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DRILLING

ASSESSMENT REPORT

ON THE

KLIYUL PROPERTY

LATITUDE: 56° 30'N LONGITUDE: 126° 08'W

JUNE 1993

GEOLOGICAL BRANCH ASSESSMENT REPORT

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Author: D.G. Gill (Project Geologist)

Operator: Noranda Exploration Company, Limited (No Personal Liability)

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1.0 INTRODUCTION

During the period between February 1 and March 9, 1993, Noranda Exploration Company, Ltd. and Midnight Sun Drilling Company, Ltd. of Whitehorse, Y.T. conducted a 6 hole, 560 meter reserve circulation drill programme on the Kliyul Property to test both magnetic highs and coincident copper-gold soil anomalies associated with a Cu-Fe-Au skarn zone which has had limited drill testing by both Sumac Mines Ltd. (1974) and Kennco/Vital Pacific Resources Ltd. (1981).

Prior to the actual fieldwork, computer integrated maps using all historic geochemical (soils) and geophysical (mag, I.P., resistivity) survey data were generated by Noranda. The integrated magnetic data revealed that the historic drilling was concentrated on the northern flank of one of several magnetic highs which constitute a larger zone of high magnetic susceptibility measuring 450 m x 350 m. Examination of the old drill core and drill logs confirms that the higher grade copper and gold values are associated with the higher magnetite content. Based on this information the 1993 drill programme concentrated on the area surrounding the main skarn zone to delineate the extent of the known mineralization, on areas of high magnetic susceptibility within the 450 m x 350 m magnetic anomaly which were previously untested and upon untested coincident copper-gold soil anomalies occurring immediately to the east of the large magnetic high.

1.1 Location and Access

The Kliyul property is located approximately 200 kms northnortheast of Smithers, B.C. on NTS Mapsheets 94D/8 and 9 in the Omineca Mining Division.

Camp mobilization was achieved by both helicopter based at the Osilinka Logging Camp and along an old cat trail which follows the Asitka River valley to the west of the property and connects with the Omineca Mining Road approximately 8 kms west of Johanson Lake (refer to Drawing 1 for a rough location of the property).

1.2 <u>Topography and Physiography</u>

The Kliyul property is situated above treeline with elevations ranging from 5600 to 7000 feet. The claims straddle an eastnortheast trending glacial valley which is drained to the east and northeast by Lay Creek and to the southeast and southwest by tributaries at the headwaters of Kliyul Creek.



Slopes to a maximum of 45° occur in the northern portion of the property along an east-west trending ridge whereas the southwestern part of the claim group covers a gently sloping, wide, marshy valley floor. The southeastern area of the property is dominated by two northwesterly trending ridges with moderate to steep relief.

1.3 <u>History</u>

Below is a brief outline of documented work performed on the property in chronological order.

<u>1970-1972:</u> Original property staked and geochemically and geophysically surveyed by Kennco Explorations. These surveys delineated a 2.5 km x 1.0 km I.P. chargeability anomaly and coincident (yet smaller) copper soil geochemical and magnetic anomalies.

<u>1973:</u> Property optioned to Sumac Mines Ltd. who drilled 3 x-ray holes (no results available).

<u>1974:</u> Sumac Mines drilled 6 'BQ' holes to test the West and East Zone copper soil anomalies and 5 'BQ' holes into the magnetic high. The latter drill holes intersected magnetite-copper-gold mineralization within a well fractured, sericite, chlorite, epidote, carbonate, quartz, pyrite skarn hosted by calcareous andesite tuffs and agglomerates and lesser dioritic units. A reserve of 2.5 million tons of 0.3% Cu and 0.03 opt Au was returned from this skarn zone.

<u>1981:</u> Kennco and Vital Pacific drilled 4 more holes into the central skarn zone all in a southerly direction. Results on this programme were less than favourable.

<u>1984:</u> BP Minerals relogged and sampled portions of available core and conducted geological mapping and geochemical sampling.

<u>1990:</u> Placer Dome conducted linecutting, magnetometer and VLF-EM surveying, soil and rock sampling and prospecting in order to delineate magnetic anomalies similar to the known skarn zone, possible porphyry style mineralization and/or mineralized structures parallel to the large glacial valley.

<u>1992:</u> Noranda conducted 1:5,000 geological mapping concentrating on alteration assemblages as well as rock and minor soil sampling.



1.4 <u>Claims</u>

The Kliyul property is comprised of 40 two-post mineral claims (Drawing 2) owned by Kennco Explorations (Western) Ltd. (50%) and by Vital Pacific Resources Ltd. (50%).

Claim		Record		Anniversary	
Name		Number	Units	Date	
	1	245065	1	Aug 10 1970	λυα 10 1997
KT.T	2	245065	1	Aug. 10, 1970 Aug. 10, 1970	Aug. 10, 1997
KLL KLL	2	245067	1	Aug. $10, 1970$	Aug. 10, 1997
KLT	4	245068	1	Aug. 10, 1970	Aug. 10, 1997
KLT	5	245069	1	Aug. 10, 1970	Aug. 10, 1997
KT.T	6	245005	1	Aug. 10, 1970	Aug. 10, 1997
KI.T	7	245070	1	Aug. 10, 1970	Aug. 10, 1997
KLT	, 8	245071	1	Aug. 10, 1970	Aug. 10, 1997
KLT	ä	245072	1	Aug. 10, 1970	λ_{10} 10, 1997
KT.T	10	245075	1	Aug. 10, 1970	Aug. 10, 1997
KT.T	11	245074	1	Aug. 10, 1970	Aug. 10, 1997
KLT	12	245075	1	λυσ 10 1970	Aug. 10, 1997
KT.T	13	245070	1	Aug. 10, 1970	Aug. 10, 1997
KL.T	14	245077	1	Aug. 10, 1970	Aug. 10, 1997
KT.T	15	245070	1	Aug. 10, 1970	Aug. 10. 1997
KLT	16	245075	1	Aug. 10, 1970	λ_{10} 10, 1997
KT.T	17	245081	1	Aug. 10, 1970	Aug. 10, 1997
KLT	18	245001	1	Aug. 10, 1970	Aug. 10, 1997
KLT	19	245083	1	Aug. 10, 1970	Aug. 10, 1997
KLT	20	245085	1	Aug. 10, 1970	Aug. 10, 1997
KLT	21	245054	1	Sen 11 1970	Sen 11 1997
KLT	25	245155	1	Sep. 11, 1970	Sen 11 1997
KT.T	26	245150	1	Sep. 11, 1970	Sep. 11, 1997
KLT	27	245158	1	Sep. 11 1970	Sep. 11, 1997
KLT	28	245150	1	Sep. 11 1970	Sen 11 1997
KLT	39	245682	1	Tul 12 1971	300, 12, 1997
KT.T	40	245002	1	[Ju] 12, 1271	Tu1 12 1997
KT.T	40	245384	1	T_{11} 12, 1971	J_{11} $12, 1997$
KLT	42	245385	1	Jul 12, 1971	Jul 12, 1997
KLI	43	245386	1	Jul. 12, 1971	Jul 12, 1997
KT.T	44	245387	1	Tul 12 1971	J_{11} J_{2} J_{997}
KLT	45	245507	1	Tu 12, 1971	Tu = 12, 1997
KLT	45	245500	1	[Ju] 12, 1971	T_{11} 12, 1997
KLT	47	245505	1	Tu = 12, 1971	T_{11} 12, 1997
KLT	18	245550	1	Tu = 12, 1971	T_{11} 12, 1997
KI'L	40	240001	⊥ 1	.Tul 19 1071	Tul 12 1007
KT.T	79 50	240002	1	Jul 12, 13/1	Jul. 12, 1337
КПŤ	50	240000	T	JUI. 16, 19/1	JUL. 12, 1997
UTA	4	245777	1	Aug. 29, 1973	Aug. 29, 1997
UTA	6	245778	1	Aug. 29, 1973	Aug. 29, 1997
UTA	8	245779	1	Aug. 29, 1973	Aug. 29, 1997

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Although work done described in this report was conducted on the Kli 13 & 15 claims several different groupings of surrounding claims have been made in order to fulfil assessment requirements for each claim within each group. In addition several two-post claims were staked in order to keep all claims within such groupings contiguous.

Please refer to the Statement of Exploration forms at the beginning of this report for further clarification of assessment in each group. Following are a series of maps showing the claim groupings involved.

1.5 <u>Economic Potential</u>

Historical drilling on the Kliyul property by Sumac Mines Ltd. and Kennco/Vital Pacific in 1974 and 1981 has outlined a Cu-Fe-Au skarn zone estimated to contain 2.5 million tons of 0.3% Cu and 1.03 gpt Au which is situated on the northern flank of one of several magnetic highs which constitute a larger zone of high magnetic susceptibility measuring 450 m x 350 m. Previous examination of the old drill core and drill logs suggests that the higher grade copper and gold values are associated with higher magnetite content. Based on this information the potential for increasing the tonnage of the Cu-Fe-Au skarn zone is considered excellent.

1.6 <u>Survey Control</u>

The surveying of drill hole collars during this programme was conducted utilizing Placer Dome's 1990 metrically chained and slope corrected grid for control. This grid was also used by Noranda field personnel during the 1992 mapping programme.

1.7 <u>Sampling</u>

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Sampling of the reverse circulation chips was done at 2.0 m intervals. All samples were sent to Noranda Exploration laboratory at Unit #1, 7550-76th Street, Delta, B.C. A total of 254 samples were analyzed by ICP (30 element) and geochem-ed for gold. Refer to Appendix II for a more detailed description of rotary drilling sampling techniques and Appendix I for descriptions on laboratory analytical techniques.













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2.0 <u>GEOLOGY</u> (See Drawing 13)

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The Kliyul property is situated within the Intermontane Belt which is comprised of Upper Triassic to Lower Jurassic island arc volcanics, volcaniclastics and minor sediments of the Takla Group which hosts such Cu-Au porphyry deposits as Mt. Milligan and Kemess. The dominantly volcanic package in the Kliyul Creek area has been intruded by Jura-Cretaceous aged diorites, monzonites and syenites associated with the Hogem Batholith.

The 1992 programme completed by Noranda included the confirmation of geological mapping completed by BP Minerals in 1984 (Assessment Report #13258) as well as mapping alteration assemblages.

The mapping programme confirmed the fact that the majority of the Kliyul property is underlain by mainly propylitized andesitic tuffs and flows of (Unit 1) the Takla Group which also includes a section of calcareous andesite tuffs and agglomerates (Unit 8) containing large chunks of limestone up to 30 cms in length. Minor beds of black pyritic shales and limestones (Units 7 & 6) are also found within the major volcanic pile. Field observations reveal that these units strike in a north-northwest direction and dip mainly to the west and locally eastward. Intrusive activity found on the property is confined to small plugs and dykes of diorite to gabbro composition (Unit 5) and a large listwanite dyke (Unit 10) which is observed in both the southeast and northwest corners of the claim block.

Several large gossanous areas occur throughout the claim block and can be attributed to pyrite and silica (Unit 9), sericite (Unit 4) and a combination of quartz-sericite-clay-pyrite (Unit 3) altered zones ranging in intensity from weak to intense. All gossanous zones appear to be related to large structural breaks which are delineated by the presence of deeply incised gulleys, alignment of creeks and lakes, patches of ferrocrete (Unit 2) and large dykes, i.e. listwanite. These structural breaks appear to generally trend east-west, north-south and east-southeast-westnorthwest.

The largest of these gossans is the quartz-sericite-kaolinitepyrite (up to 15%) zone located in the southeast corner of the claim block which extends to the southeast onto ground held by Golden Rule Resources (BAP claims). This area is underlain by foliated andesitic tuffs and a large number of radiating diorite dykes and precious to base metal rich quartz veins. This gossan is exposed for at least 200 metres vertically, 300-400 metres in width and over a kilometre in length. Although very minor copper mineralization (malachite, chalcopyrite) was observed in this area the widespread alteration and occurrence of radiating dykes and quartz veins may suggest that this exposed section is the surface manifestation of a buried intrusive body which may contain porphyry



or epithermal type mineralization. To the southeast this large gossan is covered by talus slopes of the East Kliyul Creek canyon and is offset to the south by a small northeast trending fault. Northward the zone appears to become more sericite altered before intersecting a large east-west trending fault (paralleling Lay Creek and marked by the presence of ferrocrete outcrop) which offsets the zone approximately 3.2 kms to the west where it is exposed again in the northwest corner of the property. (This offset is also confirmed by the similar westward movement of the calcareous andesitic tuff and agglomerate unit and possibly the large listwanite dyke).

Other zones of weak to intense pyrite-silica alteration were also observed to occur along north-south trending fracture sets and overlying & adjacent to, east-west trending fault zones such as those roughly paralleling Lay Creek in the centre of the claim block and along the base of the slope south of 'Andesite Ridge' in the north-central portion of the property.

Field observations of the Kli skarn zone were limited to only one outcrop located on Line 2800E, 2055N due to the large amount of drift cover in the centre of the large east-west glacial valley although the calcareous andesitic tuff and agglomerate unit is seen to lie directly on strike 500 meters to the south.

3.0 PRESENTATION OF DRILL HOLE DATA

Drilling parameters of Noranda's 1993 programme are listed in the table below. Refer to Drawings 14-21 for plans and sections of all holes. Sections are in bar graph form showing copper and gold results for each sample interval (2.0 meters).

HOLE #	TOTAL LENGTH (meters)	Coord East	INATES NORTH	AZIMUTH (True)	DIP	DATE COLLARED	DATE Complete:	
RC-KL-93-1	88.0	2805	 1730	. 	-50°	Feb 20/93	Feb 21/93	
RC-KL-93-2	112.0	3150	1900	0°	-50°	Feb 22/93	Feb 23/93	
RC-KL-93-3	60.0	3132	1992	033°	-50°	Feb 23/93	Feb 24/93	
RC-KL-93-4	120.0	2790	1937	050°	-50°	Feb 24/93	Feb 25/93	
RC-KL-93-5	100.0	2654	1931	060°	-50°	Feb 25/93	Feb 26/93	
RC-KL-93-6	80.0	2715	1880	060°	- 50°	Feb 26/93	Feb 27/93	

3.1 Synopsis of Drill Holes

RC-KL-93-1

This hole was drilled to test the southern most lobe of high magnetic susceptibility with associated elevated copper, gold geochemistry. The hole encountered silica-chlorite, silica & calcsilicate altered andesites containing fine grained, disseminated and fracture filled pyrite and magnetite and very little copper mineralization. Between 84-88 m problems were encountered with a large fault zone which also appears to mark the beginning of a magnetite-silica skarn zone. This is reflected in the increase in gold values seen between 82 and 88 m at the end of hole RC-KL-93-1.

