rock	Slightly rough surfaces Separation (1 mm Hard joint wall rock	Slightly rough surfaces Separation (1 mm Soft joint wall rock	Slickensided surfaces or Gouge <5 mm thick or Joints open 1 - 5 mm Continuous joints		open >	5mm	
	Rating		25	20	I2 ·	6		0		
		Inflow per IOm tunnel length	No OR	ne.	<25 litres/min.	25 - 125 litres/min.	> 12 OR	25 litres	s/min.	
5	Ground . water	Ratio joint water pressure major principal stress	0		0.0 - 0.2 OR	0.2-0.5		> O.5	·	
		General conditions	Complete		Moist only (interstitial water)	Water under moderate pressure		Severe r probl	ems	
	Ro	ating	IC)	7	4 0				

The calculated RMR values by drill interval for the AL 2009 boreholes are plotted on the striplogs included in Figure B1. Also shown on these striplogs are the lithology obtained from AREVA's geologists, and the ISRM strength index value (R) assessed in the field. As discussed in Appendix A, the R values agree reasonably well with the available laboratory testing. The RMR and R values have been color coded as per the legend shown on these figures. Records of Drillholes are presented in the factual data report (Golder, 2009).

Several geotechnical boreholes from 2007/2008 (SRK 2009) were also cross-checked following a similar rock mass classification approach using the geotechnical data provided by AREVA. The cross-checks showed that the calculated RMR (1976) values for the drill run intervals followed closely to previous rock mass classification carried out by SRK using the 1989 RMR system. The 2007/2008 data are not presented in this document, however some of these boreholes were used to aid the interpretation as discussed herein.



AND09-01 From To Geology R RMR 17 18 08 0 51.7 18 20 08 0 44.0 20 21.25 Geography (left) STRENGTH 0 to 20 20 to 40 80 to 100 R1: Very Weak Rock 40 to 60

AND09-02

А	ועטו	J9-UZ	_	
From	То	Geology	R	RMR
11.06 12	12 15	GnPsaPel LOST	2	52.0 31.1
15	18	GnPsaPel	2	38.3
18 21	21 24	GnPsaPel GnPsaPel	3	51.7 53.2
24	25.4	GnPsaPel	3	53.1
25.4 26.91	26.91 27	GnPsaPel LOST	3	13.9 53.4
27	30	GnPsaPel	3	55.2
30	33 35	GnPsaPel GnPsaPel	3	48.2 49.6
35	37	GnPsaPel	2	53.0
37 39	39 42	GnPsaPel GnPsaPel	2	57.4 47.1
42	45	GnPsaPel	2	52.9
45 47.37	47.37 48.5	GnPsaPel GnPsaPel	3	52.9 57.1
48.5	51	GnPsaPel	1	44.5
51 54	54 57	GnPsaPel GnPsaPel	2	50.6 40.0
57	57.15	GnPsaPel	2	35.9
57.15 58.5	58.5 60	GnPsaPel LOST	3	47.7 33.2
60	60.62	GnPsaPel	2	56.9
60.62 62.66	62.66 63	GnPsaPel LOST	2	40.8 56.3
63	66	GnPsaPel GnPsaPel	2	41.8
66 69	69 72	GnPsaPel	3	42.6 47.0
72	75	GnPsaPel	3	50.8 48.5
75 78	78 81	GnPsaPel GnPsaPel	3	53.7
81 84	84 87	GnPsaPel	3	54.3 61.1
87	90	GnPsaPel GnPsaPel	3	59.8
90 93	93 96	GnPsaPel GnPsaPel	3	48.2 53.5
96	97.1	GnPsaPel	3	65.7
97.1 99	99 102	GnPsaPel GnPsaPel	2	57.6 50.1
102	105	GnPsaPel	2	50.1
105 107.67	107.67 108	GnPsaPel GnPsaPel	2	51.7 56.9
108	111	GnPsaPel	2	56.0
111 114	114 117	GnPsaPel GnPsaPel	2	60.5 52.5
117	118.65	GnPsaPel	2	54.0
118.65 120	120 122.06	GnPsaPel GnPsaPel	3	57.6 57.2
122.06	123	GnPsaPel	2	45.7
123 126	126 129	GnPsaPel GnPsaPel	2	57.0 54.6
129	132	GnPsaPel	2	49.1
132 134.42	134.42 135.37	GnPsaPel GnPsaPel	0	51.9 26.1
135.37	138	GnPsaPel	3	60.6
138 139.64	139.64 140	GnPsaPel GnPsaPel	3	59.8 14.5
140	141	GnPsaPel	3	50.3
141	144 147	GnPsaPel GnPsaPel	3	62.0 63.5
147	148.55	GnPsaPel	3	38.1 59.7
148.55 150	150 153	GnPsaPel GnGran	2.5 2.5	63.7
153 154.06	154.06 156	GnGran GnGran	2.5	48.1 48.2
156	159	GnGran	3	55.8
159 162	162 165	GnGran GnGran	3	63.3 68.2
165	168	GnGran	3	41.6
168 169.05	169.05 171	GnPsa GnPsa	2.5 2.5	55.5 55.0
171	174	GnPsa	2.5	46.7
174 174.64	174.64 177	GnPsa GnPsa	1.5 2.5	36.5 59.5
177	178.42	GnPsa	2.5	60.4
178.42 178.8	178.8 180	GnPsa GnPsa	2.5 2.5	29.6 55.5
180	183	GnPsa	3	65.7
183 186	186 189	GnPsa GnPsa	3	67.4 61.1
189	192	GnPsa	3	70.5
192 195	195 197.12	GnPsa GnPsa	3	61.4 51.7
197.12	198	GnPsa	3	51.6
198 201	201 204	GnPsa GnPsa	3	58.1 78.9
204	207	GnPsa	3	59.9
207 210	210 213	GnPsa GnPsa	3	52.1 48.6
213 216	216 219	GnPsa GnPsa	3	60.6 52.6
219	221.4	GnPsa	2.5	47.6
221.4 222.86	222.86 223.86	GnPsa GnPsa	2.5	52.9 53.6
223.86	224.6	GnPsa	0	25.8
224.6 225	225 228	GnPsa GnPsa	2.5 2.5	46.6 52.3
228	228.62	GnPsa	2.5	45.4
228.62 229.13	229.13 230.34	GnPsa LOST	2	27.1 41.6
230.34	231	GnPsa	2	50.9
231 234	234 237	GnPsa GnPsa	3	47.3 58.0
237	240	GnPsa	3	70.7
240 243	243 246	GnPsa GnPsa	3	61.9 68.2
246 249	249 252	GnPsaPel GnPsaPel	3	66.4 60.5
252	255	GnPsaPel	1	48.1
255 258	258 261	GnPsaPel GnPsaPel	2	55.2 41.8
261	264	GnPsaPel	2	36.2
264 267	267 270	GnPsaPel GnPsaPel	5 5	74.7 69.4
270	273	GnPsaPel	5	74.1
273 276	276 279	GnPsaPel GnPsaPel	5 5	80.8 73.7
279	282	GnPsaPel	5	79.8
282 285	285 288	GnPsaPel GnGran	5 5	71.7 85.4
288	291	GnGran	5 5	73.6
291 294	294 297	GnGran GnGran	5	85.4 80.0
297 300	300	GnGran	5 5	72.1
300 303	303 306	GnGran GnGran	5	66.4 72.1
306 309	309 312	GnGran GnGran	5 5	90.1 74.0
309 312	312 315	GnGran GnGran	5	74.0 74.8
315 318	318	GnGran	5 5	80.4
318 321	321 324	GnGran GnGran	5 5	84.8 83.8
324	327	GnGran	5	78.3

AND09-03

	MINL	709-0	<u> </u>	
From 7.5	To 9	Geology Episyenite	R 4	57.4
9	12 15	Episyenite	4	62.8
15	18	Episyenite Episyenite	4	61.8 57.5
18	18.9	Episyenite	3	58.3
18.9	20.55	GnPsaPel		59.3
20.55	21.9	GnPsaPel	4	66.6
	23.35	GnPsaPel	3	58.6
23.35	25.7	GnPsaPel	3	46.4
25.7	27	GnPsaPel	3	45.4
27	30	GnPsaPel		51.3
30	33	GnPsaPel	3	67.9
33	36	GnPsaPel		62.5
36	39	GnPsaPel	3	67.2
39	42	Episyenite		63.5
42	45	GnPsaPel	3	61.0
45	48	GnPsaPel	3	57.1
48	51	GnPsaPel		61.4
51	54	GnPsaPel	3	54.0
54	57	GnPsaPel		52.0
57	60	GnPsaPel GnPsaPel	3	59.1 59.7
60 63	63 66	GnPsaPel	3	48.7
66	69	GnPsaPel	3	50.8
69	71.53	GnPsaPel		50.6
71.53	72	GnPsaPel	3	55.5
72	75	Episyenite		56.6
75	78	Episyenite	3	50.1
78	81	Episyenite	3	42.7
81	84	Episyenite		61.3
84	87	Episyenite	3	58.4
87	90	Episyenite		45.7
90	93	Episyenite	3	56.2
93	96	Episyenite		59.8
96	99	Episyenite	3	63.0
99	102	Episyenite	3	55.9
102	105	GnPsaPel		59.5
105	106.54	GnPsaPel	3	66.1
106.54	108	GnPsaPel		55.4
108	111	GnPsaPel	3	56.6
111	112.02	GnPsaPel	0	49.7
112.02	114	GnPsaPel		14.1
114	117	GnPsaPel	1.5	37.2
117	120	GnPsaPel	2	49.9
120 123	123	GnPsaPel GnPsaPel	2	53.9 52.5
126	126 129	GnPsaPel	3	55.4
129	129.56	GnPsaPel	3	49.2
129.56	129.98	Episyenite		56.6
129.98	132	Episyenite	3	49.1
132	135	GnPsaPel		60.6
135 138	138 141	GnPsaPel	2	56.5 54.7
141	141.86	GnPsaPel Episyenite	2	55.0
141.86	143.37	Episyenite	3.5	58.2
143.37	144	GnPsaPel	2	61.2
144	147	GnPsaPel	2	55.0
147	148.36	GnPsaPel		51.5
148.36	150.19	GnPsaPel	2	45.5
	151.06	GnPsaPel	1.5	25.6
150.19 151.06	153	GnPsaPel	2	50.0
153	156	GnPsaPel	2	54.2
156	159	GnAlt		56.9
159	162	GnAlt	2	68.0
162	165	GnAlt		66.2
165	168	GnPsaPel	2	67.6
168	171	GnPsaPel	3	48.1
171	174	GnPsaPel		64.9
174	177	GnPsaPel	3	65.7
177	180	GnPsaPel		53.0
180	183	GnPsaPel	3	60.0
183	184	GnPsaPel		63.5
184	186	GnPsaPel	3	64.9
186	189	GnPsaPel	3	61.5
189	192	GnPsaPel		61.2
192	195	GnPsaPel	3	58.8
195	198	BxQtz		56.8
198	199.29	GnPsaPel	3 2	63.1
199.29	201	BxQtz		41.6
201	204	BxQtz	2	48.1
204	206.03	BxQtz	2	37.5
206.03	207	GnPsaPel	4	69.1
207	210	GnPsaPel	4	56.0
210	213	GnPsaPel		64.1
213	216	GnPsaPel	4	64.2
216	219	GnQtzFd		65.3
219	222	GnPsaPel	4	62.3
222	225	GnPsaPel	4	72.9
225	228	GnPsaPel		71.6
228	231	GnPsaPel	4	66.7
231	234	GnPsaPel		71.4
234	237	GnPsaPel	4	78.0
237	240	GnPsaPel	4	76.2
240	243	GnPsaPel	4	73.1
243	246 249	GnPsaPel GnPsaPel	4	75.2 51.2
249	252	FLT	4	86.8
252	255	FLT		67.3
255	258	GranT	4	71.7
258	261	GranT		68.1
261	264	GranT	4	77.1
264	267	GranT	4	71.5
267	270	GranT		75.0
270	273	GranT	4	72.4
273	276	GranT	4	72.0
276 279	279 282	QtzVein	4	72.4 61.8
282	285	GranT GranT	4	60.8
285	288	GranT	4	64.1
288	291	GranT		53.2
291	294	GranT	4	65.8
294	296.5	GranT		62.0
296.5 299.5	299.5 302	GranT	4	73.0 55.8
302	303.58	GranT BxQtz	1	48.2
303.58	304.41	GranT	3	48.8
304.41	305.7	GranT		45.4
305.7	308.26	GranT	4	39.8
308.26	309.36	GranT		38.1
309.36	312	GranT	4	66.4
312	313.2	GranT	3	57.5
313.2	313.95	GranT		49.4
313.95	315	GranT	4	59.8
315	318	GranT		62.5
318	319.55	GranT	4	64.7
319.55	319.85	GranT		48.5
319.85	321.15	GranT	3	55.0
321.15	324	GranT	2.5	63.0
324	327	GranT	3	71.7

Geology provided by AREVA.
See Appendix A for lithology code descriptions.

R3: Medium Strong Rock

R5: Very Strong Rock



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GnGran

Proj. No.: 08-1362-0613

Date: November 2009

Drawn: BC

Check: MR

ANDREW LAKE 2009 HOLES ROCK MASS RATING (RMR) AND STRENGTH (R)

Kiggavik

Figure B1



3.0 ROCK MASS CLASSIFICATION RESULTS

The 2007/2008 and 2009 geotechnical borehole rock mass rating (RMR) and rock strength index (R) values were plotted in the 3D Surpac geology model for the Andrew Lake pit. These models include the lithology, mineralization and faulting structure data provided by AREVA. On these sections, the boreholes show the R values (left bar), the lithology (middle bar), and the RMR values (right bar). Several sections were plotted across the pit boundary and some general interpretation was carried out to delineate geotechnical domains.

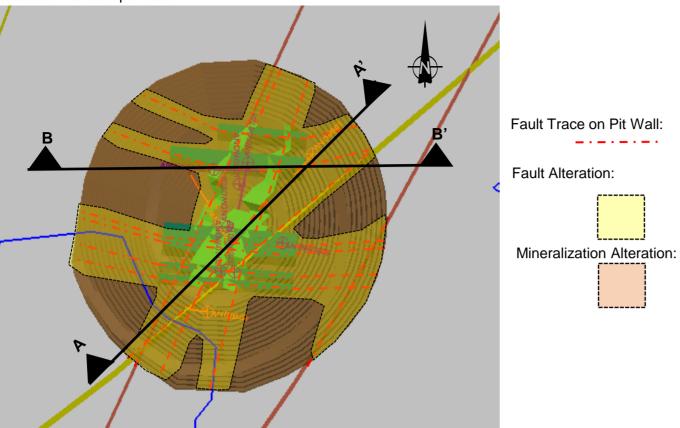
The Andrew Lake geology is complicated by the faulting and mineralization horizons which appear to coincide with mineralogical alteration, leading to weakening and reduced quality in the rock mass. Plotting these inferred fault and mineralization alteration zones on the sections tends to coincide with the lower quality zones in the borehole data. Currently, the rock mass conditions in the majority of the planned pit walls is uncertain, although it is inferred that the rock mass quality improves with increased distance away from the mineralization zones.

Figure B2 presents geological sections with the inferred alteration zones and lithological domains on the Andrew Lake pit shell.

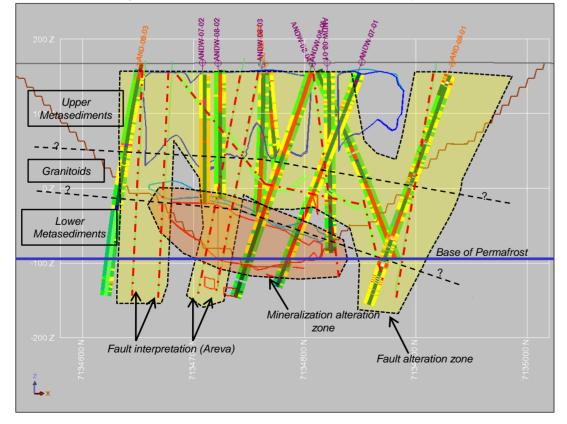
Figure B3 (from Figure B2) presents a 2D section taken SW-NE across the Andrew Lake pit. Several faults transecting the pit are shown in this section, as well as a fairly significant fault trending near parallel to the section trend. The rock associated with these faults and alteration zones tend to range from fair to good quality with some infrequent zones of poor quality, and are generally weak to moderately strong. The lithological units away from the central mineralization and faulting region are uncertain. As seen in the section, a three tier lithology arrangement has been generalized which includes; upper metasediments underlain by granitoids, underlain by lower metasediments/paragneiss. This interpretation loosely follows the geological model presented in the SRK geological model update (SRK 2008). A geological section taken from this document, showing the layout of the ore zone and lithological domains, is presented in Figure B4. Comparing Figures B3 and B4, the upper ore zones have not been clearly delineated in the AREVA geology model.



a) Andrew Lake - pit plan view showing faulting and inferred fault alteration zone on pit walls.



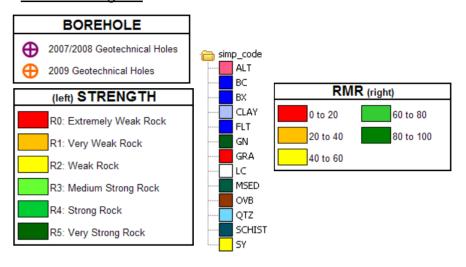
b) Section A – borehole projections of strength (R) and RMR with inferred alteration zones and lithological domains. This section is predominantly fault influenced.



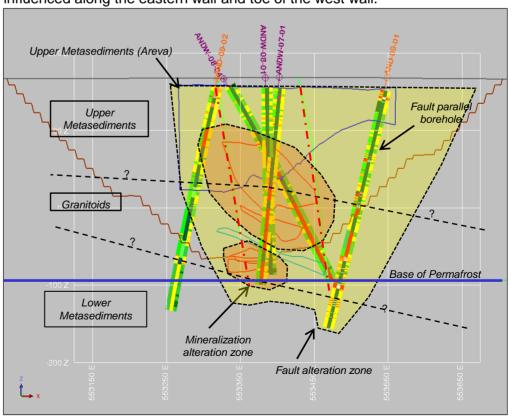
ANDREW LAKE PIT GEOLOGICAL SECTIONS WITH INFERRED ALTERATION ZONES AND LITHOLOGICAL DOMAINS

FIGURE B2

Borehole Legend



c) Section B – borehole projections of strength (R) and RMR with inferred alteration zones and lithological domains. This section is predominantly fault influenced along the eastern wall and toe of the west wall.