RC-KL-93-2

RC-KL-93-2 was drilled to test a zone of coincident anomalous copper-gold soil geochemistry outside of the main zone of high magnetics. As in RC-KL-93-1 this hole drilled through a series of silica, silica-chlorite and calc-silicate altered andesites with varying amounts of pyrite (1-4%), magnetite (trace to 4\%) and chalcopyrite (trace to 1\%). No significant skarn mineralization or alteration was intersected. However, the zone between 6 & 34 m returned 0.27 gpt Au, 0.28% Cu from siliceously altered andesites containing a moderate amount of quartz-calcite stringers as did the interval between 62 m and 112 m (0.47 gpt Au, 0.16% Cu) which also revealed an increase in quartz-calcite stringers and a rough increase in magnetite. The results obtained and alteration observed in this hole suggest that this area, although elevated in copper-gold, is peripheral to the main area of mineralization 300 metres to the west.

<u>RC-KL-93-3</u>

This hole was drilled to test the same coincident copper-gold soil geochemistry anomaly as RC-KL-93-2 but approximately 100 metres to the north where siliceous, carbonate and chlorite altered andesites were cored. No significant skarn mineralization was encountered in this hole although elevated copper-gold results were returned from areas containing slightly more disseminated and veined magnetite as well as increased quartz-calcite veining. As in RC-KL-93-2 the results from this hole suggest this area is peripheral to the main mineralized zone and that the surface geochem signature may be a result of the combination of eastward transport and/or an increase in quartz-calcite veining in this vicinity.

<u>RC-KL-93-4</u>

RC-KL-93-4 was drilled to test the eastern extension of the Kliyul skarn zone as defined by KL-74-5, 6 and 13. The hole encountered an intercalated series of calc-silicate, chlorite-carbonate and silica, chlorite, magnetite altered andesite package which revealed high copper and gold values associated with an increase of disseminated magnetite, pervasive silica alteration and an increase in quartz-calcite veining as seen between 30-120 m which returned 0.75 gpt Au, 0.29% Cu including 20 meters (68-88 m) of 2.10 gpt Au and 0.51% Cu.

<u>RC-KL-93-5</u>

This hole was collared approximately 50 m southwest of KL-74-07 to test the western extension of the Kliyul skarn zone underlain by high magnetics. Calc-silicate altered andesites, limestone breccia (agglomerate) and magnetite (silica) skarn zones were the rock types most frequently encountered. Several small quartzsericite altered (& sheared?) zones were also intersected which reflects the amount of faulting present in this vicinity. The best results in this hole were from 12-74 m (1.28 gpt Au, 0.26% Cu) including an interval between 50-74 m containing 2.4 gpt Au, 0.24% Cu where calc-silicate altered andesites and a magnetite-silica skarn zone were encountered. Higher grade gold values appear to be associated with an increase in magnetite as is the case in all other mineralized holes.

<u>RC-KL-93-6</u>

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RC-KL-93-6 was collared approximately 90 metres southeast of RC-KL-93-5 in order to test for the southern extension of mineralization encountered previously in holes 74-7 and 8 and RC-KL-93-5 assuming that the mineralization was striking approximately 157° and dipping steeply to the west as revealed by a 3 point structural interpretation. However, no significant mineralization was intercepted at the projected depth.

The majority of this hole intercepted calc-silicate and chlorite-carbonate altered andesites which contained much less magnetite than other well mineralized holes in this area. Only the interval between 52 and 80 metres revealed an increase in magnetite to 4-5% where 0.21% Cu and 0.21 gpt Au were returned.

4.0 CONCLUSIONS

The 1993 reverse circulation drilling programme on the Kliyul property has extended the known skarn mineralization to the west (RC-KL-93-5), southeast (RC-KL-93-4) and to depth in the south (RC-KL-93-6 and perhaps RC-KL-93-1) and has returned enough encouraging copper and gold results to warrant further drilling.

A review of a section at 060° azimuth incorporating holes KL-93-5, KL-74-7, KL-74-5, KL-81-16 and KL-81-17 and projecting KL-74-6 and 13 onto this section suggests that the mineralized zone may strike roughly east-west. It should be mentioned that the zone does not appear in holes KL-81-16 or 17 but this may be due in part to resampling of the core by BP Minerals in 1984 and lack of assay data from assessment files. If this same zone is represented by the intersections returned in RC-KL-93-4 and the bottoms of RC-KL-93-6 (& perhaps RC-KL-93-1) this implies a moderate dip of the zone to the south.

The mineralized zone encountered at depth in hole KL-74-08 may in fact represent a second, parallel mineralized zone not intersected in any of the other holes.

If this scenario is correct than it would support the fact that hole KL-81-18 would have drilled under the first zone and hole KL-81-19 would have undercut the second zone intersected in KL-74-8.

A drilling programme involving holes collared between RC-KL-93-5 and 6 at 2675E, 1900N, another to the southwest of RC-KL-93-4 at 2850E, 1930N and another near the collar of KL-81-18 and 2740E, 2000N, all trending northward would test this theory sufficiently and add tonnage to the existing mineralization if proven correct.

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APPENDIX I

LABORATORY ANALYTICAL TECHNIQUES

ANALYTICAL METHOD DESCRIPTIONS FOR GEOCHEMICAL ASSESSMENT REPORTS

The methods listed are presently applied to analyse geological materials by the Noranda Geochemical Laboratory at Vancouver.

Preparation of Samples:

Sediments and soils are dried at approximately 80°C and sieved with a 80 mesh nylon screen. The -80 mesh (0.18 mm) fraction is used for geochemical analysis.

Rock specimens are pulverized to -120 mesh (0.13 mm). Heavy mineral fractions (panned samples * from constant volume), are analysed in its entirety, when it is to be determined for gold without further sample preparation.

Analysis of Samples:

Decomposition of a 0.200 g sample is done with concentrated perchloric and nitric acid (3:1), digested for 5 hours at reflux temperature. Pulps of rock or core are weighed out at 0.4 g and chemical quantities are doubled relative to the above noted method for digestion.

The concentrations of Ag, Cd, Co, Cu, Fe, Mn, Mo, Ni, Pb, V and Zn can be determined directly from the digest (dissolution) with a conventional atomic absorption spectrometric procedure. A Varian-Techtron, Model AA-5 or Model AA-475 is used to measure elemental concentrations.

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Elements Requiring Specific Decomposition Method:

Antimony – Sb: 0.2 g sample is attacked with 3.3 ml of 6% tartaric acid, 1.5 ml conc. hydrochloric acid and 0.5 ml of conc. nitric acid, then heated in a water bath for 3 hours at 95° C. Sb is determined directly from the dissolution with an AA-475 equipped with electrodeless discharge lamp (EDL).

Arsenic - As: 0.2 - 0.3 g sample is digested with 1.5 ml of perchloric 70% and 0.5 ml of conc. nitric acid. A Varian AA-475 equipped with an As-EDL is used to measure arsenic content in the digest.

Barium - Ba: 0.1 g sample digested overnight with conc. perchloric, nitric and hydrofluoric acid; Potassium chloride added to prevent ionization. Atomic absorption using a nitrous oxide-acetylene flame determines Ba from the aqueous solution.

Bismuth - Bi: 0.2 - 0.3 g is digested with 2.0 ml of perchloric 70% and 1.0 ml of conc. nitric acid. Bismuth is determined directly from the digest with an AA-475 complete with EDL.

Gold - Au: 10.0 g sample is digested with aqua regia (1 part nitric and 3 parts hydrochloric acid). Gold is extracted with MIBK from the aqueous solution. AA is used to determine Au.

Magnesium - Mg: 0.05 - 0.10 g sample is digested with 4 ml perchloric/nitric acid (3:1). An aliquot is taken to reduce the concentration to within the range of atomic absorption. The AA-475 with the use of a nitrous oxide flame determines Mg from the aqueous solution.

Tungsten - W: 1.0 g sample sintered with a carbonate flux and thereafter leached with water. The leachate is treated with potassium thiocyanate. The yellow tungsten thiocyanate is extracted into tri-n-butyl phosphate. This permits colourimetric comparison with standards to measure tungsten concentration.

Uranium - U: An aliquot from a perchloric-nitric decomposition, usually from the multi-element digestion, is buffered. The aqueous solution is exposed to laser light, and the luminescence of the uranyl ion is quantitatively measured on the UA-3 (Scintrex).

N.B.: If additional elemental determinations are required on panned samples, state this at the time of sample submission. Requests after gold determinations would be futile.

LOWEST VALUES REPORTED IN PPM:

Ag - 0.2	Mn – 20	Zn – 1	Au - 0.01
Cd - 0.2	Mo – 1	Sb - 1	W - 2
Co - 1	Ni - 1	As - 1	U - 0.1
Cu - 1	Pb - 1	Ba - 10	_
Fe - 100	V - 10	Bi - 1	7

APPENDIX II

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ROTARY DRILLING SAMPLING TECHNIQUES

NORANDA EXPLORATION ROTARY DRILLING SAMPLING TECHNIQUES BLACK PINE PROPERTY, IDAHO

Drill in use is a down hole hammer, rotary drill. Some 60% of the material from the hole is lost to "blow-by" at the bottom of the hole. The remaining 40% is taken up the drill pipe.



The sample comes out of the drill pipe into the cyclone where the fines are "condensed" along with the chips. Some estimate of the amount of fines lost can be made by the amount of material coming out of the top of the cyclone. Within the cyclone, the sample ends up stratified, with the top of the hole in the bottom of the cyclone.



If necessary the extremely fine dust can be also collected, though it takes a great amount of effort. It may be justified if the drilling is for economic evaluation purposes and there is evidence that the gold content is higher in the fines.

The sample is first split below the cyclone, where the retained volume is adjusted to get the necessary sample size (about 30lbs). The samples are taken every 5ft, which is estimated by the driller watching the rods go down.

The sample is then split a second time into two 15lb samples, one of which is left on site and the other is sent for analysis. The sample left on site is probably useless after a few months due to deterioration of the bag and loss of the writing on the bag.



Two scoops are taken out of the reject bag. One is put as is into a chip tray and is labelled as an unwashed sample. The second is put into a kitchen sieve and dipped into a pail of water to produce a washed chip sample. This is retained in a second chip tray.

The unwashed sample gives information about colour of sample (which reflects oxidation, carbon content, quartz or calcite content, etc.), clay content, etc. The washed sample is used to study the chips for lithiological logging. Small amounts of the unwashed sample may be used at a later date, if the original sample is lost, for geochemical studies (alteration haloes, arsenic content, etc.) APPENDIX III

REVERSE CIRCULATION DRILL LOGS

NORANDA EXPLORATION CO. LTD.

DIAMOND DRILL LOG

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HOL	E No.	(: KLIYUL .: KL93-1												
Col.	lar E	lastings: 2805.00	Colla	r.In	clina	tion:	-50,	.00			Log	czed by	GRC	
Col	lar N	orthings: 1730.00	Grid	Bear	ing:	0.0	00				Dat	e: MAR	9/93	
001	lar c	Levation: 1753.00	Final	neb	tn:	00.00	meti	es			DOV	an-noie	survey.	
							AS	SATS						
7208	10	LITHOLOGICAL DESCRIPTION	FROM	10	VIDTE	Cull	As gatl	Ag gatL	Culs	Au gets	Ag gals			
0	4	OVERBORDEN												
4	22	SILIC-CHLORITE ALT ANDESITE	- 4,00	6,00	2,00				0.012	0,005	0,200			
		Slightly calcareous.	8.00	10.00	2.00				0.015	0,005	0.200			
		Trace to 11 pyrite, trace magnetite.	12,00	14.00	2.00					. 0,005	0,200			
			16.00	18,00	2.00				0.016	0,005	0,200			
			20.00	22,00	2,00				0,190	0.190	0,200			
22	30	SILICEOUS ANDESITE	24.00	26.00	2.00				0.086	0,050	0,200			
		1-21 pyrite, trace magnetite,	- 28.00	30.00	2.00				0.090	0.070	1.200			
		trace chalcopyrite.								×.,				
30	50	CALC-SIL ALT ANDESITE	30.00	32.00	2.00	0.105	0,110	0.200			1			
		Noderately calcareous.	32.00	34.00	2.00				0.074	0,100	0,200			
		1-21 pyrite, trace to 11 sagnetite.	34.00	36.00	2.00	0.174	0,140	0,400						
			36.00	38,00	2,00				0,206	0,150	0,200			
			38.00	40.00	2.00	0,163	0,130	0,800						
		The second s		42.00.	2,00				0.142	0,130	0,200			
			42.00	44,00	2,00	0.104	0,090	0,200						
			44.00	46,00	2.00				0.116	0.060	0.200			
			46.00	48.00	2,00	0.100	0.070	0,200						
			48.00	50.00	2.00				0.176	0,140	0.210			
50	58	CALCARROUS ANDESITE	. 50.00	. 52.00	. 2.00.		0.070	.0.200						
		Strongly calcareous.	52.00	54,00	2,00				0.154	0,120	0.200			
		1-21 pyrite, trace - 11 magnetite.	54.00	\$6.00	2.00	0.082	0.080	0.200						
			56.00	58.00	2.00				0.093	0.050	0,200			
58	82	CELORITE-SILIC ALT ANDESITE	60,00	62.00	2.00				0.091	0.030	0.200			
		Slightly calcareous.	64.00	.66.00	2.00				0.070	0.040	0.200			
		31 pyrite, <1-21 sagnetite,	68.00	70.00	2.00				0.158	0.050	0,200			
		Nagnetite increases where silica	72.00	74.00	2.00				0.173	0.120	0.200			
		flooding intensifies.	76.00	78.00	2.00				0.077	0.080	0.200			
		74-76 Calcite stringers.	78.00	80.05	2.00	0.094	0.050	0.400						
		•	80.00	82.00	2,00	1			0.185	0.120	0,200			
82	83	RACKSTITE STARR?	82.00	84.00	2.00	0.221	0.270	0,400	· ·					
		Najor increase in sagnetite	84,00	86.00	2.00				0.139	0.300	0,200			

BOLE No: KL93-1

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NORANDA EXPLORATION CO. LTD.

DIAMOND DRILL LOG

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PROPERTY: KLIYUL HOLE No.: KL93-1

FROK	10	LITEOLOGICAL DESCRIPTION	FEOK	10	WIDTR	ASSATS Cult do cott de cott	Culs Au gets Ag gets	
		content as well as carbonate. 10-121 magnetite, 31 pyrite.	86.00	88,00	2.00			

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10-121 magnetite, 31 pyrite. 84-88 Fault zone. Open space rods squeezed - abandon hole.

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BOLE Ko: K193-1

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Page 2
DIAMOND DRILL LOG

HOLE No.: KL93-2		· · · · · · · · · · · · · · · · · · ·	
Collar Eastings:	3150.00	Collar Inclination: -50.00	Logged by: GRC
Collar Northings:	1900.00	Grid Bearing: 0.00	Date: MAR 9/93
Collar Elevation:	1743.00	Final Depth: 112.00 metres	Down-hole Survey:
Grid: METRIC		Reverse Circulation	Midnight Sun Drilling

PROPERTY: KLIYUL

FEOM	TO	LITHOLOGICAL DESCRIPTION	FROM	10	VIOTE	Cult	AS AN ENLL	SAYS Az zatl	Culs	Ac	Az zet5			
		AUF5125574												
4	1.2	CTLICEONE ANDREIPE	6 00		1.64	4 197			4 110					
	10	Rea to clightly esterees	0.00	10.00	2,00	0.120	0.100	0.400	4.128	0,150	0.200			
		laff av 1-27 avenutite	10.00	10.00	2.00	0.107	0.100	0.400	4 160					
		tere gr, tere auguetite.	15.65	11.00	1.00	0.302	0.100	0.400	0,120	4.234	0.200			
			14.00	16.00	2.00	0.107	0,200	0.400	0 170	A 184	1 100			
			16 65	18.00	3.00	0.116	0.110	0.000	0.410	0.300	1.400			
			10.00	10,44	4,40	0.240	. 4.134	. 0.200						
8	32 .	SILICEOUS ANDESITE	 . 18.00	20,00	2.00	0.309	6,280	0.400	0.277	0,180	0.200			
		Moncalcareous.	20.00	22.00	2,00	0.387	0,460	1,600						
		2-31 py, 41 segnetite.	22.00	24.00	2.00	0.408	0.610	0.800	0.375	0.670	0,200			
		30-32 Quartz stringers.	24,00 .	26.00	2.00	0.276	0,220	0.800						
			26.00	28,00	2.00	0,269	0.250	0.400	0.304	0.250	0.400			
			28,00	30.00	2.00	0,262	0,230	0.200						
			 30.00	32.00	2,00	0.257	0,360	. 1,600	0.185	0,160	0.400		 	
2	40	CALC-SIL ALT. ANDESITE	32.00	34.00	2.00	0,100	. 0.240	1,200	0.095	0.220	0,200			
		Node-ately calcareous.	34.00	36.00	2.00				0.077	0.110	0.200			
		2-31 py, 1-21 sagnetite, trace chalcopyrite.	36.00	38.00	2.00				0.107	0.080	0.200			
		38-40 Quartz stringers.	38,00	40,00	2,00			•	0.067	0.040	0,200			
	4	STLTCROUS ANDESITE	 40.00	42.00	2.00				0.147	0 110	0.200		 	
· · · ·		Slightly calcareout.	42.00	14.00	2 00				0 101	0 070	0 200			
		11 purite trace - 21 sagnetite	44.00	46.00	2 00				0.129	0 0.80	0.200			
		40-48 Oparts stringers.	14.00	18.00	2.00				0.117	0.060	0 200			
		te te çente straçtist	42.00	50.00	2.00				0.090	0 010	0 200			
			50.00	52.00	2 00				0.094	0.050	0 100			
			52.00	54.00	2.00				0.089	0,100	0.200			
			61 MA											
	14	SIGICK-CROBILE ALL, ADDESITE	24.00	20.00	2.00				0.033	0,150	0.200			
		Reveracely calculations,	38,00	30.00	2,00				0.078	0,010	0,200			
		ariges increase in epicole,	35,00	60.00	2.00				0,133	0.070	0.200			
		se pyrite, trace to re segletite,	-, 00.00 ·		1.00	1			0.077	0.050	0,200	-		
		tiece to is chalcopylice.	61.00	64.00	2.00				0.228	0.170	0.800			
			04.00	00.00	2.00				0.133	0.120	0.200			

R01.8 No: \$193-2

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DIAMOND DRILL LOG

PROPERTY: KLIYUL HOLE No.: KL93-2

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_____ ASSATS Cull Au gatt Ag gatt Cull Au gats Ag gats LITROLOGICAL DESCRIPTION 720M 10 VIDTE 10 FROK 55.00 68.00 2.00 0.251 0.240 0.200 68,00 70,00 2,00 0.194 0.180 0.200 0.176 0.280 70.00 72.00 2.00 0.400 0,165 : 5,800 12,400 72.00 74.00 2.00 6.000 0.156 1.200 74.00 76.00 2.00 74 86 CALC-SIL ALT. ANDESITE 0,14 0.250 0.200 76.00 78,00 Roderately calcareous. Increase 0.157 0.270 0.200 78.00 80.00 2.00 in sagnetite content. 0,124 0,280 0,400 80.00 82.00 2,00 3-41 pyrite, 6-81 sagaetite, 0,200 82.00 84.00 2.00 SIL 0,145 0,310 trace to 11 chalcopyrite. 0.111 0.170 0,800 84.00 86.00 2.00 82-86 101 magnetite & quartz stringers. 2.00 0.028 0.030 0.200 SILICEOUS ANDESITE 86 92 88.00 50.00 2.00 0.047 0.160 0.400 Slightly - moderately calcareous. 92.00 2.00 0.074 0.140 0,200 90.00 21 pyrite, 21 sagnetite. 1,600 92.00 94.00 2,00 0.087 0.120 112 SILICA-CELOBITE ALT. ANDESITE 92 96.00 2.00 0.154 0.230 0,400 94.00 Slightly calcareous. Increase in . 58.00 chloritized mafics and fractures. 0,200 48.00 100.00 2.00 0.109 0.080 1-31 pyrite, 1-21 sagnetite. 0.200 0.138 0.150 0.200 100.00 102.00 2.00 0.139 0.230 92-94 Quartz stringers. 0.200 0.254 0.380 0.400 102.00 104.00 2.00 0.230 0.320 96-108 Quartz stringers. 0.200 0.129 0.180 0.200 106.00 2,00 0.133 0.230 104.00 £.0.8. 0.159 0.240 0.400 0,200 0.200 106.00 108.00 2.00 0.168 0.200 .108.00 110.00 0.1340.180 110.00 112.00 2.00 0.153 0.220 0.200 0.157 0.180 0,400

BOLE No: KL93-2

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DIAMOND DRILL LOG

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PROPERTY: KLIYUL HOLE No.: KL93-3 Collar Eastings: Collar Northings: Collar Elevation:	3132.00 1992.00 1735.00		Collar Inclination: -50.00 Grid Bearing: 33.00 Final Depth: 60.00 metres	Logged by: GRC Date: MAR 9/93 Down-hole Survey: Midnight Sun Drilling
Grid: METRIC			Reverse Circulation	Midnight Sun Drilling

							15	SATS			
F818	T0	LITHOLOGICAL DESCRIPTION	FEOK	TO	VIDTE	Cull	Au gatt	As sail	Culs	Au gats	AE ENTS
	8	OVERBUIDEN									
	14	SILICEOUS ANDESITE	8.00	10.00	2,00	0,383	2,300	1,200	0.310	1,600	1,200
		Fine grained, silica flooded.		12.00	. 2.00	.0.178	0,200	0.400	0.178	.0.370	0.800
		Slightly calcareous. 2-31 ovrite.	12.00	14.00	2,00	0,193	0,800	0,200	0.068	0,310	0.200
	34	CALC-SIL ALT ANDESITE	14.00	16.00	2.00				0.066	0.050	0.200
		Koderately calcareous.	16.00	18,00	2.00				0.053	0.060	0,400
		21 pyrite, 1-21 sagnetite.	18.00	20.00	2.00				0.107	0.060	0,400
		14-16 Sericitic fractures.	20.00	. 22.00	. 2.00					0.030	0,400
		16-18 Quartz stringers.	22.00	24.00	2,00				0.095	0,040	0,200
		22-24 Quartz stringers.	24.00	25.00	2.00				0.090	0.060	0,400
		26-28 Quartz stringers.	26.00	28.00	2.00				0.128	0.060	0.400
			28.00	30.00	2.00				0.061	0.040	0,400
			30.00	32.00	2.00				0.095	0,100	1,200
				34.00	- 2,00 -				.0.087	.0.050	. 0,800
	42	CALCAREOUS ANDESITE	34.00	36,00	2,00				0,070	0,060	0,400
		Strongly calcareous.	36.00	. 38.00	2.00				0.054	0.040	0.800
		21 pyrite, trace sagnetite.	38.00	40.00	2.00				0,102	0,060	0,400
			40.00	42,00	2.00	0.087	0,100	0,200	0.085	0,140	0,800
	60	SILICA-CELORITE ALT ANDESITE	42.00	44.00	2.00	0.117	0,160	0,200	0.115	0,100	0,800
		Slightly calcareous.	44.00	46.00	2.00	0,211	0,150	0,400	0,212	0,200	1,200
		1-21 pyrite, 11 magnetite, trace	46.00	48.00	2.00	0,145	0,260	0,200	0.155	0.280	0.800
		chalcopyrite.	48.00	50.00	2.00				0.043	0.050	0.400
		42-44 Quartz stringers.	50.00	52.00	2.00	0.063	0.050	0.200	0.066	0.050	0.400
		46-48 Quartz stringers.		.54.00	2.00	. 0.132	0.140	0.200	0.132		.0,800
		50-52 Quarte stringers.	54.00	56.00	2.00	0.117	0.070	0.200	0.131	0.080	0,400
		58-60 Quartz stringers.	56.00	58.00	2.00	0.156	0,100	0.400	0.073	0.040	0,400

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ROLE No: 1193-3

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DIAMOND DRILL LOG

PROPERTY: KLIYUL HOLE No.: KL93-4			
Collar Eastings:	2790.00 1937.00	Collar Inclination: -50.00	Logged by: GRC
Collar Northings:		Grid Bearing: 50.00	Date: MAR 9/93
Collar Elevation:	1737.00	Final Depth: 120.00 metres	Down-hole Survey:
Grid: METRIC		Recerse Circulation	Midnight Sun Drilling

			 				15	SAYS				
FROM	10	LITEOLOGICAL DESCRIPTION	· 7108	TO	VIDTE	Cult	Au gatL	As sott	Culs	Au gols	As sals	
0 10	10 22	OVERBORDEN CALC-SIL ALT ANDESITE Strongly calcareous 1-21 pyrite, 31 magnetite, trace chalcopyrite. 10-12 Quartz stringers. 18-22 Quartz stringers. 18-22 Trace to 12 chalcopyrite.	10.00 12.00 14.00 16.00 18.00 20.00	12.00 14.00 16.00 18.00 20.00 22.00	2.00 2.00 2.00 2.00 2.00 2.00 2.00	0.053 0.023 0.040 0.196 0.378 0.261	0.050 0.020 0.010 0.070 0.200 0.130	0.400 0.400 0.400 1.600 0.400 3.200	0.060 0.032 0.050 0.345 0.477 0,223	0,040 0,010 0,110 0,130 0,280 0,120	0.400 0.400 0.800 3.200 4.800 2.800	
22	32	SKARMED ANDESITE Strongly calcareous. Increase in magnetite, chlorite 4 epidote. 1-21 pyrite, 4-51 magnetite. 30-32 Quartz stringers.	 22.00 24.00 16.00 28.00 30.00	24.00 26.00 28.00 30.00 32.00	2.00 2.00 2.00 2.00 2.00	0.113 0.078 0.048 0.049 0.087	0.090 0.050 0.020 0.040 0.070	1.600 0.800 0.400 1.200 1.200	0.059 0.052 0.053 0.081 0.109	0.030 0.020 0.040 0.080 0.110	1.200 0.800 1.200 1.200 1.600	:
12	. 60	CELOEITE-CAES ALT ANDESITE Increase in chlorite content. 21 pyrite, 11 sagnetite, trace chalcopyrite.	 32.00 34.00 36.00 38.00	34.00 36.00 38.00 40.00	2.00 2.00 2.00 2.00	0.200 0.157 0.139 0.413	0.110 0.140 0.120 0.240	1.600 2.400 2.000 5.600	0.154 0.139 0.337 0.178	0,280 0,130 0,240 0,180	2.400 2.000 4.000 2.800	 •
40	50	INTERSE CALC-SIL ALT ANDESITE Extremely flooded to very pale green color. 21 pyrite.	-40.00 42.00 44.00 46.00 48.00	42.00 44.00 46.00 48.00 50.00	2.00 2.00 2.00 2.00 2.00 2.00	0.159 0.162 0.256 0.179	0,180 0,170 0,520 0,370	2.800 2.800 4.600 3.200	0.050 0.200 0.268 0.235 0.235	0.100 0.240 0.540 0.240 0.240 0.380	1.200 2.800 4.000 1.200 3.200	
50	120	SKARMED AMDESITE Dark green, dense, fine grained, magnetite and chlorite rich andesite. Slightly calcareous. 1-31 pyrite, 8-101 magnetite, trace to 11 chalcopyrite. 70-76 Increase in pyrite to 51. 62-50, Quartz stringers. 100-102 Quartz-sericite altered zone. 116-118 Quartz-sericite altered zone.	50.00 52.00 54.00 58.00 60.00 62.00 64.00 66.00 68.00	52.00 54.00 56.00 58.00 60.00 62.00 64.00 66.00 68.00 70.00	2.00 2.00 2.00 2.00 2.00 2.00 2.00 2.00	0.242 0.244 0.403 0.300 0.252 0.446 0.244 0.183 0.157 0.527	0.330 0.320 0.670 0.270 0.280 0.520 0.340 0.170 0.200 3.000	3.600 2.800 3.400 2.400 2.400 2.800 1.600 1.200 1.200 4.800	0.291 0.481 0.446 0.354 0.401 0.551 0.266 0.225 0.186 0.405	0,340 0,620 0,510 0,320 0,410 0,700 0,350 0,220 0,250 1,600	2.800 3.600 5.200 2.400 2.800 3.200 1.600 1.600 1.200 2.800	