Date: November 2009

Project: 09-1362-0613

Drawn: BC Chkd:: MR

Golder Associates



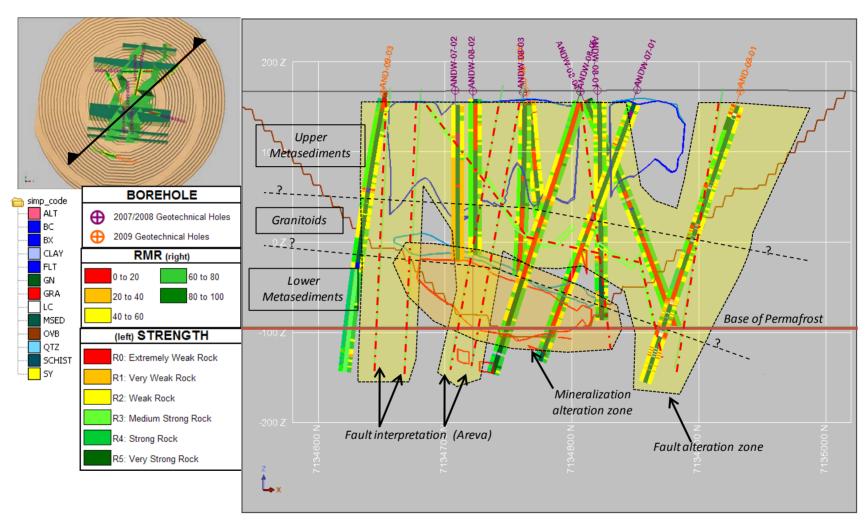


Figure B3: Andrew Lake Pit - Generalized Geotechnical Section (SW-NE). Inferred Fault and Mineralization Alteration Zones are Highlighted.





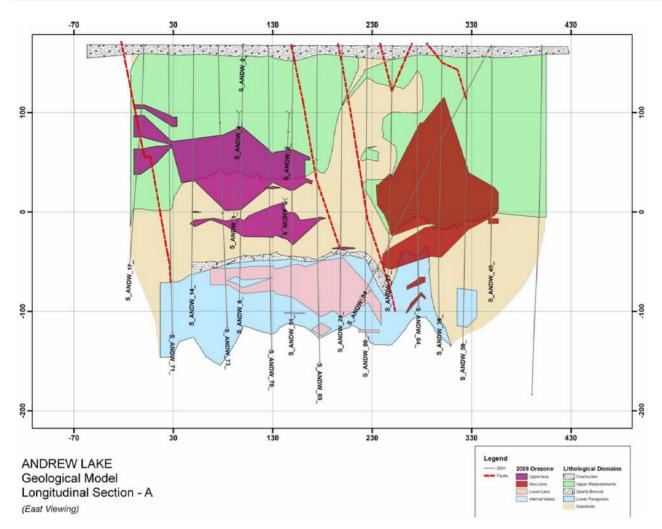


Figure B4: Andrew Lake Pit - Geological Model of Ore Zones and Lithological Domains (SRK, 2008)

Both AND09-1 and AND09-03 are plotted on Figure B3. Borehole AND09-01 appears to follow very closely to both an E-W trending fault and the major NE-SW trending Andrew Lake fault. The overall quality and strength of the rock mass intersected by this borehole are lower than the other 2009 boreholes, and rock mass conditions do not improve with depth. The poor quality of this hole is assumed to be due to the effect of alteration and faulting at close proximity to the hole.

Borehole AND09-03 shows fair quality, weak to moderately strong conditions in the initial 200 m identified as metasediments. This zone is possibly fault influenced as it trends in close proximity to the NE-SW trending fault structures. Below 200 m the rock is mainly good quality, moderately strong to strong rock identified as granitic. A lower quality zone is intersected at 300 m to 310 m, possibly associated with a second fault structure.





Figure B5 presents a 2D section taken W-E across the Andrew Lake pit, towards the north end wall. Again the inferred mineralization appears to be associated with lower quality and weaker ground conditions. Borehole AND09-02 is plotted on this section. The upper 180 m of the borehole is shown to be generally fair quality, and generally weak, identified as metasediments. This zone is possibly fault influenced. From 180 m to 264 m, the rock shows similar fair quality with a slight improvement in strength, also identified as metasediments. The borehole appears to be trending towards an E-W trending fault in this zone. Below 264 m, the rock is predominantly good quality and strong, identified as granitic gneiss.

The 2009 Andrew Lake RMR data has been plotted in cumulative frequency distribution diagrams as shown on Figures B6 through B8. These plots were generated by summing the core lengths within given RMR ranges, and taking these sums as the statistical percentage of the overall drilled length. The RMR distributions exemplify the potential variability in rock mass conditions that could be encountered in the rock mass. The statistically significant RMR values have been taken as the Lower Bound (20% cumulative), Average (50% cumulative), and Upper Bound (80% cumulative) distributions. This provides a probable range of RMR values comprising the rock mass.

Figure B6 shows the cumulative frequency distribution of RMR values by borehole. The associated lower bound, upper bound, and average RMR values are summarised in Table B2. In general, hole AND09-01 shows the lowest relatively quality of the 3 holes, likely due to its association with the NE-SW faulting. The statistically significant range in RMR values is between 40 (fair to poor) to 68 (good), although locally RMR values are shown to range between 10 and 85. Average RMR values for the three holes are fairly consistent at between 54 and 58 (fair).





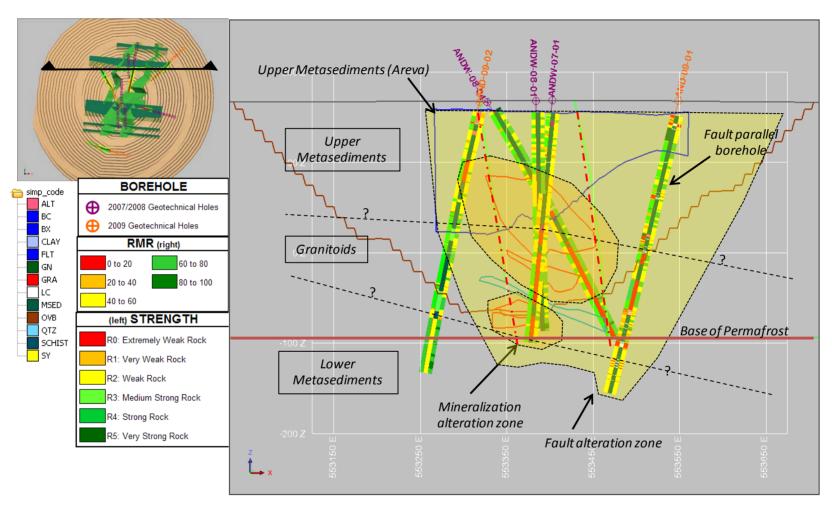


Figure B5: Andrew Lake Pit - Generalized Geotechnical Section (W-E). Inferred Fault and Mineralization Alteration Zones are Highlighted.





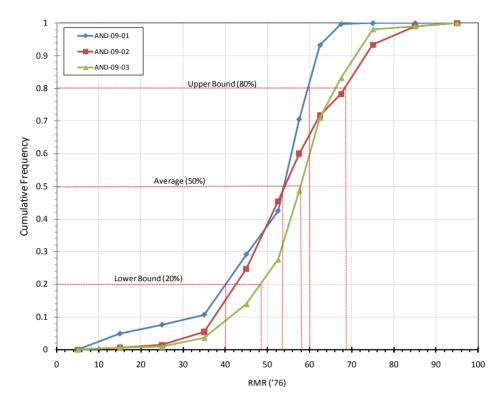


Figure B6: Andrew Lake - 2009 Boreholes. Cumulative Frequency Distribution of RMR by Borehole.

Table B2: Andrew Lake – 2009 Boreholes. Summary of RMR values by borehole.

Hole	RMR (1976)					
11010	Lower Bound	Upper Bound	Average			
AND09-01	40	60	54			
AND09-02	42	68	54			
AND09-03	48	66	58			

The RMR values are plotted by rock type in Figure B7, with corresponding values summarised in Table B3. The rock units have been summarised into metasediments at varying depths (less than 100 m, 100 m to 200 m, and greater than 200 m), as well as granites. It is noted that these depths are downhole depths. It is shown that the upper 200 m of metasediments shows fairly similar distribution (average RMR = 52 to 54), but for the data below 200 m, the average quality increases considerably (average RMR = 68). This increase in quality with depth is possibly both associated with the rock types, as well as increased offset from the fault and mineralization zones. A trend of increasing strength with depth below 200 m was also noted in Appendix A. The granites show slightly improved quality compared to the upper metasediments, with an average RMR of 58.





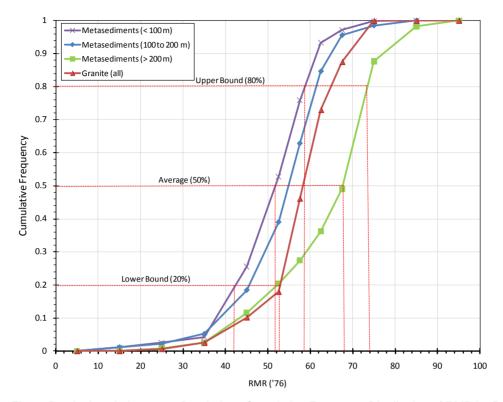


Figure B7: Andrew Lake - 2009 Boreholes. Cumulative Frequency Distribution of RMR by Rock Type.

Table B3: Andrew Lake – 2009 Boreholes. Summary of RMR values by rock type.

Lithology	Depth	RMR (1976)			
Limbiogy	Бериі	Lower Bound	Upper Bound	Average	
	<100 m	42	58	52	
Metasediments (GnGran,GnPsa,GnPsaPel)	100 to 200 m	45	62	54	
	>200 m	52	74	68	
Granite (GranT)	all	52	65	58	

GnGran, GnPsa, GnPsaPel = Granitic Gneiss, Psammitic Gneiss, Psammo-pelitic Gneiss; GranT = Granite

The Andrew Lake RMR data is again plotted by alteration in Figure B8, with RMR values summarised in Table B4. The alteration factor (Af) has been used, which is a product of both the argillization rating ($A_{arg} = 0$ to 4) and chloritization rating ($A_{chl} = 0$ to 4) used by AREVA. The alteration factor is discussed in Appendix A. The Andrew Lake rock was divided into 3 alteration domains; nil to minor alteration (Af = 0 to 1), moderate alteration (Af = 2 to 3), and highly altered (Af \geq 4). As expected, there is shown to be a decrease in rock mass quality with increased alteration. However, the overall range between nil to highly altered rock is not excessive, with average RMR ranging between 50 and 58. There was shown to be a more prominent variability in rock strength with increased alteration, as discussed in Appendix A.