EDLE No: KL93-4

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DIAMOND DRILL LOG

PROPERTY: KLIYUL HOLE No.: KL93-4

ASSAYS FEOM TO WIDTE Cull As gall Ag gall Cols An gats Ag gals FROM 10 LITHOLOGICAL DESCRIPTION 70,00 72.00 .2.00 0.700 2.700 4,800 0.849 5,600 6,000 6.050 0.879 3,200 6,400 72.00 74.00 2.00 0.799 2,900 74.00 76.00 2.00 0.444 1,200 2,400 0.499 1,300 3,200 76.00 78.00 2.00 0.695 2.000 6,800 0.476 1.210 4,000 0.265 0.850 2,000 0.700 2.000 78.00 80.08 2.00 0.305 2,000 80.08 82.00 2.00 0.551 2.200 3.200 0,432 1,800 .82.00 84.00 2.00 0.185 0.650 . 0.800 0,191 0.660 1,200 86.00 0.258 0.800 1,600 0.555 3.500 3,200 84.00 2.00 1,500 5,600 2.00 0.520 2.300 2,400 0.576 86.00 88.00 88.00 90,00 2.00 0.294 0.570 2.400 0.239 0.540 2,800 0.270 2.000 92.00 2.00 0,195 0.190 1,200 . 0,207 90.00 0,516 0.980 4,400 2.00 0.276 0.310 1,600 92.00 94.00 96.00 ... 2.00 0.241 0,430 1,600 0.256 0.410 2,000 94.00 0.194 0.460 2,000 96.00 98,00 2,00 0,177 0,270 1,200 0,300 2,000 0,201 0,320 2,400 0.182 98.00 100.00 2.00 100.00 102.00 2.00 0.204 0,400 3,600 0,186 0.410 4,400 102.00 104.00 2,00 0.075 0.110 0.400 0.081 0,120 1,200 0,109 0,120 0,400 104.00 105.00 2.00 0.120 0.140 0.400 106.00 108.00 ... 2.00 . 0.233 0.340 0.166 0.230 1.200 1,600 0,218 0,730 108.00 110.00 2.00 0.116 0,160 0.400 2.00 0.133 - 0.140 0.400 0,124 0,120 0.800 110,00, 112.00 1,200 112.00 114.00 . 2.00 0.214 0.280 0,400 0.194 0.430 080.0 000.0 0.800 114.00 116.00 2.00 . 0.138 0,140 0.400 2.000 0,250 . 1,200 0.096 0.310 -116.00 118.00 2.00 0.252 118.00 .120.002.00 ...0,204.....0,260....1,600 .0.225 0.300 2.000 ~

BOLE No: \$193-4

Page 2

DIAMOND DRILL LOG

PROPERTY: KLIYUL		1	
Collar Fastings:	2655.00	Collar Inclination: -50.00	Logged by: GRC
Collar Northings:	1930.00	Grid Bearing: 60.00	Date: MAR 9/93
Collar Elevation:	1737.00	Final Depth: 100.00 metres	Down-hole Survey:
Grid: METRIC		Reverse Circulation	Midnight Sun Drilling

FROM	70	LITBOLDGICAL DESCRIFTION	 TROM	70	VIDTH	Cull	AS Au gath	Ag gatt	Cuis	Au gat5	Ag gats	 	 	
	12	OTTO LIBRATE												
12	22	CALC-SIL ALT ANDESITE	12.00	14.00	2.00									
		Strongly calcareous.	14.00	16.00	2.00	0.137	0,180	0,400	0.173	0,180	1.200			
		2-31 pyrite, 7-81 magnetite, trace to	16,00	18.00	2.00	0.399	0.570	2,800						
		11 chalcopyrite.	18.00	20.00	2.00	0.405	0.860	10,800	0.461	0,300	22,400			
			20,00	22,00	2,00	0.324	0.670	4,000	0,354	0,640	4,800			
22	32	LINESTONE BRECCIA	22,00	24.00	2.00	0,200	0,650	2,000	0.214	0,490	2,400			
		Limestone clasts within andesitic matrix.	24,00	26.00	2,00	0.135	. 0.180	. 0.400	0.154	0,240				
		Strongly calcareous.	26.00	28.00	2.00	0.081	0,100	0,400	0.061	0,100	1.200			
		1-21 pyrite, 41 sagnetite, trace chalcopyrite.	28,00	30.00	2,00	0.235	0.350	2,000	0,208	0,360	2,410			
			30.00	32.00	2.00	0.278	0.400	2,400	0,231	0,360	2,400			
32	44	CALC-SIL ALT ANDESITE	32.00	34.00	2.00	0.368	0.430	2,400	0,384	0,430	2,800			
		Strongly calcareous.	 14.00	36.00		. 0.184	_0.250		0.185	.0.250	1,600	 		
		11 pyrite, 5-71 sagnetite, trace to	36,00	38,00	2.00	0.552	0.680	7.200	0.868	0,940	9,200			
		11 chalcopyrite.	38,00	40,00	2.00	0.297	0,900	8,800	0,269	0.810	8,400			
			40.00	42.00	2.00	0.265	0.630	6,400	0.252	0.820	4,000			
			42.00	44,00	2.00	0,245	0,840	3,600	0.085	1,100	4,810			
44	48	QUARTZ/CALCITE VEIN STOCKVORK	_ 44.00	46.00	. 2.00	0.157	0.630			0,810	5.200	 	 	
		50-601 quartz/calcite vein material	46.00	48,00	2,00	0.117	1,200	6,400	0.295	0,650	5,600			
		10 storsite tost.												
48	62	CALC-SIL ALT ANDESITE	48.00	50.00	2.00	0.274	0.810	6.000	0.172	0.450	2.800			
		Strongly calcareous. Increase in	-30.00	52.00	2.00	0.195	0,630	3,200	0.283	1.000	4,000			
		sagnetite. Quartz-calcite stringers throughout.	. 52,00	54.00	2.00		.1.500	3,600	0.303		4.010			
		2-31 pyrite, 8-101 magnetite, trace to	54.00	56.00	2.00	0.229	1.200	2,800	0.277	1,100	3.600			
		<11 chalcopyrite.	56,00	58,00	2,00	0.240	1,600	4,000	0,273	2,400	5.200			
			\$8,00	60.00	2.00	0.227	2.400	4,800	0.222	2,000	5,600			
			60.00	62.00	2.00	0.154	1,700	3,600	0.127	2,000	6.400			
62	70	MAGHETITE SRARN	62.00	64.00	2.00	0.276	8,200	6,800	0.240	. 8,200	7,200			
		Fine grained silics & magnetite flood zone.	64.00	66.00	2.00	0.465	3,400	4,000	0.398	3,000	4.000			
		41 pyrite, 20-301 magnetite, <1 to 21	66.00	68,00	2.00	0.616	4,100	6,000	0.535	4,600	6.000			

HOLE No: KL93-5

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DIAMOND DRILL LOG

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PROPERTY: KLIYUL HOLE No.: KL93-5

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FROM	TO	LITROLOGICAL DESCRIPTION	FROM	TO	VIDTS	Cull	AS Au gatL	SATS As satL	Cuis	Au gats	At tats	
		chalcopyrite. Koderately calcareous.	68,00	70,00	. 2.00	0.179	0.080	0,800	0.120	0.490	1.600	
70	12	QUARTZ-SERICITE ALT ZONE Slightly calcareous with quartz, calcite stringers.	70,00	72.00	2,00	0,087	2,000	2,000	0,038	1,600	3.200	
12	76	CALC-SIL ALT ANDESITE	72.00	74.00	2.00	0.057	1.500	3.600	0.057	1.300	4,800	
		Quartz, calcite stringers throughout. 21 pwrite, 21 mannetite.	74.00	76.00	2,00	0.025	0.220	0.200	0.024	0.160	1,200	
76	78	QUARTZ-SERICITE ALT 2082 Same as 70-72 m. 21 moving 1-22 magnetite.	76.00	78,00	2.00	0.024	0,260	0,200	0.02?	0.240	0.400	
78	100	CALC-SIL ALT ANDESITE	78.00	80.00	2.00	0.024	0.060		0.021	0.070	0,200	
		Slightly to moderately calcareous.	00,08	82,00	2.00	0.031	0,060	0,200	0.034	0.080	0,400	
		1-21 pyrite, trace to 21 magnetite.	82.00	84.00	2.00	0.076	0.250	0.800	0.083	0.350	1,200	
		82-84 Quartz-sericite altered zone.	84.00	85.00	2.00	0.021	0.110	0,200	0.014	0.080	0.200	
		88-90 Quartz-sericite altered zone.	86.00	88,00	2.00	0.750	2,400	6.000	0.827	1,500	6,000	
			88.00	90.00	2.00	0.039	0.280	1,600	0.036	0.230	0,800	
			90,00	.92,00	2.00		0,120		. 0.061	0.140	. 0,800	
			92.00	94.00	2.00	0.015	0.050	0,200	0.013	0.050	0.200	
			94.00	96.00	2.00	0,036	0.190	1,200	0.046	0.180	0.800	
			96.00	- 58,00	2.00	0.077	0,310	2,800	0.087	0.240	2,400	
			98,00	100.00	2.00	0.022	0,140	1,200	0.022	0,180	0,400	

BOLE No: KL93-5

DIAMOND DRILL LOG

PROF HOLE Coll Coll Grid	erty No. lar E lar N lar E l: ME	f: KLIYUL .: KL93-6 Eastings: 2727.00 Northings: 1880.00 Elevation: 1737.00 ETRIC		Colla Grid Finai Reven	ar In Bear Dep rse C	clina ing: th: ircul	tion: 60.0 80.00 ation	-50. 0 metr	00 es			Log Dat Dow Mid	ged by: GRC e: MAR 10/93 m-hole Survey: night Sun Drill:	ing	
7808	TO	LITHOLOGICAL DESCRIPTION		FEON	T0	VIDTH	Cull	ASS Au gati	ATS AT LALL	Culs	Au gats	Ar rats			
		AUT010000													
12	42	CALC-SIL ALT ANDESITE		12.00	14.00	2.00	1			0.024	0.070	0.200			
		Strongly calcareous.			16.00	2.00	· · ·			0.027	0.020	0.200			
		1-21 pyrite, trace to 11 magnetite.		16,00	18.00	2.00									
		12-20 Eematite stained fractures,		18,00	20.00	2.00				0.068	0.010	0.200			
		Ji pagnetite (lault?).		20,00	22.00	2.00				0.010	0.010	0.200			
				26.00	26.00	2.00				0.012	0,050	0.400			
				25.00	28.00	2.00				0.045	0.040	0.200			
				28,00	30.00	2.00				0.053	0.050	0,400			
				30,00	32.00	2.00	•			0.044	0.040	0,400			
				32.00	34.00	2.00				0.037	0.040	0.200			
				34,00	36.00	2,00				0.025	0.030	0.400			
				35,00	38.00	2.00				0,099	0,090	0,400			
					40.00	2.00				0.029	0.030	0,200			
				40.00	42,00	4.00				0.001	0.030	0.000			
42	50	ALTERED ANDESITE		42.00	- 44.00	2.00				0.039	0.030	0.200			
		Chlorite, sericite increases.		44.00	46.00	2,00				0.075	0,110	0.800			
		Slightly calcareous,		46.00	48.00	2.00				0.045	0.120	0.400			
		31 pyrite, trace sagnetite.			50.00	. 2.00 .				.0.075	0.030	0.800		· · · · · · · · · · · · · · · · · · ·	
50	56	CALC-SILIC ALT ANDESITE			52.00	2.00				0.070	0.080	0.400			
		Increase in pervasive carbonate.		52,00	\$4,00	2.00				0.110	0.070	1.600			
		Roderately to strongly calcareous.		54.00	56.00	2.00	0.235	0.110	2,400	0.220	0,100	1.600			
		21 pyrite, trace magnetite, trace													
		chalcopyrite.													
4	62	feineteiten unbeite		55 88	59.00	2 65	0 114	0.150	1 204	0 101	0 110	1 640			
~	**	Barker ereen, denser, increase in		58.00	60.00	2.00	0.282	0.200	3,200	0.259	0,170	2.400			
		sagnetite content.		60.00	62.00	2.00	0,191	0.180	1,600	0,189	0,130	1,200			
		Slightly to noderately calcareous.	1	1											
		21 pyrite, 2-31 sagnetite, trace													
		chalcopyrite.													
62	70	QUARTZ-SERICITE ALT ZONE		62,00	64,00	2.00				0.150	0,240	2,400			

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DIAMOND DRILL LOG

PROPERTY: KLIYUL HOLE No.: KL93-6

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FROM	TO	LITEOLOGICAL DESCRIPTION	FROM	ŤŨ	VIDTE	Cull	Au gmtl	As soil	Culs	Au getS	As sets
		Noderately calcareous.	64.00	66.00	2.00	0,181	1.00	4.000	0.223	0.550	4,800
		4-51 pyrite, 1-21 pagaetite.	66.00	68.00	2.00	0.217	0,300	2.400	0.323	0.340	2.400
		66-68 Less altered, <11 cpy.	68.00	70.00	2.00	0,118	0,200	2.000	0,176	0,260	1,600
70	80	CELORITIZED ANDESITE	70.00	72.00	2.00	0.095	0.120	1,600	0,122	0.130	0,800
		Slightly calcareous.	72.00	74.00	2,00	0.201	0.210	1.200	0.235	0.190	1,200
		2-31 pyrite, 4-51 margetite, trace chalcopyrite.	74.00	76.00	.2.00	0.169	0.190	1.200	0,193	0.180	1,200
			76.00	78.00	2.00	0.247	0.240	2.400	0.221	0.190	1,600
			78.00	00.08	2.10	0.319	0.320	2.400	0.274	0.270	2,000

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BOLE No: KL93-6

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APPENDIX IV

SAMPLE DESCRIPTIONS/ASSAY SHEETS

PROJECT # 118

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N.T.S. <u>940/8+9</u> DATE <u>FC820/93</u> Ł

LAB REPORT # _____

PROJECT KLINDL RC-KL-93-1

ROCK SAMPLE REPORT

SAMPLE NO.	LOCATION & DESCRIPTION	% Sulph.	TYPE	WIDTH (m)					SAMPLED BY
70252	4-6 meters	-	CORECTUPS	2.0			<u> </u>		
70253	6-8 1		1						
702.54	8-10								
70255	10-p								
70256	(2-(4-))						L	 []	
70257	14-16								
70258	16-18								
70259	<u> (£-x0</u>							 	
70260	20-22						<u> </u>		
70261	22-24								
20262	19-26						<u> </u>	 	
70263	26-28								
70264	28,-30								
70265	20-32								
70266	32-34				<u> </u>				
7026.7	34 - 36								
70268	36-39			l					
70269	38-10								
70770	40-42								
1120271	42-44								
27200	44-16								
70273	<u>+6-48</u>								
70274	19-50								
276775	50-52								
30,505	52-54 V		<u> </u>			_			
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PROJECT # _ { +8 . ____

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N.T.S. <u>940/8+9</u> DATE <u>Feb20/93</u>

LAB REPORT # _____

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PROJECT KLIYUL RE-KL-93-1

ROCK SAMPLE REPORT

SAMPLE NO.	LOCATION & DESCRIPTION	% Sulph.	TYPE	WIDTH (m)				SAMPLED BY
70277	54-56 meters		CORE CHIPS	2.0				
70278	56-58							
70279	55-60							
70250	60-67						 	
70281	62-64					1		
70282	64-66					<u> </u>	 	
70283	46-68		-					
70284	68-70						_	
70255	70-72							
70256	72-74			II			 _	
70297	79-76	· · · · · <u>· · · · · · · · · · · · · · </u>	· ·		i		 	4
70288	76-78							
70289	78-30		\vdash	!			 	
70290	50-82						 	+
70291	32-84							L
70292	84-86	ļ		<u> </u>				
70273	196-88 V E.C.H.	<u> </u>	¥ V	20		···	 <u> </u>	
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N.T.S. 940/8+9

LAB REPORT #

PROJECT KLINUL RC-46-93-2 .