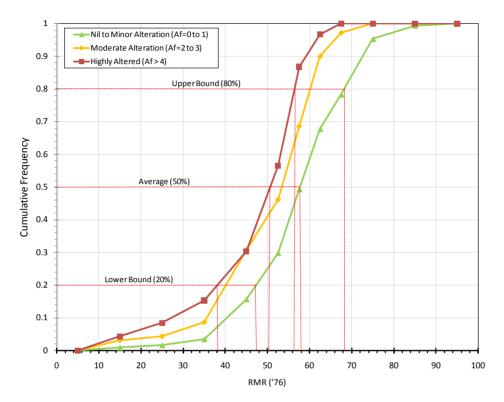


Figure B8: Andrew Lake - 2009 Boreholes. Cumulative Frequency of RMR by Alteration Factor (Af)

Table B4: Andrew Lake – 2009 Boreholes. Summary of RMR values by alteration factor (Af).

Alteration	RMR (1976)					
	Lower Bound	Upper Bound	Average			
Nil to Minor (Af = 0 to 1)	48	68	58			
Moderate (Af = 2 to 3)	40	60	54			
Highly Altered (Af >= 4)	38	56	50			

4.0 CONCLUSIONS AND RECOMMENDATIONS

The proceeding sections have presented the methodology and results of the rock mass classification work carried out for the 2009 geotechnical data for the Andrew Lake pit. Generalized interpretations were made based on the available information. Appreciably more geotechnical information should be collected to make better interpretations on rock mass quality and strength, predominantly for the final planned pit walls, where currently data is lacking.

The Andrew Lake rock mass is interpreted to be mainly comprised of metasediments, granitics, and lower metasediments or paragneiss. Mineralization and faulting related alteration halos appear to reduce the quality and strength of the rock in close proximity to this alteration. Reduced strength and quality will occur on the final pit floor, and the lower slopes of the pit walls. Potentially more significant, fault related alteration reduced strength and quality appears likely along the pit walls in various locations as shown on Figure B2.





Recommendations for RMR and strength parameters for the main rock units at Andrew Lake are given in Table B5. Assumptions related to the recommended values are also given in the table.

Table B5: Andrew Lake – Recommendations for RMR and strength parameters.

Rock Unit	RMR (1976)	Comment	Strength	Comment
Upper Metasediments (<200 m depth)	42 to 62 (fair)	Range of RMR for lower to upper bound limits for the 2009 Upper Metasediment data. Average RMR of nil to moderately altered ground.	R2/R3 (weak to moderately strong)	Average UCS = 29.9 MPa and average Is(50) = 2.2 MPa for upper metasediments (Appendix A)
Granites (all)	52 to 65 (fair to good)	I 2009 Granite data		Average UCS = 66.5 MPa and Is(50) = 5.5 MPa (Appendix A)
Lower Metasediments/ Paragneiss (>200 m depth)	52 to 74 (fair to good)	Range of RMR for lower to upper bound limits for the 2009 Lower Metasediments data. Average RMR of nil to slightly altered ground.	R4 (strong)	Average Is(50) = 7.3 MPa (Appendix A)
Fault or Mineralization Altered Zones	42 to 52 (poor to fair)	Lower Bound RMR of all 2009 lithology data. Average RMR of 2009 highly altered data.	R2 (weak)	Average UCS = 7.4 to 24.7 MPa for moderate to highly altered rock or fault related zones (Appendix A)

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APPENDIX C

Stereonet Analysis from Oriented Core Data





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1.0 INTRODUCTION

This appendix outlines the methodology used to select the major and minor discontinuity sets within the rock masses at Andrew Lake.

Details of the 2009 geotechnical drilling program that was conducted at Andrew Lake have been presented in a geotechnical data report prepared by Golder entitled "2009 Kiggavik Geotechnical and Hydrogeological Investigation Data Report". This data report outlines the drilling program conducted, and presents the data collected during the 2009 season. As reference, a total of 3 holes were oriented during the 2009 investigation at Andrew Lake. Oriented boreholes are shown on Figure C1, along with oriented boreholes logged by SRK Consulting (SRK) in 2008 for reference, and are summarized on Table C1.

Table C1: Summary of Oriented Boreholes from 2007, 2008, and 2009 Andrew Lake

Borehole #	Year Drilled ^(a)	Northing	Easting	Collar Elevation (masl)*	Azimuth	Dip	Drill Depth (mAH)	Vertical Depth (mbgs)	Bit Size
AND09-01	2009 (G)	7134927	553548	167.94	228	-70	342	321	NQ3
AND09-02	2009 (G)	7134809	553319	167.07	330	-65	330	299	NQ3
AND09-03	2009 (G)	7134574	553312	166.21	284	-70	327	307	NQ3
ANDW-08-04	2008 (SRK)	7134893	553326	168	092	-60	300	260	HQ3

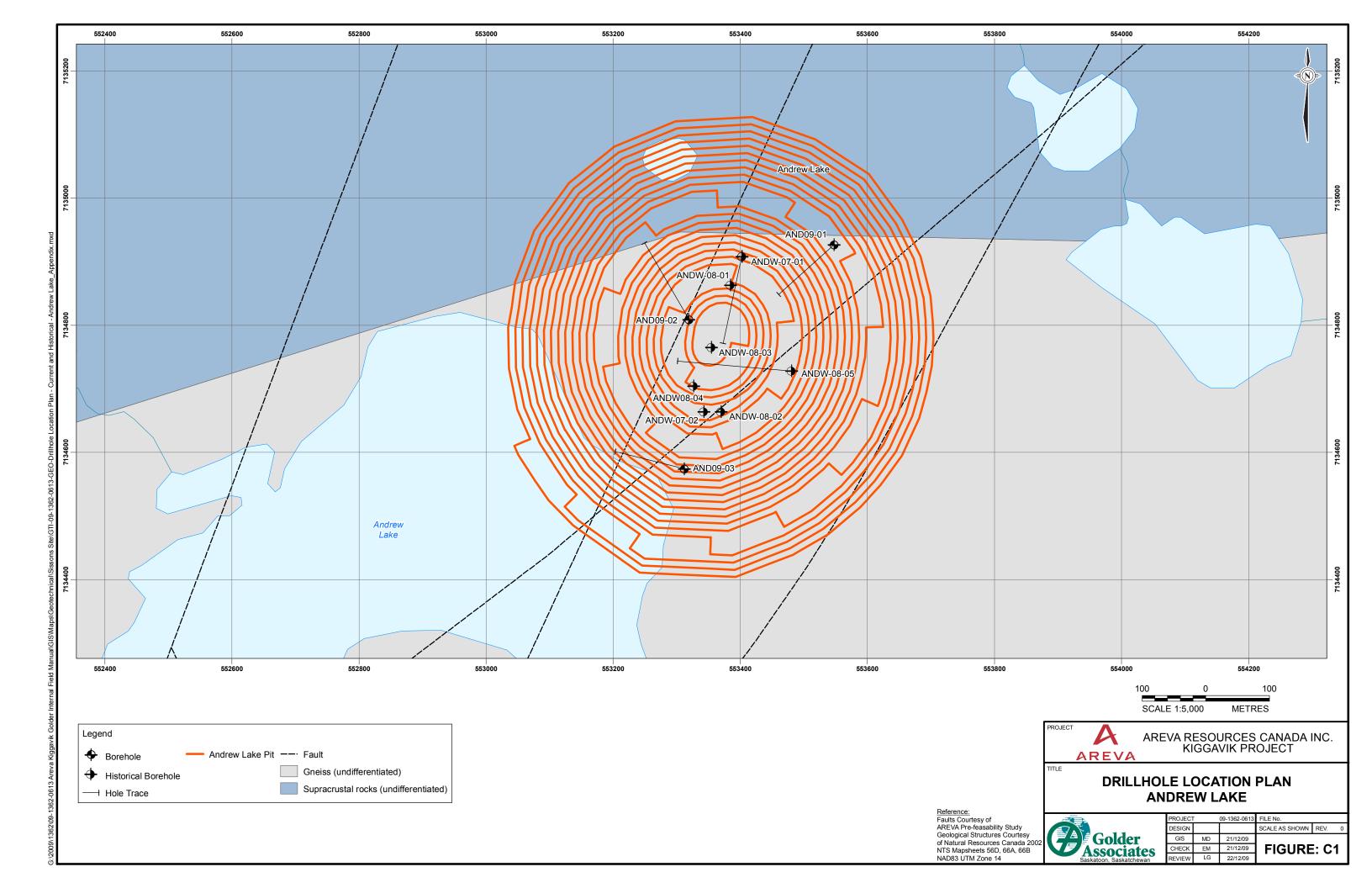
A = G = Golder (2009), SRK = SRK (2007/2008); Coordinates in UTM NAD 83 Zone 14, Collar elevation for 2009 boreholes are an estimate based on point data collected from a LiDAR survey, masl = metres above sea level; mAH = metres along hole; mbgs = metres below ground surface

2.0 CORE ORIENTATION METHODOLOGY

In 2009, core orientation was undertaken by the drilling contractor using an Ace Core Orientation Tool (ACT), made by Reflex Instruments. ACT was a fully electronic system that used accelerometers to reference the low side of the borehole (Reflex, 2009). This information was then used by the drill staff to place a reference mark at the bottom of each drill run which corresponded to the low side of the borehole.