ROCK SAMPLE REPORT

SAMPLE NO.	LOCATION & DESCRIPTION	% Sulph.	TYPE	WIDTH (m)				•		SAMPLED BY
70297	6-8 meters		GRECHIPS	2.0m						
74298	8-16									
70299	10-12									
70300	17-14					1 1				
70301	14-16					1				
70302	16-18	· · · · · · · · · · · · · · · · · · ·								
70303	18-20									
70304	20-22		· · · ·				 			
70305	22 - 24									
70306	24-26									
70307	26-28						 · . · · - · ·			
70.308	28-30									
70309	36-32					!	 			
70310	32 ~34		_		·	I			<u> </u>	
703/1	34-36						 			
70317	36-38					- 	 			
20.31.3	38-10					↓ . ↓				
70314	10-42					 	 			
70315	42-44									
70316	49-96	· · · · · · · · · · · · · · · · · · ·				↓	 			
1 703/7	46-48	- <u>}</u>					 			
70318	48-50						 			
70317	50-52					 	 			
70320	52-59	. <u> </u>	-							
7032/	54-56 V	<u> </u>	<u> </u>	V V			 			
			1							

DATE <u>Feb 22/93</u>

PROJECT # 148

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N.T.S. <u>94 D/879</u> DATE <u>Fth 22/93</u>

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LAB REPORT 🖌 _____

PROJECT KLIYUL RC-46-93-2

ROCK SAMPLE REPORT

SAMPLE NO.	LOCATION & DESCRIPTION	¥ Sulph.	TYPE	WIDTH (m)								SAMPLED BY
70322	56-58 neters		CORE 44175	2.00	1							
70323	58-60 1		1									
20324	60-62											
70325	42-64]				
70326	64-66				I	I				-		
76327	66-68						L					
20328	63-70	}					I	I				
7032.9	76-77											
20330	72-74											
70.331	74-76	Ì										·
70332	76-78	<u>}</u>										
-20333	78-50											
70354-	50-52				Ļ	<u> </u>		 .	ļ			·
70735	82-89					<u> </u>	<u> </u>					
20336	54-56				<u> </u>	<u>[</u>	ļ	ļ				
70337	\$6+88					<u> </u>	<u> </u>					
70338	88-90		↓		i	1	<u> </u>		<u> </u>			·
70337	96-92		ļ.				{	ļ				
20340	92-94		<u>} </u>	L	ļ		į	_				
- 70391	94-96				i							
70342	96-18		<u>] </u>		I	ļ	 	}				
76343	98-00				· ·		_	<u> </u>				
27344	100-102	.l	1	ļ	I		<u> </u>	1			<u> </u>	
70345	102-104	ļ <u>.</u>			 		<u> </u>	l			<u> </u>	
70346	104-100 V	1	V.	<u>F. d</u>	L	ļ.,	L	.				

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N.T.S. <u>940/279</u> DATE <u>Ee523/22</u>

LAB REPORT # _____

PROJECT KLIYUL RC-KL-93-7

ROCK SAMPLE REPORT

SAMPLE NO.	LOCATION & DESCRIPTION	¥ Sulph.	TYPE	WIDTH (m)								SAMPLED BY
70347	166-108 meters		CODE CMP	2.0								
23 +8	108-110				<u> </u>	[
20349	110-112 V E.O.H.			<u> </u>	<u> </u>	ļ			<u> </u>			
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ROCK SAMPLE REPORT

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SAMPLE NU.		SULPHIDES				510			ICP	(30e	(ener	() +	\mathbf{v}	87
7/1352	8-10 m.		COI C H	LE	2.	õ								
70353	in -12 m			١	T	-								
70354	16-14 m													
70355	14-16 m													· · · · · · · · · · · · · · · · · · ·
70356	16-18 m													
70357	18-20 m													
70358	20-22 M													
2359	22-24 m													
70360	24-26 m				-									
70361	26-28 m													
70362	28-30m									·				
70363	30-32 m		<u> </u>	<u> </u>			<u>_</u>							······
70364	32-34m	<u> </u>		<u> </u>										
70365	34-36 m								ļ					
7:36	36-38 m													
70367	38-40 m			_	$ \downarrow \downarrow$									
7028	40-42m				$\downarrow \downarrow$							<u> </u>		
76369	42-44 m	ĺ		ļ										
70570	44-46 m				\square									
70371	46-48 m		<u> </u>	<u> </u>									:	
70372	18-50 m		-	<u> </u>										
70373	50-52 M			-				 						
70374	52-54-m			3	1									

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NORANDA	EXPLORATION	COMPANY,	LIMITED

N.T.S. 24D/ DATE Feb 24/93

PROPERTY KLIYUL RC-KL-93-3

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PROJECT 148

ROCK SAMPLE REPORT

SAMPLE NO	LOCATION & DESCRIPTION	%	TYPE	WIDTH	<u> 6 🗆 A 🗆</u>		GOAO	G 🗆 A 🗋	G 🗖 A 🗌	GCAD	GOAD	SAMPLED
		SULPHIDES				Ict (30)+	AU /		•		87
70375	54-56m		chip	200								
70376	56-58m											
7/377	55-60 m		J									
		-										
		-				-						
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PROPERTY KLIYUL

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N.T.S. <u>940/9</u> 13 DATE <u>Feb 25/93</u>

ROCK SAMPLE REPORT

PROJECT 148

SAMPLE NO	LOCATION & DESCRIPTION	%	TYPE	WIDTH	GOAD	G 🗌 A 🔲	GOAD	G 🗆 A 🛄	G 🗌 🔺 🗍	G 🗌 🗛 🗋	G 🗖 A 🗋	SAMPLED
		SULPHIDES				Ic	0(30	+A	<u> </u>	•		BY
70378	10-12m		CORE	2.0.4		 						
10379	12-14 m											
70380	14-16 m				ļ							
70351	16-18 m		¦		<u> </u>			ļ				
70352	18-20 m						l					
703 83	20-22 m		<u> </u>				 					A 44 4414 M
70384	22-24 m			<u> </u>								
-10385	24-26 m			┝╌╿──╸	. <u> </u>							······································
70386	26-28 m											
70387	28-30 m											
70388	30-32 m		 									
70389	32-34 m	<u></u>		Ц				.				
70390	3-3-36 m			<u> </u>	ļ					····		
70391	36-38 m				<u> </u>	<u></u>						
70392	38-40 m		ļ_	<u> </u>								
763 93	40-42 m		 	 		 			,			
703 94	42-44 m		<u> </u>	<u> </u>								
70395	44-46 m											
703 96	46-48 m											
703 97	48-50 m] +								
703 98	50-52 m											
703 99	52-59 m					-						·
70400	54-56m		ΙV	1.1								

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NORANDA EXPLORATION	COMPANY, LIMITED

N.T.S. <u>940/9</u> D'ATE <u>F-10-25/93</u> PROJECT<u>148</u>

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PROPERTY KLINUL RC-KC-83-4

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ROCK SAMPLE REPORT

SAMPLE NO		%	TYPE	WIDTH	G 🗋 ∧ 🗋	G □ A □	G 🗌 A 🗌	G 🗆 A 🗖	G 🗆 A 🗖	G 🗋 A 🗖	G□∧□	SAMPLED
		SULPHIDES					ΖCP	(30)	+ A-	ا ر		BY
70401	56-58 m		CORR Chips	2.00								
70402	58-60 m			1								
70A03	60-62 m											
70404	62-64 m											
70405	64-66 m											
70406	66-68 m											
70407	68-70 m										:	
70:408	70-72 m											
70410	72-74 m											
70411	74-76 m											
70412	76 - 78 m											_
70413	78-80 m											
70414	80-82 m						···· · · · · ·					
70415	82-84 m				·							<u> </u>
70416	84-86 m		·									
70417	86-88 M											
70418	88 - 90 m											
70419	90-92 m											
70420	92 - 94 m											
70421	94 - 96 m											
70422	96 - 48 m											
70423	98 - 100 m		\/									
70424	100 - 107 m		V							1	Į	



PROPERTY KLIYUL BC-HL-93-4

N.T.S. 940/9 D'ATE Fib-25/93 PROJECT 148

ROCK SAMPLE REPORT

SAMPLE NO.	LOCATION & DESCRIPTION	%	TYPE	WIDTH	6 🗆 A 🗋	G 🗋 A 🗖	G 🗌 A 🗌	G 🗌 A 🗌	G 🗌 A 🗋	G 🗌 🗚 🗍	G 🗌 A 🗌	SAMPLED
		SULPHIDES				I I CF	<u>(30)</u>	t.A.	2	•		6 Y
70425	102-104 m	<u></u>	CARE	2.0M								
70426	104 - 106 m											
70427	106-108m											
70420	1085 110 m											
70429	110 - 112 M											
70430	112 - 114 m				~							
70431	117-116m											
70932	116-118 m											
70433	118-120.1						·					
· · · · · · · · · · · · · · · · · · ·												
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			··									
·	<u> </u>											
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	i		t <u></u>	i								·····

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•	NORAN	IDA EXPLO	RATI	ON COM	ИРАМ	IY,	LIMITEC	0		N	.T.S	94 D	19.	_ 1/2
	PROPERTY RC-KL-93-5	KLI	100	<u> </u>						D	ATE	Feb	26/9	3
	R	OCK S	AMP	LE F	REP	OR	Т			- P	ROJEC	r	18	
SAMPLE NO.	LOCATION & DESCRIPTION	% SULPHIDES	Т	YPE	wipt	тн.	<u>G 🗋 A 🗌</u>	G A	G A		G 🗆 A 🖸		G □ A □	SAMPLEC BY
70435	12-14m.		COR	Eurs	2.0	0							1	
36	14.16]	{			·· · _						
37	16- 18													
38	19-20													
39	20-22													
70440	22-24													
41	24-26													
42	26 ~ 28													
43	28-30													
44	30 - 32													
45	32-34									1				
46	34-36													
47	36 - 38								 <u>.</u>	 				
48	38 - 40													
49	40-42													<u> </u>
70450	42 - 44													
\$[44-46 -													
52	46 - 49,													
53	48 - 50													
54	50-52													
55	52-54													
56	54-56													
70457	56-58			-	્ટ	/								·

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PROPERTY KUNUL RC-KL-93-5

N.T.S. 940/9 2/2 DATE <u>Feb26/73</u> PROJECT 148.

"ROCK SAMPLE REPORT

			%	Γ __	YPE	win	тн	G∐ ∧⊡		G 🗌 A 🗋	G 🗋 A 🗌	G 🗆 A 🗔	G	G 🗖 A 🗍	SAMPLED
			SULPHIDES			(~	2		7	CP (3	e) r 1	L.			B Y
70458	58-60 m.			COR	e (PS	2.	0		<u> </u>						
59	60-62			۱ ا											
70460	62-6A									l					
61	64-66								<u> </u>						
62	66 - 68								<u> </u>						
63	68 - 70								ļ						
64	70-72								ļ						
65	72 - 74								<u> </u>						
66	74 - 76								 						<u></u>
67	76- 78	<u></u>													
68	79 - 80	<u></u>							<u> </u>						
69	80-82														
70470	82-84												. 	ļ	
ור	84-86														
72	86 - 88								<u> </u>						
73	58 - 90										···				
74	90-92								ļ						
75	92-94		· · ·								·····				
76	94 - 96														
77	96-98	, 		<u> </u>		_ _		n							
70478	98 - 100m.			4	·	ې ب			 _	· · · · - · - · ·					
·····				 					ļ					· · · · · ·	
		· · · · · · · · · · · · · · · · · · ·													

PROJECT # 148

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N.T.S. 940/6+9 DATE Feb 27/93

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LAB REPORT #

PROJECT KLIMUL RC-KL-93-6

ROCK SAMPLE REPORT

SAMPLE NO.	LOCATION & DESCRIPTION	¥ Sulph.	TYPE	WIDTH (m)							SAMPLED BY
70482	12-14 meters		CARE CHUPS	2.0							
70983	14-16 1		1								
70484	14-18										
70+85	18-20		<u> </u>	<u>} </u>	 			ì		l	
70486	26-22			r I				1	ĺ		
70+87	22-24										
70+88	24-26				 			<u> </u>			
76489	26-28							1			
70490	28-30			·				<u> </u>	<u> </u>		
70491	30-32									·	
70492	72-34		· · · · ·			<u></u>		ļ	ļ	l	
74.493	34-76										
70494	36-38			_	 				I		
70495	28-40										
70196	40-42										
70497	42-44						<u> </u>				
70498	44-96										
70199	46-49										
70500	16-50										
10501	50-52						t				
70502	52 - 54	}				_					
70503	54-56										
70504	56-58										
70505	50-60										
70506	6-62 1		Ú V	4							

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PROJECT # <u>(48</u>

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LAB REPORT # _____

PROJECT KLIYUL RC-KL-93-6

ROCK SAMPLE REPORT

SAMPLE NO.	LOCATION & DESCRIPTION	t Sulph.	TYPE	WIDTH (m)			*	SAMPLED BY
70507	62-64 metris		CHE CHIR	2.00				
70.508	64-66		l i					
70509	66-69.							
70510	68-70							
70511	70-72	`				1		
70512	72 - 74				 			l
705/3	74							
21514	76-78							
705/5	TE-80 F.O.H.							
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			1					
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<u> </u>								
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N.T.S. 94<u>0/8+9</u> DATE <u>Feb-27/93</u>

NORANDA VANCOUVER LABORATORY Geochemical Analysis

Project Name &	No.: KLIYUL - 148	Geol.: G.G	
Material:	159 RC (Bulk Samples)	Sheet: 1 of	4
Remarka:	* Sample screened @ ~35 MESH (0.5 mm)		
	¹⁰ Organic, A Humos, S Sullide		٨

Date completed: MAR. 21

Date received: MAR. 11

LAB CODE: 9303-016

Au - 10.0 g sample digested with squarregia and determined by A.A. (DL. 5 PPB)

ICP ~ 0.2 g sample digested with 3 ml HCtO4/HNO3 (4:1) at 203 °C for 4 bours diluted to 10 ml with water. Leeman PS3000 ICP determined elemental contents.

N.B. The major oxide elements and Ba, Be, Ce, La, Li, Ga are rarely dissolved completely from geological materials with this acid dissolution method,

Т.	SAMPLE	٨ı	٨g	Al	A	Ba	Be	Bi	Ċ.	Cđ	ŝ	S	Ċ		Fe	K	La	Li	Mg	Mn	Mo	Na	Ni	P	Pb	Sr	π	v	Ze
<u>o.</u>	No.	ррь	ppm	%	ppm	ррш	ppm	ppm	%	ppm	ppm	ppm	opm.	ppm	_ %	95	ppm	ppm	5	ppm	ppm	%	ppm	56	ppm	ppa	- %	рра	poth
	70265	110	0.2	4.44	2	337	0.4	5	1.61	0.2	52	15	22	1051	5.05	1.00	16	13	1.06	298	- 1	0.09	3	0.11	2	125	0.06	93	41
	70267	140	0.4	5.44	2	304	0.4	7	2.56	0.2	58	21	18	1741	5.41	1.13	15	17	1.58	558	- 4	0.11	8	0.09	2	141	0.10	156	61
	70269	130	0.8	6.16	5	401	0.4	5	2.14	0.2	- 53	27	13	1634	5.22	1.57	14	18	1.95	660	2	0.12	7	0.09	2	89	0.12	172	62
	70271	90	0.2	5.70	2	262	0.4	8	2.85	02	57	18	14	1039	5.09	0.95	13	19	2.04	626	. 3	0.10	8	0.08	2	132	0.12	171	62
	70273	70	0.2	5.58	2	206	0.4	9	3.07	0.2	56	28	56	1000	6.08	0.54	14	20	2 .5 5	639	4	0.08	29	0.09	2	169	0.27	197	66
	70275	70	0.2	6.03	5	215	0.4	9	3.17	0.2	60	25	40	966	5.81	0.71	15	21	3.01	721	2	0.08	25	0.09	2	146	0.27	200	65
	70277	80	0.2	6.07	2	299	0.5	10	2.35	0.2	55	26	51	815	5.92	1.15	13	22	2.61	730	4	0.07	18	0.08	2	120	0.17	168	74
1	70289	50	0.4	4,21	2	266	0.3	- 5	1.75	0.2	45	18	27	939	4.18	0.81	11	. 12	1.33	432	3	0.09	7	0.08	· 2	141	0.19	135	44
	70291	270	0.4	5.21	2	395	0.8	5	2.44	03	65	31	22	2212	5.44	1.30	17	16	1.63	570	8	0.09	8	0.09	2	124	0.29	187	74
	70297	100	0.4	4.78	4	600	0.6	5	1.25	0.2	48	16	30	1269	4.48	1.36	13	18	1.58	665	3	0.10	17	0.10	5	194	0.31	163	75
	70298	160	0.4	5.63	2	506	0.5	5	1.03	0.2	43	12	15	1687	4. 6 7	1.81	13	: 19	1.72	- 551	3	0.08	10	0.09	Ż	91	0.34	158	67
	70299	160	0.4	5.49	2	280	0.4	5	0.86	0.2	39	14	11	3619	6.04	1.40	11	25	1.99	377	1	0.08	5	0.10	2	63	0.28	164	47
	70300	260	0.4	6.09	3	338	0.4	5	1.54	0.2	49	14	12	3754	6.61	1.81	14	19	1.40	584	3	0.09	6	0.10	2	101	0.26	193	57
•	70301	210	0.8	5.52	2	350	0.4	5	1.24	0.2	47	17	11	4070	5.52	1.56	14	20	1.54	507	3	0.09	5	0.10	2	111	0.18	165	- 54
	70302	130	02	4.84	2	361	0.4	5	0.66	0.2	37	15	20	2456	4.88	1.51	13	17-	1.34	309	4	0.09	6	0.10	2	59	0.13	144	49
	70303	280	0.4	4.36	2	298	0.4	5	1.16	0.2	44	14	16	3089	4,67	1.10	12	18	1.37	305	2	0.09	7	0.10	2	116	0.18	150	41
•	70304	460	1.6	3.95	2	281	0.4	5	1.38	0.2	46	13	20	3867	6.53	0.92	13	16	1.11	365	2	0.09	6	0.09	2	152	0.20	168	43
•	70305	610	0.8	4.42	3	313	0.4	5	1.74	0.2	54	13	21	4063	5.03	1.01	15	17	1.30	482	2	0.08	7	0.10	2	170	0.22	150	51
	70306	Z20	0.8	3.72	2	301	0.9	5	1.05	0.5	59	19	24	2763	631	0.96	20	21	1.17	385	35	0.09	7	0.09	. 9	99	0.23	211	- 54
•	70307	250	0.4	3.86	2	339	0.4	5	1.06	0.2	44	14	15	2692	6.14	1.08	13	14	1.08	321	3	0.08	4	0.10	2	112	0.17	155	- 44
	70308	230	0,2	4.02	2	353	0.5	5	1.57	0.3	56	15	23	2624	5.69	0.75	14	18	1.54	523	3	0.09	13	0.11	4	209	0.23	156	62
	70309	300	1.6	3.82	2	35 5	0.5	6	1.60	1.1	60	15	15	2574	5 <i>_</i> 53	1.02	16	16	1.32	576	45	0.09	5	0.12	··. 7	- 97	0.11	164	71
•	70310	240	1.2	5.05	2	479	0.5	7	2.05	0.2	58	22	15	1003	5.79	1.29	15	21	1.65	- 598 :	8	0.11	6	0.10	20	90	0.08	178	- 78
	70337	30	0.2	3.88	2	722	0.7	5	2.64	0.2	70	11	14	278	4.22	1.28	19	12	1.02	812	1	0.09	5	0.12	2	165	0.13	107	71
	70338	160	0.4	3.53	3	730	0.6	5	2.31	0.2	70	15	20	470	4.63	1.06	19	12	0.98	786	3	0.11	6	0.12	3	191	0.18	117	70
:	70779	140	02	4.53	2	605	0.4	7	0.87	0'2	39	27	13	743	5.26	1.44	12	14	1.40	462	5	0.09	5	0.08	2	90	0.14	128	76
i.	70340	120	3.6	4.74	2	480	0.4	5	1.18	02	42	22	12	866	5.42	1.44	11	16	1.51	395	7	0.08	Ś	0.07	ः् <u>ट</u>	86	0.10	167	62
	70341	230	04	4.85	2	392	0.4	6	1.01	0.2	41	22	12	1542	5.53	1.41	12	18	1.54	329	4	0.08	Ś	0.07	ંડ	47	0.07	150	50
	70342	160	0.2	5.21	3	444	1.1	8	1.03	05	57	33	21	1241	5.66	1.59	18	26	1.55	318 2	5	0.11	11	0.08	8	42	0.06	179	53
	70343	80	0.2	5.04	2	445	0.4	š	0.99	02	44	29	14	1090	\$35	1.47	13	20	1.64	378	4	0.11	6	0.08	2	32	0.07	172	56
				2.0	_			-									. [
\$	70344	230	0:2	4.63	2	346	0.4	6	0.93	02	44	27	13	1389	5.67	1.31	13	19	1.63	323	7 ;	0.10	5	0.07	2	36	0.07	147	1
3	70345	320	02	4.68	2	339	0.4	6	0.94	0.2	42 -	~ 27	16	2296	5.35	1.34	12	19	1.79	356	19	0.10	7	0.07	2	28	0.06	145	- 50
i	70346	230	0.2	5.01	2	326	0.3	6	1.33	0.2	49	36	11	1327	5.81	1.23	13	19	1.67	355	6	0.10	5	0.08	2	94	0.07	162	92
	70347	200	02	4.61	4	309	0.3	9	1.51	0.2	54	34	14	1682	5.88	0.99	13	19	1.60	400	5	0.11	5	0.09	2	134	0.07	159	- 50
:	70348	180	0.2	5.06	3	_384_	_0.4	8	1.85	0.2	56	27	10	1649	5.58	1.17	14	17	1.59	445 :	3	0.11		0.09	2	147	0.10	159	<u> </u>

Ally GP 116 and

NORANDA DELTA

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ſ. T .	SAMPLE	An	٨	A	A	Ba	Be	Bi	Ca	Cd	Q:	Co	à	Cu	Fc	K	L	Li	Mz	Ma	Mo	Na	Ni	P	Pb	Sc	Ti	v	70 8303-016
×0.		ppb	ppm	_ %6	рра	ppm	ppat	рош	96	ррш	ррш	ppm	ppm	ppm	%	%	200	200	%	DOM	ppin	56	PDE	%	DOM	000	%	DOM	DOM Po. 2 of 4
-8	70349	220	02	4.36	2	285	0.4	8	1.86	0.2	56	32	14	1530	6.04	0.85	14	17	1.46	404	16	0.10	5	0.09	2	151	0.08	148	53
39	70352	2300	12	4.71	4	258	0.4	8	1.51	0.2	46	41	12	3829	5.73	1.20	11	23	1.51	703	10	0.12	7	0.09	4	34	0.06	161	54
N)	70353	200	0.4	4.86	2	281	0.4	8	0.95	0.2	41	28	16	1781	5.09	1.41	11	21	1.59	484	5	0.12	6	0.09	2	27	0.05	169	46
4	70354	800	Q.2	5.19	2	312	1.0	7	0.58	02	53	- 30	18	1934	4.97	1.42	19	29	1.91	338	7	0.12	9	0.09	50	27	0.06	153	39
12	70368	100	0.2	5.22	2	307	0.4	5	1.31	0.2	44	20	14	869	4.61	1.44	12	23	1.72	484	2	0.12	6	0.10	2	42	0.09	183	45
														17.30				÷ .											- 17
13	70369	160	Q.2	5.50	2	302	0.4	5	1.19	0.2	45	- 21	15	1170	5.35	1.52	12	: 23	1.79	428	4	0.13	6	0.11	2	43	0.08	188	40
14	70370	150	0.4	6.17	2	395	0.4	5	0.81	0.2	38	20	12	2111	5.28	2.18	12	16	1.39	268	. 3	0.12	6	0.09	2	35	0.12	177	29
13	70371	260	02	5.21	2	416	0.4	5	1.00	02	40	20	11	1451	4.58	1.56	11	18	1.72	414	3	0.09	6	0.08	2	51	023	150	41
¥0 • 7	70373	50	0.2	4.83	2	398	0.5	5	1.94	0.2	60	16	12	633	5.69	1.11	17	18	1.52	560	2	0.09	6	0,11	2	178	0.26	168	48
37	70374	140	0.2	5.47	2	442	0.4	5	1.13	0.2	46	28	12	1321	5.99	1.78	13	16	1.50	261	3	0.10	6	0.09	2	61	0.24	172	31
· e	70275	70	0.3	* 00	2	1.11	0.7		1.21			20													-				
-2 -1	70375	100	0.4	4.50	2	J41 774	0.0	2	1.25	02	45	25	13	1171	5.03	127	10	19	1.68	247	12	0.10	6	0.10	2	88	0.25	189	31
5	70370	40	0.4	4.51	4	249	0.0	¢	1.35	0.4	42	40	21	1200	5.10	1.00	10	24	1.57	3/0		0.09	7	0.10	3	98	0,19	161	43
.1	70378	50	0.4	4.50	4	2/40 \$76	0.5		1.39	0.2	44	10	14	674	5.32	0.50		. 15	1.44	200	2	0.09	4	01.0	. 2	107	0.19	142	30
ฉี	70379	20	04	50	14	595	0.5	0	2.14	0.2	49	'n	21	229	5.32	1.90	14	20	210	11/4	2	0.07	24	0.09	2	38	0.30	162	159
-		20	0.4	5.54	14	005	0.5	,	2.14	0.5	33	25	51	252	5.49	1.80	15	<i>a</i>	4.6	BH	5	0.08	24	0.11	1	41	0.03	107	212
5	70380	10	0.4	3.99	10	475	0.4	5	2.02	0.5	52	20	32	400	594	LOR	17	16	1.88	1092	4	0.00	70	20.0	5	78	034	163	712
8	70381	70	1.6	5.02	4	644	0.4	6	2.01	1.0	51	21	22	1961	5.84	1.62	13	16	1.48	1393	2	0.08	19	0.10	7	100	0.38	181	326
7	70382	200	4,0	5.70	10	729	0.4	6	2.13	22	53	16	10	3778	5.38	211	14	. 14	1.17	1490	20	0.07	6	0.10	ģ	96	0.34	140	480
-8	70383	130	3.2	5.84	12	670	0.6	6	2.52	1.4	55	19	14	2606	5.03	253	13	15	1.29	1608	11	0.06	9	0.09	8	35	0.22	135	252
·9	70384	90	£.6	5.94	5	626	0.6	9	2.53	0.8	55	16	10	1133	5.17	2.28	14	17	1.56	1698	4	0.07	6	0.09	. 6	72	0.26	155	309
														1.1															
0	70385	50	0.8	5.08	7	444	0.4	5	2.04	0.4	.54	18	12	781	6.43	1.56	14	17	1.51	1504	4	0.08	6	0.10	8	102	0.39	182	372
1	70386	20	0.4	5.32	6	230	0.9	5	5.16	0.4	72	33	40	483	7.8S	0.71	16	33	2.96	2330	3	0.0S	22	0.11	8	107	0.49	321	322
-2	70387	40	12	4.60	11	Z20	0.6	6	6.36	0.2	67	36	165	490	6.81	0.70	13	26	3.34	2148	2	0.05	83	0.10	3	73	0.25	233	
0	70368	/0	12	4.90	14	473	0.7	ŝ	5.18	03	65	29	97	869	6.11	1.43	13	Z	2.60	1836	7	0.06	42	0.09	7.	62	0.16	183	250
4	70389	110	10	5,10	10	208	0.7	2	4.02	0.2	58	26	40	996	6.10	1.63	13	Z	2.65	1657	4	0.06	25	0.10	10	45	0.12	172	215
.5	70390	140	2.4	6 21	16	646	0.8	7	354	0.2	62	22	41	1444	6.26	2 12	14	25	2 RA	1647	· .	0.07	24	0.10		14	A 11	101	776
6	70391	120	2.0	5.20	14	149	0.7	Á	6 76	02	70	3 6	158	1390	6 04	1 20	14	25	1 19	2457	2	0.05	$\tilde{\tau}$	0,10		5	0.08	171	240
7	70392	240	5.6	4.61	9	379	0.6	Š	4.53	0.8	63	29	88	4134	6.05	1.36	14	21	2.43	1769	4	0.06	46	0.09	Å.	46	015	160	250
8	70393	180	2.8	4.51	8	314	0.6	5	5.98	0.2	68	29	164	1585	4.92	1.40	12	20	2.39	1839	4	0.05	80	0.08	- ĝ	42	0.08	149	237
я	70395	170	2.8	4.79	14	411	0.6	5	4.87	0.2	65	17	35	1615	3.92	2.08	12	13	0.91	1596	7	0.07	19	0.08	3	35	0.14	125	155
										. je .				おとい				: '											
0	70396	520	4.8	5.76	14	415	0.5	5	1.74	0.2	49	17	18	2955	4,87	2.50	11	13	0.94	1019	.11	0.07	9	0.08	5	28	0.19	130	120
1	70397	370	3.Z	5.03	11	36Z	0.9	5	4.02	0.2	72	21	20	1787	423	2.20	18	16	0.67	1459	16	0.07	9	0.08	8	43	0.19	129	91
2	70398	330	3.6	6.04	10	510	0.6	7	3.03	0.2	60	17	15	2421	5.22	2.70	12	14	1.00	1211	6	0.07	10	0.10	7	35	0.17	155	132
3	70359	320	2.8	5.00	~	461	0.5	5	2.40	0.2	22	14	12	2437	5.45	213	12	15	1.14	964	4	0.07	7	0.09	3	39	0.24	154	187
4	/0400	0/0	3.4	4,50	9	403	0.4	10	2.01	0.2	33	18	17	4025	6.88	I.48	13	18	1.48	811	3	0.08	8	0.09	. 4	81	031	173	284
· t	70401	770	24	A 6.8	6	468	64	5	2 27 .	0.2	57	19	12	2000	7 1 5	1.64	12	14	• • • •	0.67		0.00	-	0.00		120	0.24	101	271
6	70402	290	24	5.73	š	ŝ	04	ŝ	1.88	0.2	51	10	10	2020	7.62	2.12	11	14	1 20	1064	4	0.06	6	0.09		60	0,34	104	226
-7	70403	520	2.8	4.08	2	ŝ	04	5	1.63	02	50	15	12	462	5 50	1 61	12	10	1.00	510		0.08	2	0.10	2	62	0.30	164	196
8	70404	340	1.6	4.35	2	1418	0.4	\$	1.11	02	41	14	11	7435	678	171	11	12	0.04	510	·:	0.09	5	0.00	2	75	0.00	175	201
Ŷ	70405	170	1.2	4.16	7	766	0.4	â	1.40	0.2	46	16	17	1827	681	1 24	12 5	16	1.35	718		0.00	Å	0.00	1	101	0.32	190	215
								5							5.51		1			. 10		3,03	v	v.u,		101	0,04	100	
Ð	70406	200	12	3.56	3	658	0.3	5	1.17	0.2	40	13	14	1567	6.41	1.19	10	13	1.06	572	1	0.08	5	0.09	2	72	0.30	170	204
1	70407	3000	4.8	3.59	2	432	0.6	5	1,50	0.7	51	23	20	5268	7.26	1.28	14	14	0.84	696	2	0.08	7	0.13	7	69	0.22	136	210
2	70408	2700	4 <i>B</i>	3.74	4	596	0.4	5	1.27	0.3	42	29	18 '	2002	7.84	1.36	11	10	0.68	540	. 1	0.08	7	0.10	7	78	0.18	136	199
3	70410	2900	6.0	4.00	7	716	0.4	6	1.05	0.2	40	30	16	7991	9.85	1.57	10	10	0.71	656	2	0.08	6	0.11	5	59	0.23	156	209
4	70411	1200	2.4	3.63	6	559	0.3	5	1.26	0.2	40	15	16 4	1440	8.22	1.27	10	11	0.80	702	2	0.09	4	0.11	3	62	0.27	155	186