During the core logging process, Golder personnel used the driller's reference mark to scribe an orientation line on the low side of the drill core along the length of the drill run. A good match of reference lines between consecutive drill core runs was considered valid with 'high confidence'. When the orientation lines varied at approximately 40 to 60 degrees from one another, this data was considered valid with 'moderate confidence'. In some cases, poor core conditions in zones of highly fractured or altered core prevented the line from being extended the length of the drill run and the core orientation line was lost. This frequently occurred for the Andrew Lake holes as the drill core was often broken and damaged in the weaker ground. When there was no match between orientation lines, or no continuity between oriented zones, the data was considered invalid or 'low' confidence.







Using the orientation line, alpha and beta angles were measured for each logged discontinuity. Alpha angles were a measurement of the apparent dip of the feature with respect to the core axis. These measurements were taken at the steepest part of the discontinuity, and ranged from 0 degrees (i.e., horizontal to the core axis) to 90 degrees (i.e., perpendicular to the core axis). Beta angles were the measurement of the apparent trend of the discontinuity, with respect to the orientation reference line. Beta angles were measured using linear protractors, where the zero degree mark was held on the reference line, with the measurement being taken in a clockwise direction on the "down dip" end (i.e., bottom of the discontinuity when looking downhole) of the discontinuity surface. Measurements ranged between 0 to 360 degrees. A summary of all orientation measurements taken from the 2009 boreholes at Andrew Lake are shown in Table C2. Due to the drilling conditions and relative rock mass quality, there was generally a low percentage of data which was validated for the Andrew Lake holes. The use of this data is discussed further in the following sections.

Core drilled in 2007 and 2008 was oriented using ACT in a similar manner to that used in 2009 (based on SRK, 2009).

Table C2: Summary of Oriented Features from 2009 Boreholes at Andrew Lake

2009 Borehole	# of Features Oriented	% of Features Oriented	# of Intervals Oriented	% Intervals Oriented
AND09-01	192	17%	34	17%
AND09-02	514	39%	65	48%
AND09-03	537	42%	67	49%

3.0 ASSESSMENT OF ROCK MASS FABRIC

3.1 Major Geological Structure

The major geological structures at Andrew Lake have been summarized below:

3.1.1 Fault Trends

At the Andrew Lake deposit, the regional Andrew Lake Fault, which is steeply dipping and trends northeast at 030 degrees (AREVA 2007), was assumed to lie within the footprint of the proposed pit. It is possible that four dominant north-northeast trending faults mapped from historical core data are related to the Andrew Lake fault. The four mapped fault features appear to control the lateral extents of the uranium mineralization at Andrew Lake, and strike in the same orientation as the mineralization (SRK 2008). At Andrew Lake, there is also evidence of structural features which cross-cut these faults, although there is limited drill core data to support this (SRK 2008). It is estimated that these cross-cutting features run sub-parallel to the north-northeast trending faults, and may have an influence on the mineralization at Andrew Lake (SRK 2008). It has been estimated these cross cutting features dip steeply to the southwest (SRK 2008).

The inferred faults provided in the geological model supplied by AREVA, as well as the regional fault trends for Andrew Lake are plotted on Figure C2 below. A number of fault structures are inferred to intercept the pit walls and floor at Andrew Lake, mainly trending north-south to northeast-southwest, or orthogonal to this trend striking east-west. Again there is not always agreement between the fault interpretations as given in the geological model, and the regional fault trends, which were provided by AREVA and based on geophysical anomalies identified. Future work should aim to clarify the locations and orientations of the major fault structures.





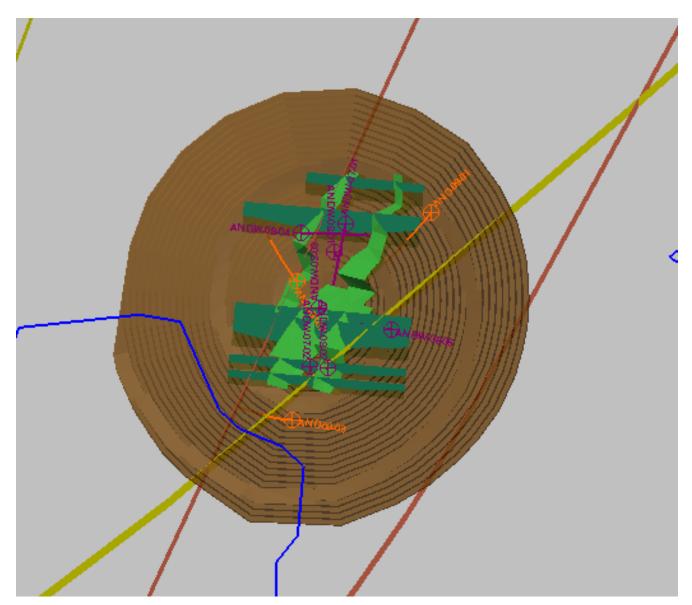


Figure C2: Andrew Lake Pit Showing the Inferred Faults from the Geological Model (green) and the Regional Fault Trends (Beaudemont Faults) Cross-cutting the Pit Walls (brown and yellow)

3.1.2 Foliation/Bedding

At Andrew Lake, the identified foliation trend dips at 20 to 50 degrees towards the east or south-east (see Figure C3). This trend, possibly related to bedding in the metasedimentary rock units, suggests a slight uplift and tilting has occurred for the lithologies at the Andrew Lake site. A single lithology contact was identified at Andrew Lake dipping at 40 degrees towards the north, although with moderate confidence in the orientation. This contact could be related to a dyke or intrusion.







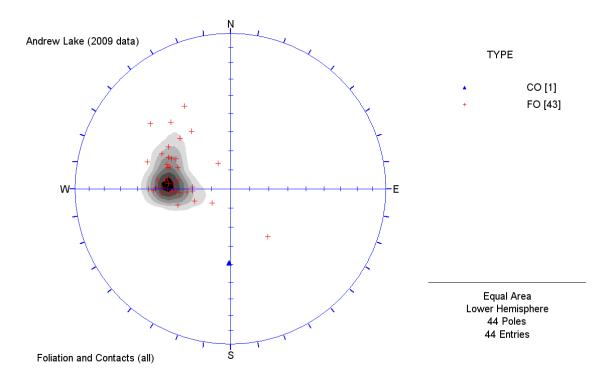


Figure C3: Andrew Lake - 2009 Oriented Core Data for Foliation (FO) and Contacts (CO)

3.2 Stereonets from Oriented Core Data

Discontinuity data was analysed statistically using the software DIPS©, distributed by Rocscience. DIPS© allows the user to analyze and visualize structural geological data using the same techniques developed for manual stereonet analysis (Rocscience 2009a). It also allows the user to contour data, analyze discontinuity statics, and select discontinuity sets.

In order to enter borehole discontinuity data (alpha and beta angles) into DIPS©, the borehole orientation had to be input as traverses. Borehole traverses were selected based on downhole surveys that were conducted by the drilling contractor as the borehole was advanced. The downhole surveys were conducted using a Reflex EZ-Shot tool, manufactured by Reflex Instruments. These surveys recorded borehole dip and dip direction at specific depth points along the borehole, with the survey interval generally every 51 m. This survey data was entered into DIPS© as separate traverses, to which the discontinuity data was assigned. This allowed for the incorporation of any borehole deviation into the analysis of the discontinuity data. As the reflex tool provided measurements relative to magnetic north, a correction for magnetic declination was applied in DIPS©. The magnetic declination was calculated based on the Geological Survey of Canada online magnetic declination calculator, and was found to be 1.4 degrees west for the Kiggavik site.

The pole to a plane is a convenient geomechanical construction that can be used to uniquely define the inclination and orientation of the plane of a discontinuity on the stereonet projection. As each pole is located 90° from its plane, a plot of poles represent the dip and dip direction of all measured discontinuities. Thus in turn, statistical contouring allows the pole density for a given discontinuity fabric to be delineated on the stereonet, which provides a 3D representation of structural data. The analysis of the structural data from Andrew Lake was



NA.

APPENDIX C - STEREONET ANALYSIS

carried out using lower hemisphere equal area stereonet projections, and a Fisher distribution. Discontinuity sets are represented on the stereonets as areas with high pole densities or concentrations as indicated by the contours.

As much of the data was from line sources (boreholes), a bias correction (referred to in DIPS© as the Terzaghi correction) was applied to the oriented core data to help eliminate the problem of data misrepresentation. The bias correction calculates a geometrical weighting factor to each discontinuity measured, with the highest correction applied to the structures that are subparallel to the borehole orientation. Discontinuities that are perpendicular to the core axis receive the smallest weighting factor, since these features are intersected more often in the borehole, and are therefore measured more frequently during the core logging. Since the weighting function tends to infinity as the angle between the discontinuity and the borehole axis (α) approaches zero, a maximum weighting corresponding to a 15° bias angle was applied to any plane with $\alpha \le 15^\circ$ orientation (Rocscience, 2009b).

3.3 Definitions of Pole Concentrations

For Andrew Lake, there was minimal validated orientation data. The method used for identifying major and minor sets was based on pole density. When treating data per drillhole, a pole density or concentration between 1% and 3% was selected to represent a minor discontinuity set, while a pole density greater than approximately 3% was selected to indicate major or dominant joint sets. When assessing all drillholes together, generally pole densities of 1% to 2% and greater than 2% were used to define minor and major sets, respectively. However, due to the low number of total data poles considered, consideration was also given to the continuity of the set between boreholes. If the set appeared in more than one borehole, and was apparent when all the data from all the boreholes were plotted together, it was considered a major set. If it was only apparent in one borehole, and not apparent when all the data was plotted together, then it was considered a minor set within the particular borehole it occurred in.

The discontinuity sets for Andrew Lake were selected manually by identifying the peak concentrations and recording the orientation of that point. This was done using a Terzaghi weighted contour plot of all the borehole data in DIPS© (including oriented core data from 2007/2008), using the "add plane" function.

The discontinuity sets were numbered and labelled with upper and lower case fonts representing, respectively, major and minor discontinuity sets. For example, 'JN' or 'jn' was used to differentiate major or minor discontinuity sets. Since these major and minor sets show some variations in terms of both dip and dip directions, for the kinematics analyses, these variations have been addressed by considering sub-sets, which were labelled with the letter 'A' (or 'a') and 'B' (or 'b').

3.4 Results from Stereonet Analysis

3.4.1 Assessment of Oriented Core Data

Figures C8 to C12 appended to this text show the stereographic projections for Andrew Lake. These figures show the oriented core data plotted by the individual borehole, as well as data symbolic pole plots of grouped data showing types of features logged (i.e., bedding, foliation, joints, shears, etc.) as well as the joint alteration index value referring to the degree of alteration of the joint surface (Golder 2009). Some additional comments and interpretation are included on these figures.





As discussed previously, there was some difficulty maintaining orientation in the drilled core, particularly in zones of highly fractured or mechanically broken rock. Assessment was made in the field as to the high, moderate, and low confidence based on alignment of consecutive orientation reference lines.

For the Andrew Lake data, there is appreciably more pole scatter, likely related to the drilling and rock conditions, and the difficulties maintaining valid core orientation. All Andrew Lake 2009 data (1239 features) was compared to the moderate to high confidence data (487 features) as shown on Figure C4 below. By removing the low confidence data, much of the apparently random pole scatter is removed from the data set, therefore, only the moderate to high confidence data was used in analysing the specific discontinuity sets.

The borehole directions are also plotted on Figure C4 for comparison. Holes AND-09-01 to -03 are drilled at azimuths ranging between 225 and 330 degrees. The 2008 borehole ANDW-08-04 (119 features) was drilled in a complimentary direction, at approximately 090 degrees. This data is also plotted on a separate stereonet on Figure C4. In order to reduce potential borehole direction bias, and optimize the amount of available features for orientation, the 2008 and 2009 (validated) data was combined for analysis of discontinuity sets at Andrew Lake.





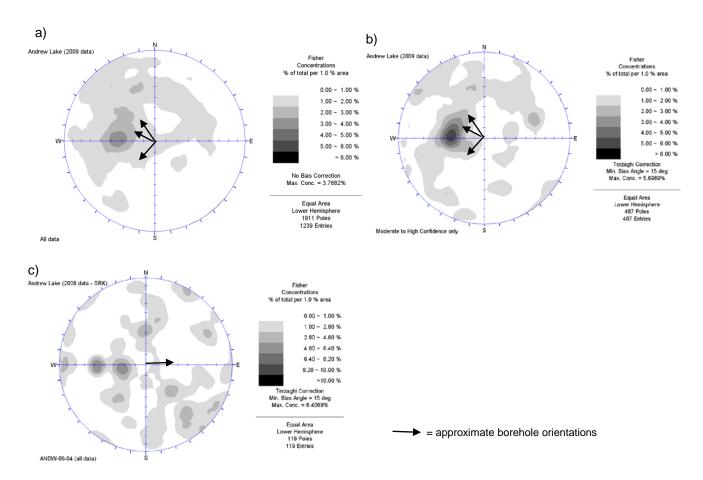


Figure C4: Andrew Lake - Oriented Core data Contoured; a) All 2009 Data; b) Moderate to High Confidence 2009 Data; and c) 2008 data (SRK)

3.4.2 Selected Discontinuity Sets

For the Andrew Lake structural data analyses, the validated oriented core data sourced from the 3 geotechnical boreholes were analysed by borehole for the identification of major and minor sets following the conventions discussed previously (major = greater than 3% pole concentration). The 2008 oriented core data from borehole ANDW-08-04 was also assessed following similar criteria. The selected major and minor discontinuity sets are summarised by trend and inclination in Table C3. Also given on this table are the summary of discontinuity sets identified by SRK (SRK, 2009) based on their assessment of the ANDW-08-04 data. Curiously, some of the discontinuity sets identified by Golder for the ANDW-08-04 were not previously identified by SRK. It is uncertain as the procedures used by SRK for selection of their reported sets, therefore some uncertainty surrounds this data.

The selected major and minor discontinuity sets by borehole are given on the stereoplots in Figure C5. There is generally considerable scatter in the selected major sets, as most of the main data trends show an association with major sets as identified in one or more boreholes. A naming convention has been developed for the set



identification based on interpretations of the main data trends associated with bedding, foliation, or inferred fault trends, as discussed later.

Overall, 8 main discontinuity trends were identified at Andrew Lake, and some of these trends were identified with possible sub-sets or localized variations to the major trends. The sets identified as FO1, JN1, JN2, and JN5 appear to be the most representative of the overall rock mass fabric at Andrew Lake. Some interpretation of these sets is given below:

- FO1A (fo1b/fo1c) sub-horizontal to inclined foliation or bedding set identified in all boreholes, generally dipping between 20 and 55 degrees to the east. Apparently associated with bedding in the metasediments, and possibly foliation/stress-relief in the non-bedded rock units. Identified features include mainly foliation with some veins and joints.
- JN1 (JN1A/JN1B) inclined set, striking east-west and dipping south between 30 and 60 degrees. Possibly associated with the east-west striking faults identified in the Andrew Lake geology model. Identified as a major set in 3 of 4 boreholes. Associated features include mainly joints, with some veins and occasional shears.
- JN2 (JN2A/JN2B) sub-vertical set, striking northwest-southeast and dipping steeply either to the northeast or southwest. Trends orthogonal to the main regional fault trends through Andrew Lake possibly associated with stress-relief. Identified at a major set in 2 of 4 boreholes. Identified features include mainly joints, with some veins and occasional shear. Considerable infilling of joints with clay noted for the sub-set JN2A trend.
- JN5 sub-vertical set, striking near parallel to the main Andrew Lake fault trend (northeast-southwest). Noted as a major set in 3 of 4 boreholes. The main identified features include joints, with some veins and occasional shears.

The delineation of sets is based on limited valid data with only 606 oriented features. Additional structural data collection is suggested for boreholes taken at various orientations to better help define the rock mass fabric for Andrew Lake.





Table C3: Summary of Discontinuity sets for the 2009 Golder, 2008 (SRK), Oriented Core Data at the Andrew Lake Site.

Trend	North-South				Northeast-Southwest			East-West			Southeast-Northwest								
Dip	Sub-Vertice to 9	al (dip 60° 90°)	Inclined (6		Sub-Vertice to 9	al (dip 60° 90°)	Inclined (60	dip 30° to)°)		cal (dip 60° 90°)	Inclined (d	lip 30° to 60°)	Sub-Vertice to	al (dip 60° 90°)	Inclined (60	dip 30° to)°)	Sub-Horizontal to Shallow Inclination (dip 0° to 30°)		
Dip Direction	Dip E	Dip W	Dip E	Dip W	Dip SE	Dip NW	Dip SE	Dip NW	Dip N	Dip S	Dip N	Dip S	Dip NE	Dip SW	Dip NE	Dip SW			
Set ID	JN6	-	fo1b/fo1c	JN4B	JI	N 5	-	-	-	-	CJN1A	JN1A/JN1B	JN2A	JN2B	JN3	-	FO1A	CJN1B	JN4A
AND-09-01	63/072 66/055	-	54/093 44/116	54/261 56/288	-	-	-	-	-	-	57/338 69/348	46/155 52/188	79/041	-	63/072 49/056	-	-	32/359 37/336	22/270 29/252 27/290
AND-09-02	-	-	-	60/245	86/141 66/132	-	-	-	-	-	-	51/177 50/205	81/012	70/215 83/234	64/073	-	29/094	27/313	-
AND-09-03	85/078	-	-	50/250	84/123 73/143	-	-	-	-	-	50/341	61/171	70/030	85/203	62/057	-	31/091 16/059	-	-
ANDW-08-04	81/090	-	46/088	68/243	87/113	82/300 63/321	-	-	-	-	62/352	32/180 53/191	-	85/209	41/040	-	22/081	33/316	-
SRK Reported (SRK, 2009)	-	-	-	-	-	75/316	-	-	-	-	-	50/194	-	70/244	-	-	16/186	37/312	-

Upper case set names indicate major sets, lower case set names indicate minor sets. Major sets highlighted in grey. Sets given by dip/dip direction: N = North, S = South, E = East, W = West, SE = southeast, NW = Northeast, NW = Nort





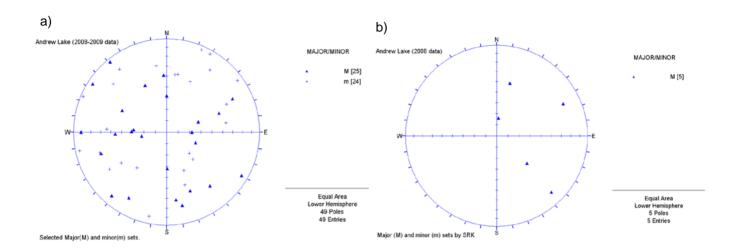


Figure C5: Andrew Lake - Selected Major and Minor Sets by Borehole; a) 2008 & 2009 Sets by Golder; and b) ANDW-08-04 Sets Reported by SRK

3.5 Structural Domains

As part of the analysis of the structural stereonet data, an assessment of structural domains that may be present at Andrew Lake was conducted. A structural domain is an area within the deposit which has different structural conditions from other areas within the deposit. This could be due to a change in rock type, hanging wall versus footwall rocks, proximity to major features, etc.