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NORANDA DELTA-

T.T.	SAMPLE	Au	~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~	N	A	Ba	Be	Bi	0	¢₫	Cc	Co	Ċ	Cu	Fc	ĸ	La	Li	Mg	Mn	Mo	Na	Ni	P	Pb	Sr	Τί	v	Za \$303-0
<u>NO.</u>	No.	ppb	ppo	96	pp m	_ ppm	ppm	ppm	<u>%</u>	ррш	ррш	ppm	ppm	ppm	96	%	DOM	PDB	- 5	POm	00m	56	DOD	56	DD in	000		DOD	mm Po 1 d
85	70412	2000	6.8	4.25	24	874	0.4	5	1.40	0.5	49	27	15	6956	9.77	1.58	13	10	0.80	850	3	0.08	5	0.10	24	70	0.26	148	27%
56	70413	700	2.0	4.39	4	1133	0.4	5	1.31	0.2	47	15	12	3045	7.66	1.56	12	12	1.05	977	. 1	0.00		0 10		တ်	014	170	2000
57	70414	2200	3.2	3.65	2	772	0.4	5	1.26	17	47	14	15	Sene	0.55	1 20	11	11	0.04	7/1		0.09		0.10	Š		0.24	1/6	206
8	20415	650	0.8	102	2	600	0.6	Ĩ	2 21	0.2	47		10	1940	6.55	1.00	11	11	0.04	/0/	4	0.09	4	0.11		- 47	0.26	176	478
8	70416	200	14	400	2	1022	0.0	5	4-37	0.2	<u></u>	11	10	1848	3.44	1.69	15	10	0.74	865	1	0.08	4	0.10	2	- 68	0.17	122	137
~	/0410	200	1.0	4.05	4	1054	0.4	2	1.28	02	43	16	13	2583	7.29	1.70	11	10	0.78	637	·· 1	0.08	4	0.09	ં 3	45	0,29	161	135
0	70417	2300	2.4	3.45	2	921	0.4	5	1.01	0.2	38	12	17	5201	6.56	1.41	11	6	0.71	505	. · •	0.07	4	0.00	,	41	0.24	141	112
1	70418	570	2.4	5.34	6	1104	10	7	2.12	0.2	63	27	16	2042	70	2 20		16	0.04	202	. 1	0.07		0.07			0.20	141	
2	70419	190	12	6.96	3	1390	07	Ś	2 54	02	ñ	17	- 10	1051	7.54	2.30	17	10	0.90	/33	<u>.</u>	0.08	10	0.10	. 10	13	0.29	181	145
1	70420	310	1.6	5 20	2	770	0.7	5	2.34	0.2	60	1/		1221	7.04	321	14	14	0.56	829	1	0.08	9	0.11	•	62	0.34	194	116
í	70421	420	1.0	470	2	1/0	0.0		4-61	02	33	19	12	2/62	9.89	206	13	19	1.19	994	1	0.07	7	0.11	6	49	0 <u>22</u>	185	154
•	70421	4.30	1.0	4.70	4	831	0.6	10	2,47	0.2	65	16	13	2410	5.99	1.93	14	15	1.14	771	1	0.08	8	0.11	5.	- 56	0.19	161	131
3	70422	270	1.2	4.70	3	595	0.6	6	2.01	0.2	51	16	12	1765	7.78	1.71	13	20	158	200	t	0.09	۲	0 11	Å	77	0.16	145	159
5	70423	320	24	5.03	3	786	0.5	5	2.06	0.2	48	17	11	2013	6 22	2.22	12	14	1 14	045	-	0.07	1	0.11		- 20	0.10	103	108
7	70424	400	36	5 33	13	1085	0.5	ŝ	1 26	0.2	45	22	10	2024	405	2.42	17	14	1.14	743	2	0.07		0.09			0.13	145	115
	70425	110	0.4	3 42	2	1000	0.4	5	1.04	0.2	20	17	10	20.50	0.93	24/	2	13	0.92	623	2	0.08	8	0.09	1	29	0.13	145	115
1	70424	140	0.4	2.40	2	1212	0.4	2	1.04	0.2	39	1	12	752	6.12	1.41	11	12	0.92	467	1	0.09	5	0.08	2	- 30	0.16	156	112
	/0420	140	0.4	3.09	2	1312	0.7	2	1.06	0.2	45	18	18	1203	7.99	1.41	15	15	0.98	56 3	2	0 .10	6	0.10	2	ഒ	0.29	153	124
2	70427	340	1.6	4.22	2	988	20	5	1.05	0.2	40	16	12	2327	6.50	1.66	12	13	1.05	536	1	0.09	۵	0.09	2	58	030	184	155
3	70428	160	0.4	3.98	2	929	0.4	6	0.95	0.2	37	13	13	1155	6.21	1.58	12	17	1 00	497	;	0.00	č	0.00	2	~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~	0.20	104	126
1	70429	140	0.4	3.65	2	880	0.4	ŝ	1.04	0.2	28	15	11	1331	6 10	124	10	11	0.07	494	1	0.09	2	0.09	5		0.30	100	12.5
5	70430	280	04	3 47	ĩ	000	0.4	ž	1.00	0.2	20	17		2127	7.00	1.54	10	11	0.97	454	1	0.09	2	0.09	-	~~~~	0.29	181	110
	70421	140	0.4	2.70	2	505	0.4	2	1.00	0.2	30	17	13	2137	/.02	120	12	12	1.01	498	1	0.09	5	0.09	2	77	0.29	197	135
,	/0431	140	0.4	5-29	Z	849	03	2	1.04	0.2	36	14	13	1380	6.40	1.22	11	11	0.87	452	1	0.10	5	0.09	2	79	0.29	181	122
	70432	250	1.2	3.92	2	923	0.4	5	1.03	0.2	41	18	11	2516	6.56	1.75	13	ģ	0.67	414	· .,	0 10	4	0.00	2	•	0 22	109	120
	70433	260	1.6	4.70	4	603	0.4	5	2.06	0.2	55	25	11	2044	7.95	1 76	15	ú	0.87	200	- 1	0.00	-	0.00	· 2	152	020	100	140
	70436	180	0.4	4 17	2	618	04	š	1 20	11	44	15	12	1766	6.00	1.70	15		.~	/06	2	0.09	1	0.09		155	0.32	190	149
	70437	570	7.8	2 19	Ā	514	0.4	2	1.00	1.1	5	14	17	1300	3-29	1.40	11	24	1.05	534		0.09	0	0.09	4	29	0.16	155	360
	20439	940	10.0	3.10	2	274	0.4	ç	1.66	1./	24	10	17	3774	8.00	1.13	12	20	0.79	659	. 2	0.11	5	0.09	4	15	0.05	179	383
	/0438	800	102	3.09	0	312	1.1	0	2.06	4.3	62	23	31	4051	8.67	1.12	17	25	1.04	775	. 8	0.10	8	0.09	. 9	17	0.04	208	399
	70439	670	4.0	2.23	2	216	0.5	9	2.21	3.1	46	16	19	3238	7.42	0.65	11	21.	1.08	916	2	0.10	5	0.08	. 2	17	0.04	142	504
	70440	650	2.0	3.46	5	458	0.7	5	3.20	0.4	60	13	20	2004	5.96	1.64	12	0	1 00	1041		0 10	é	0.00	1	60	0.01	125	167
	70441	180	0.4	3.70	2	362	0.7	5	2.67	04	57	18	13	1351	5.62	1 40	12	10	1 22	766	1	0.10	4	0.07		20	0.07		
	70442	100	04	4 39	4	421	0.0	ŝ	1.69	0.7	so	10	20	000	8.45	1.47	10	19	1.00	755	. 0	0.10		0.09	2	45	0.05	100	192
	70443	300	2.0	4 10	7	114	0.9	2	3.00	0.5	74	10	20	606	5.05	1.68	13	10	1.63	1101	2	0.10	10	0.10	4	- 56	0.07	170	155
	70415	550	2.0	4.10	4	440	0.9	0	2.68	U,4	26	17	16	2350	6.83	1.78	12	14	1.20	815 ·	3	0.10	4	0.10	:3	39	0.09	156	191
	70444	400	2.4	3.02	6	356	0.7	5	2.26	0.9	51	13	17	2780	5,80	1.31	11	12	0.98	791	3	0.10	2	0.09	2	32	0.07	138	273
	70445	430	24	2.66	3	Z9 5	0.4	6	1.70	4.3	46	16	17	3675	6.14	0.74	11	24	1 27	774	3.	0.09	4	0.09		18	0.04	128	747
	70446	250	0.8	3.27	2	322	0.5	5	1.91	0.4	48	14	16	1844	5.69	0.02	12	- 24	1 20	704		0.00		0.10		10	0.04	140	205
	70447	680	72	343	7	144	0.5	6	232	14	ŝ	18	20	(C7)	5,00	1 14	11	20	1.40	714		0.09	2	0.10		10	0.04	142	305
	70449 6	<u> </u>	00	2.20	<u> </u>	200	0.0	10	100	0.5	54	10	20		J.46	1.10	11.	11	0.92	110	. و	0.09	5	0.08	12	22	0.04	121	369
	70446 3	X 00	8 ,8	رصد	2	290	0.8	12	2.80	0.8	60	18	26	2974	5.12	1.09	14	- 22	0.93	953 ::	3	0.09	7	0.07	86	27	0.03	101	272
	70449 5	630	6.4	4.03	3	315	0.6	5	3.81	1.9	57	19	24	2649	6.51	1.39	11	24	0.90	1125	2	0.10	7	0.08	21	38	002	135	477
	70450 S	840	3.6	3.54	7	265	0.5	5	3.30	0.4	53	17	15	7448	8.08	1 44	11	14	0.54	1077	ംപ്പ	0.00	Å	0.08		21	0.04	170	204
	70451 S	630	32	3 56	R	256	0.5	Ś	4 10	0.2	55	18	14	1574	714	1.45	10		0.00	11.42		0.00		0.00	<u>_</u>	21	0.04	140	
	704525	1200	64	3 50	Ă	280	ñÃ	ĸ	3.84	0.5	56	10	10	1121	6.36	1.45	10	19	V.49	1145	- 4	0.09	4	0.07	24	20	0.03	113	24
	70462.6	210	x 0	3.27	-	117	0.0	ç	1.54	0.0	30	19	19	116/	دفد	222	10	11	0.90	1355	- 4	0.10	9	0.07	- 34	37	0.03	114	159
	10403-5	810	0.0	2.70	1.	317	0.5	0	2.76	0.8	53	17	24	Z740	5.84	1.09	10 ្	្រា	0.76	958 _	3	0.08	7	0.07	40	28	0.02	98	236
	70454 S	630	32	3.14	2	317	0.5	5	2.95	0.4	51.	14	20	1946	5.78	1 26	ារាំំ	33	286	1105	•	0.08	7	n m .	12	20	0 00	109	101
	70455 S	1500	3.6	3.18	6	219	0.5	ő	3 14	0.2	52	18	22	7451	8 14	1 02	11	22	0.05	1140	1	0.07	~	0.00	13	20	0.03	126	
	70454 5	1200	28	2 72	2	154	0.4	6	2.14	62	44	10	24	2001	1 10	0.00	10	<u> </u>	0.95	1140		0.07		0.09	- 15	20			441
	704 49 6	1600	10	2.00	é	794	0.4	0	2.24	0.2		19	- 34 - 3	ا وجعه	1.10	0.80	10	<u></u> д (J./O)	1040	1	0.00	0	0.06	: 9	22	0.02	133	167
	704-06-3	1000	4.0	5.00	2	200	0.5	<u>د</u>	3.24	0.2	51	21	30	2402	4.81	1.34	9	10 (1.38	568	. 3	0.05	8	0.05	5	27	0.02	76 /	87
	/0459.5	2400	4.8	4.00		419	0.9	5	2.95	0.2	57	_27	_39	<u>7265</u>	4,52	1.75	14	15 (1.46	865	4	0.06	10	0.05	- 15	34	0.02	8t	78.