The Andrew Lake structural data by rock type are plotted in individual stereonets on Figure C6. There is generally shown to be agreement for the highest concentrations of data, representative of sets FO1, JN1, JN2, and JN5. The metasediments show more pole scatter, likely associated with the poorer drilling conditions, as well as potential variability in the rock mass structure in these weaker rock units. Sets FO1 and JN1 are shown to be particularly dominant in the granitic units. Sets FO1 and JN5 are dominant in the gneisses.

Overall it is difficult to make a solid interpretation on the various structural domains for the Andrew Lake pit, as there is a lack of data within the various rock types, as well as a lack of understanding as to the assemblage of the rock units and locations and orientations of the major fault features which might be encountered on the pit walls. For this reason, it was decided to use a single structural domain representative of all available data at Andrew Lake.





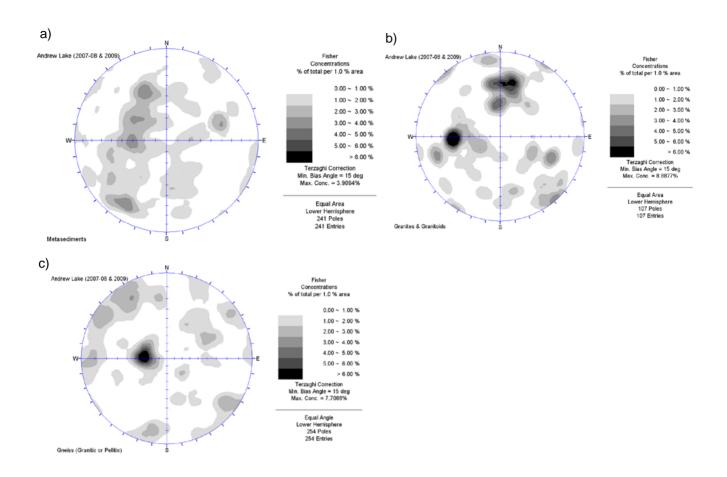


Figure C6: Andrew Lake - Combined 2008 (SRK) and 2009 Data Contoured; a) Metasediments; b) Granites and Granitoids; and c) Gneiss (Granitic or Pelitic)

4.0 DETERMINATION OF STRUCTURAL SETS FOR KINEMATIC ANALYSES

Based on the proceeding discussions related to the oriented core data obtained from the 2009 geotechnical investigation, with data supplemented from the 2008 (SRK) investigations, an assessment was made of major and minor structural sets to be used for kinematic analyses for pit slope design.

At Andrew Lake, there is a general lack of oriented core data for assessment of multiple structural domains, or for making an accurate interpretation as to the deviation of structure within the various rock units. Furthermore, the distribution of data from the oriented boreholes, as well as the data cited from previous investigations, has left some uncertainty as to the definition of major versus minor sets. Therefore all sets were assumed to have equal importance for the kinematic analyses. Additional data supplementing the 2009 investigation included regionally identified fault trends and discontinuity data identified by SRK from the 2008 oriented boreholes. It should be noted that the discontinuity information used from SRK was sourced from the geotechnical database included as Appendix E in the March 2009 version of the geotechnical data report for 20007/2008 (SRK, 2009). Several different sets based on this data were identified in a draft technical memorandum by SRK dated





November 23, 2008. Both the reported and raw data sourced from these documents were considered in supplementing the 2009 data. The selected discontinuity sets for Andrew Lake are presented in Table C4. The selected sets are plotted with the 2008 and 2009 oriented core data (contoured) on Figure C7 for comparison.

5.0 REFERENCES

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Table C4: Structural Sets for Kinematic Analysis from Andrew Lake (all sets assumed major for kinematic analysis).

Major F	S	tructu	ral	Time		
Strike/Trend	Dip	Set	Dip	Dip Dir	Туре	
N-S	E (sub-horizontal)	FO1A	29	93	Major Foliation Set - appears as the most dominant discontinuity orientation within the Andrew Lake pit area associated with foliation/bedding in the metasediment units but also to a considerable extent in the granitoids. Data trends sub-horizontal to inclined.	
E-W	S (inclined)	JN1A			Major Joint Set - perpendicular to the foliation and may related to a series of E-W striking faults which	
	S (inclined)	JN1B	45	200	cross cut the regional fault trend in the Andrew Lake area. Shown to be particularly dominant in the granites/granitoids.	
NW-SE	NE (sub-vertical)	JN2A	69	31	Major Joint Set - this set is trending orthogonal the to the regional Andrew Lake Fault orientation. Identified	
	SW (sub-vertical)	JN2B	73	214	with varying concentrations in all boreholes.	
NW-SE	NE (sub-vertical)	JN3	60	65	Major Joint Set - identified in all four geotechnical boreholes with varying degrees of persistence. Particularly dominant in the upper metasediment units possibly associated with variation on the main FO1 bedding trend.	
N-S	W (sub-horizontal)	JN4A	25	257	Major Joint Set - two possible variations to this trend	
	W (inclined)	JN4B	52	247	are sub-horizontal or inclined. It appears almost oblique to the major foliation set.	
NE-SW	SE (sub-vertical)	JN5	84	135	Major Joint Set – strikes near parallel to the main NE- SW striking regional fault trend at Andrew Lake. Identified as a major set in 3 of 4 boreholes.	
E-W	N (inclined)	CJN1A	51	344	Major Joint Set - this joint set is conjugate to JN1 and is inclined to sub-horizontal. It strikes approximately	
NE-SW	NW (sub-horizontal)	CJN1B	30	317	parallel to the E-W fault trend in the Andrew Lake area.	
N-S	E (sub-vertical)	JN6	82	90	Major Joint Set - this set strikes approximately parallel to the regional fault system in the Andrew Lake area if considering a N-S faulting trend. Identified in 3 of 4 boreholes.	

Dip Dir = dip direction; Structural dip and dip direction shown in degrees; JN = joint, FO = foliation, CJN = conjugate joint





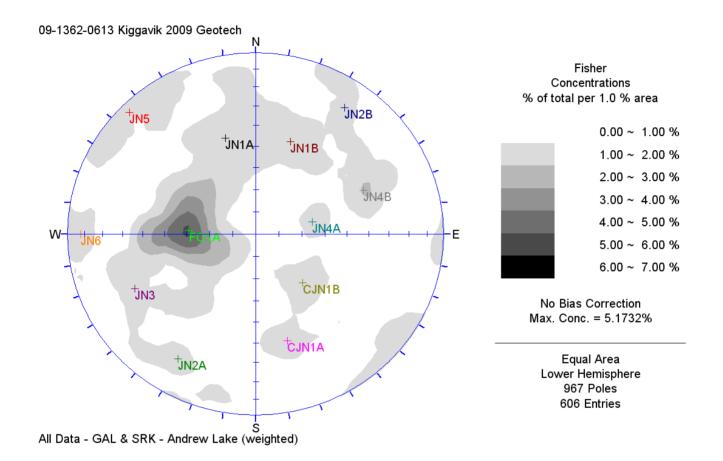


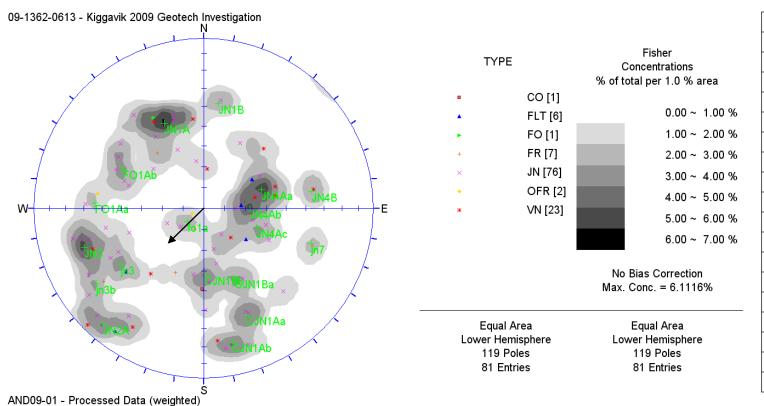
Figure C7: Andrew Lake - 2008 and 2009 Oriented Core Data (contoured) with Selected Structural Sets



Rev : 13 Nov 09

Kiggavik – Andrew Lake Identified Sets By Each Borehole – AND09-01

Figure C8

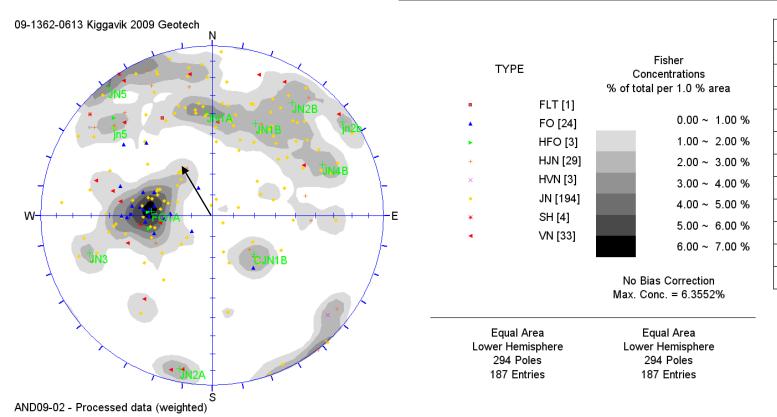


Set	Dip	Dip Dir
FO1Aa	54	93
FO1Ab	44	116
JN1A	46	155
JN1B	52	188
JN2A	79	41
JN3	63	72
JN4Aa	29	252
JN4Ab	22	270
JN4Ac	27	290
JN4B	54	261
CJN1Aa	57	338
CJN1Ab	69	348
CJN1Ba	37	336
CJN1Bb	32	359
fo1a	10	51
jn3	49	56
jn3b	66	55
jn7	56	288

- Shown are the processed data files with a Terzaghi weighting applied. The number of entries indicate the total number of data points considered, which are shown as poles in the stereonet. The number of poles shown is the Terzaghi weighted value that the Fisher concentration contours are based on.
- •Upper case indicates major sets, lower case indicates minor sets
- FO = foliation, JN = joint set, CJN = conjugate joint set, Dip Dir = Dip Direction
- Orientations (Dip and Dip Direction) are based on selected peak pole concentrations, based on the Fisher Concentration contours
- Major sets were selected based on its continuity between boreholes, and its orientation with respect to regional structure
- Major sets with a lower case letter following the name (i.e., FO1Aa) indicate different peaks which were interpreted to occur within the same set due to inherent scatter within the data.
- the arrow indicates the borehole orientation

Kiggavik – Andrew Lake Identified Sets By Each Borehole – AND09-02

Figure C9

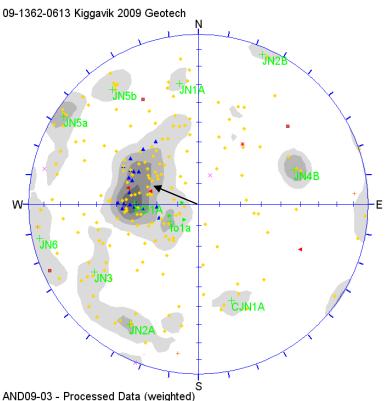


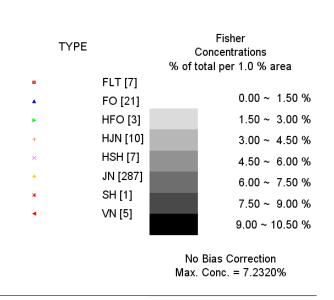
Set	Dip	Dip Dir
FO1A	29	94
JN1A	51	177
JN1B	50	205
JN2A	81	12
JN2B	70	215
JN3	64	73
JN4B	60	245
JN5	86	141
CJN1B	27	313
jn2b	83	234
jn5	66	132
-		

- Shown are the processed data files with a Terzaghi weighting applied. The number of entries indicate the total number of data points considered, which are shown as poles in the stereonet. The number of poles shown is the Terzaghi weighted value that the Fisher concentration contours are based on.
- •Upper case indicates major sets, lower case indicates minor sets
- FO = foliation, JN = joint set, CJN = conjugate joint set, Dip Dir = Dip Direction
- Orientations (Dip and Dip Direction) are based on selected peak pole concentrations, based on the Fisher Concentration contours
- Major sets were selected based on its continuity between boreholes, and its orientation with respect to regional structure
- Major sets with a lower case letter following the name (i.e., FO1Aa) indicate different peaks which were interpreted to occur within the same set due to inherent scatter within the data.
- the arrow indicates the borehole orientation

Kiggavik - Andrew Lake Identified Sets By Each Borehole - AND09-03

Figure C10





Set	Dip	Dip Dir
FO1a	31	91
JN1A	61	171
JN2A	70	30
JN2B	85	203
JN3	62	57
JN4B	50	250
JN5a	84	123
JN5b	73	143
CJN1A	50	341
JN6	85	78
fo1a	16	59

Lower Hemisphere 344 Poles 219 Entries

Equal Area Lower Hemisphere 344 Poles 219 Entries

• Shown are the processed data files with a Terzaghi weighting applied. The number of entries indicate the total number of data points considered, which are shown as poles in the stereonet. The number of poles shown is the Terzaghi weighted value that the Fisher concentration contours are based on.

Equal Area

- •Upper case indicates major sets, lower case indicates minor sets
- FO = foliation, JN = joint set, CJN = conjugate joint set, Dip Dir = Dip Direction
- Orientations (Dip and Dip Direction) are based on selected peak pole concentrations, based on the Fisher Concentration contours
- Major sets were selected based on its continuity between boreholes, and its orientation with respect to regional structure
- Major sets with a lower case letter following the name (i.e., FO1Aa) indicate different peaks which were interpreted to occur within the same set due to inherent scatter within the data.
- the arrow indicates the borehole orientation

Drawn: FAM

ANDW-08-04 - Processed Data (weighted)

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Kiggavik - Andrew Lake Identified Sets By Each Borehole - ANDW-08-04 (as logged by SRK Consulting in 2008)

Figure C11

22

46

32

53

85

41

68

87

82

63

33

62

81

Dip Dir

81

88

180

191

209

40

243

113

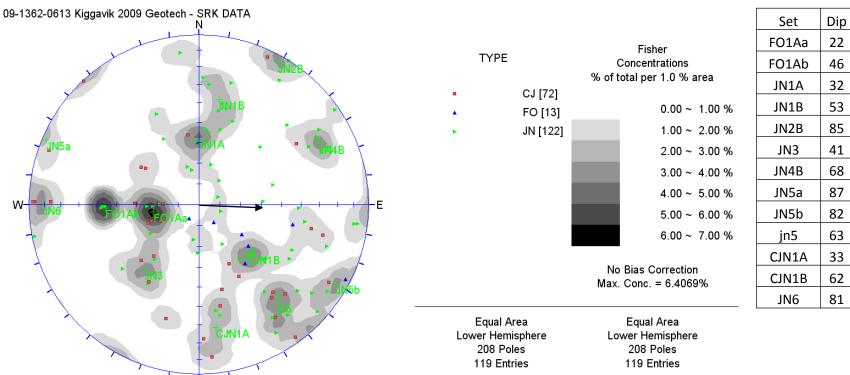
300

321

316

352

90

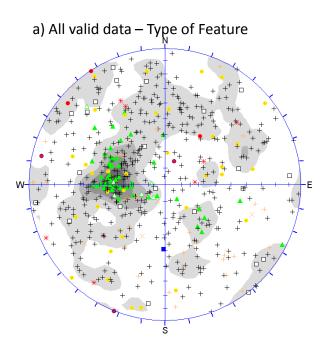


- Shown are the processed data files with a Terzaghi weighting applied. The number of entries indicate the total number of data points considered, which are shown as poles in the stereonet. The number of poles shown is the Terzaghi weighted value that the Fisher concentration contours are based on.
- Upper case indicates major sets, lower case indicates minor sets
- FO = foliation, JN = joint set, CJN = conjugate joint set, Dip Dir = Dip Direction
- Orientations (Dip and Dip Direction) are based on selected peak pole concentrations, based on the Fisher Concentration contours
- Major sets were selected based on its continuity between boreholes, and its orientation with respect to regional structure
- Major sets with a lower case letter following the name (i.e., FO1Aa) indicate different peaks which were interpreted to occur within the same set due to inherent scatter within the data.
- the arrow indicates the borehole orientation

Rev.: 13 Nov 09

Kiggavik – Andrew Lake All Validated Data with Symbolic Pole Plots Feature Type & Discontinuity Alteration Index

Figure C12

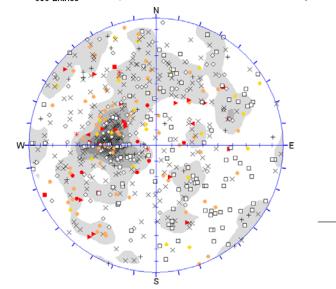


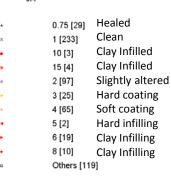
TYPE

- ^{CJ [36]} Cross-joint
- CO [1] Contact
- FLT [9] Fault
- FO [53] Foliation
- FR [5] Fracture
 - HFO [6] Healed Foliation
- HSH [3] Healed Shear
- JN [434] Joint
- SH [3] Shear
- VN [34] Vein
- Others [22]

Equal Area Lower Hemisphere 606 Poles 606 Entries

b) All valid data - Discontinuity Alteration Index (Ja)





Equal Area Lower Hemisphere 606 Poles 606 Entries



APPENDIX D

Rock Mass Stability Analysis





APPENDIX D - ROCK MASS STABILITY ANALYSIS

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APPENDIX D - ROCK MASS STABILITY ANALYSIS

1.0 INTRODUCTION

This appendix presents the results of the rock mass stability assessments carried out for the Andrew Lake pit. The stability assessments included limit equilibrium slope stability analyses, as well as floor heave stability analyses. The slope stability analyses were conducted for generalized two-dimensional rock mass slope configurations developed from the information and interpretations presented and developed in the previous appendices. Various rock mass strengths and qualities were assessed in order to estimate pit slope angles which would achieve a reasonable and representative factor of safety against deep-seated rock mass failure.

The floor heave stability analyses were developed to assess the potential for floor heave due to the high artesian water pressures acting at depths below the permafrost line within the pit floors. The floor heave stability analysis results are intended to provide a range of conservative thicknesses of pit floor above the permafrost zone at which time remedial measures such as vertical drains could be installed to depressurize the pit floor.

The following appendix presents the methodology and results of the rock mass stability assessments.

2.0 SLOPE STABILITY ANALYSES

2.1 SLOPE MODEL CONFIGURATIONS

2.1.1 Generalized Rock Mass Slope Configurations

The limit equilibrium slope stability analysis software Slide® v.5 (Rocscience Inc.) was used for assessing the stability of the design slopes against deep seated rock mass failure. Parametric trials assessing the factor of safety versus slope inclination were conducted for the various slope geometries and material configurations.

The slope configurations and engineering geology conditions used for slope stability analyses for Andrew Lake are presented in Figure D1. The material properties for the various slope configurations were inferred from the results of the rock strength (Appendix A) and rock mass classification work (Appendix B).