 $(\underline{U}_{ij}, \underline{v}_{ij}) \in \mathbb{R}^{n} \times \mathbb{R}^{$

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т. –	SAMPLE	٨₽	A¢.	A	A	Ba	Be	Bi	G	8	Ce	ŝ	<u>a</u>	- Ca	Fe	ĸ	La	Li	Mg	Ma	Mo	Na	NE	P	Pb	Sr	Ti	v	Za 1303-01	•
э.	No.	ррб	ppm	%6	ppm	ppm	ppm	ppm	96	ppm	ррш	ppm.	ррш	рраз	- %	%	ppm	ppm	- %	ppm	ppm.	%	рра	- %	ppta	ppm.	<u> </u>	ppa	ppitt Pg. 4 of	<u>*</u>
-2	70460 S	1700	3.6	4.83	2	563	0.7	5	3.37	0.2	59	23	25	1540	5.73	2.20	13	12	0.50	1031	4	0.07	9	0.08	12	37	0.04	84	. 74	
3	70461 S	8200	6.8	2.83	2	236	د.0	6	3.04	0.2	48	24	35	2757	9.90	1.05	11 :	16	0.71	1162	i 3	0.06	6	0.04	15	35	0.02	91	153	
4	70462	3400	4.0	1.52	2	117	0.3	5	1.17	0.2	29	20	54	4646	13.63	0.47	8	13	0.56	794	1	0.06	1	0.04	2	14	0.02	120	199	
5	70463	4100	6.0	0.82	2	81	0.3	5	0.67	0.2	14	18	73	6161	15.61	0.25	2 .	8	0.32	534	1	0.05	3	0.03	2	10	0,02	128	107	}
6	70464	80	0.8	1.71	2	229	0.3	5	1.12	0.2	26	10	60	1792	8.80	0.71	8 -	8	0.44	560	2	0.07	3	0.04	2	17	0.03	122	95	
										• :											1.00				1 - Jul				an taga t	
7	70465	2000	2.0	2.98	2	393	ک0	5	1.79	0.2	35	- 14	51	872	6.07	1.33	7	8	0.54	665	2	0.07	5	0.04	10	26	0.02	110	104	
8	70466	1,500	3.6	3.66	2	540	0.7	5	1.86	8.5	43	12	37	569	4.62	1.53	10	12	0.62	587	2	0.07	6	0.06	117	26	0.02	101	786	
9	70467	220	0.2	2.84	3	583	0.5	5	1.78	0.5	47	10	36	250	3.77	0.97	12	19	0.79	497	1	0.08	7	0.08	20	32	0.02	70	147	
J	70468	260	0.2	3.95	2	777	0.6	5	1.88	0.2	54	11	29	241	3.68	1.55	13	18	0.81	586	3	0.08	7	0.08	. 9	29	0.03	80	106	
1	70469	60	0,2	4.13	2	526	0.8	5	2.08	0.3	54	13	29	235	3.24	1.41	14	25	0.89	633	· . 4	0.09	8	0.08	.2	34	0.04	73	148	
																		2												
2	704 70	60	0.2	5.35	2	612	0.6	5	2.51	0.2	55	11	19	308	3.32	2.08	12	21	0.91	852	1.1	0.10	7	0.09	2	31	0.05	73	119	
3	70471	250	0.8	4.69	2	574	0.6	5	2.84	0.2	55	12	16	757	3.25	2.21	11	. 7	0.60	719	- 2	0.08	7	0.08	7	- 30	0.05	67	100	- i
\$	70472	110	0.2	4.87	2	429	0.8	9	5.26	0.2	64	15	53	214	4.66	1.82	11	25	1.89	1413	- 2	0.08	26	0.09	3	- 56	0.05	123	159	
5	70473	2400	6.0	4.31	2	418	0.7	9	4.22	03	57	19	104	7498	6.93	1.43	13	29	2.29	1434	2	0.06	35	0.10	2	- 52	0.04	139	198	j
6	70474	280	1.6	4.59	2	701	0.7	5	2.88	0.2	55	13	15	392	4.38	2.19	12	8	0.92	928	3	0.08	9	0.09	7	- 38	0.05	- 88	83	
														· · *				·												
7	70475	120	0.8	3.94	2	667	0.6	- 5	2.21	0.2	49	9	20	290	3.65	1.76	11	11	0.82	646	2	0.08	9	0.08	2	31	0.04	72	95	
3	70476	50	0.2	4.08	2	876	0.5	5	2.01	0.2	52	10	18	145	3.18	1.92	11	· 8	0.64	486	1	0.08	6	0.08	2	- 25	0.05	65	78	
1	70477	190	12	3.89	2	776	0.9	- 5	1.98	0.2	57	16	23	357	3.47	1.79	16	12	0.59	522	6	0.08	6	0.08	9	23	0.05	75	81	
2	70478	310	2_8	3.68	2	545	0.7	5	1.73	0.2	41	10	22	.771	5.37	1.75	11	7	0.55	482	14	0.08	4	0.07	3	25	0.06	91	78	
3	70479	140	1.2	4.02	5	667	0.6	5	1.61	0.2	42	13	27	219	6.05	1.81	11	8	0.46	359	2	0.09	2	0.09	2	22	0.05	79	68	
	70.500	1.10	••	1 01	•	~~~	~ .		2.54		12	~	40	min	6.56	0.00						~~~	27	A 00	· · .		A 14		139	
- i	70505	110	2.9	3.61	3	220	0.4	د م	1.74	0.4	65	14	00	2120	102	1.07	10		1 10	273		0.07	ు	0.09			0.10	120	179	
2	70505	200	20	4,44	- " - "	120	0.4	Ŷ	1.70	0.2	 S	19	10	2020	4.95	1.00	12	: <u>14</u> 10	1.17	610		0.00	с к	0.10	2	72	0.22	164	146	
9 7	70505	190	3.2	3.70	2	4.00	0.4	2	1.75	0.0	51	12	16	1007	4.15	1.00	11		0.77	500	1	0.07	7	0.09	· 1	00	0.26	129	174	
2	70,500	490	1.0	3.74	12	570	0.4	5	1.00	0.2	42	.79	20	1910	6 10	1.44	12	- 1	0.55	\$22		0.04	č	0.07		41	0.17	111	60	
>	70506	400	4.0	4.57	14		0.4	5	1_31		46		200	1010	0.17	131	14		0.00	555		0.00		0.07	·	-11	0.1)			
9	70509	300	24	3 90	7	409	0.4	5	2.50	0.2	62	12	22	2167	4.92	1.30	14	11	1.08	994		0.07	5	0.08	2	63	0.21	128	99	
á	70510	200	2.0	3.77	2	806	0.4	5	1.14	0.2	44	9	21	1180	2.99	1.76	11	៍។	0.60	379	9	0.06	4	0.08	2	23	0.12	128	42	
1	70511	120	1.6	2.97	2	628	03	ŝ	0.86	0.2	32	10	18	951	3.78	1.12	10	8	0.79	435	4	0.07	Ś	0.08	. 2	36	0.19	136	70	
2	70512	210	12	3.31	2	576	0.3	5	1.06	0.2	40	13	16	2005	4.35	1.23	12	9	0.82	435	4	0.07	7	0.08	2	69	023	139	73	
3	70513	190	12	3.73	2	560	03	5	1.23	0.2	44	12	14	1692	4.91	1.28	13	10	0.98	626	3	0.07	7	0.08	2	87	0.24	136	76	
	- +													4. ¹ . 4 ^{.1} 1. 4.4.4				11. L.						,					1977 - 1977 - 1979 - 1970 - 19	
4	70514	240	2.4	4.00	4	720	0.4	6	1.44	02	49	15	14	2465	4.90	1.30	15 -	_11	1.14	645	11.4	0.09	7	0.09	2	127	0.28	164	79	
5	70515	320	2.4	3.72	3	696	0.4	5	1.13	0.2	46	14	15	3185	4.59	1.40	12	9	0.98	446	4	0.09	7	0.08	2	91	0.26	150	76	

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55 NORANDA DELTA

NORANDA VANCOUVER LABORATORY Geochemical Analysis

		Project Materia	Name l:	& No).:	KLIY 33 RO	الل - ما/ت	148 ••• 5/		(د		Geol.: Sheet:	:G.G :1 of	1				Date Date	receiv compl	ed: cted:	MAR. MAR.	04 15	1	LAB	CODE		9303	-01	1
		Remark	3:	-	Semple	e scree	9 báo	-35 M	ESH (O	.S @ @))																		
				8	Ongano	ic, ∆ H	lumm,	S Sulfi	de					Au - 1	0.0 g sar	ople dige	ested w	ith ages	~regia	and det	ermine	d by A.	4. (D.L.	5 PPB)				
		1CP - 0.2	gaampi	le digest	ted with	3 m l H	сю₄∕н	NO3 (4	i:1) at 2	03 °C fe	x 4 hour	n dilute	ed to 10) osl wid	b water.	Leeman	PS3000	ICP de	termine	d clem	entai co	ntenu.							
		N.B. The	najoto:	side ele	ments 8	nd Ba,	Be, Ce,	Lz, Li, 1	Ga ate	carely d	lissolved	i co m pl	etely (r	ovi geo.	logical m	eterials	with thi	a acid d	issolutie	on meth	od.								
ſ.T.	SAMPLE	Aa	Ag	A1	As	Ba	Be	Bi	- Ca	G	Ce	Co	Q	Cu	Fe	K	La	Li	Mg	Mn	Mo	Na	Ni	P	Pb	Sr	Ti	v	Zo
No.	No.	ppb	ppm	<u>%</u>	_ <u></u>	ppm	ppm	ppm	<u>%</u>	ppm	ppm	PP ®	ppm	_ppm	<u>%</u>	<u>%</u>	<u>bb</u> m	ppm	. %	ppm	ppm	<u>%</u>	ppm	- 96	ppm	ppm	<u>%</u>	ppm	ppm
	-2. 70311	130	1.4	4.11	2	321	0.4	5	1.45	0.5	42_,	, 16	18	708	4.39	1.08	11	17	1.38	335	3	0.10	6	0.07	. 8	42	0.06	143	55
\$	70312	100	0.4	4.22	2	- 307	0.4	5	151	0.2	47	25	21	1016	4.82	0.95	- 14	17	1.62	327	. 5	0.10	7	0.07	4	- 88	0.12	160	. 53
5	70313	70	0.2	4.14	2	320	0.3	5	1.56	0.2	46	27	14	715	4.92	0.93	13	15	1.52	363	7	0.09	7	0.07	2	- 99	0.14	156	- 52
i	70314	110	0.2	3.99	2	269	0.3	5	1.69	0.2	50	20	16	1285	5.27	0.68	14	16	1.55	310	4	0.10	8	0.08	2	122	0.11	171	51
7	70315	200	0.2	3.98	2	324	0.3	5	1.36	0.2	42	21	16	1156	5.77	0.87	14	15	1.42	296	4	0.10	7	0.07	2	88	0.09	161	5]
						•		_			_											-		-					
3	70316	140	04	1.79	2	294	03	5	1.36	03	46	26	14	1291	5.48	0.80	13	15	1.40	213	9	0.09	7	0.07	2	95	0.06	169	44
	70317	60	0.7	1 90	2	476	0.1	š	1 25	02	41	20	11	801	174	1 12	12	15	1 44	164	- ý	011	7	0.07	2	41	0.07	162	34
.0	70317	40	0.2	1.07	-	704	0.0	2	1 75	0.2	42	10	17	765	x 11	0.97	11	16	1 64	264	k	0.10	2	0.06	2	01	0.00	110	
10	70310	100	0.2	5.00	<u></u>	271	0.0	2	1.00	0.2	74	20		1135	4.31	1 44	10	10	1 80	214		0.10		0.00		71	0.12	120	
11	70319	100	0.4	3.03		3/1	0.4	2	1.09	0.2	29	10		1120	4.51	1.40	12	10	1.00	224		0.10	<u></u>	0.07		100	0.12	140	
12	10320	110	0.4	5.10	4	337	0.4	2	1.72	0.2	41	18	U.	1176	4.04	1.31	فل	12	1.03	343	2	0.10		0.09	4	106	0.12	140	<u>a</u>
	20231	160			2	000	0.2		۸m		20			016	4.74	1 71	•		: 44	<u>эт</u>		0.10		0.07		46	0.12	163	1
13	70321	130	0.8	4.00	2	232	0.3	2	0.93	. 0.4	.50	14	11	90	4.24	101		14	1.00	610	د .	0.10		0.07	. 4	4)	0.05	105	
14	70322	90	0.4	0.20	4	394	0.5	Ş	4.13	0.2	50	19	14	232	2.24	1.65	13	.17	1.54	340		0.10		0.09	-	~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~	0.23	205	- 04
15	70323	100	0.4	5.41	2	382	0.4	5	2.20	0.2	51	21	8	1301	5.76	1.51	13	- 17	204	001	11	0.09	0	0.10	2	90	0.31	201	- 13
16	70324	80	0.2	4.42	2	349	0.4	5	1.75	0.2	44	30	11	1016	3.64	1.41	11	14	137	489	34	0.10	8	0.09	4	48	0.12	149	
17	70325	180	0.8	5.07	2	317	0.4	5	1.20	0.2	37	33	10	2062	5.27	1.55	11	18	1.65	371	8	0.11	8	0.08	4	33	0.09	154	46
									_					<u>,</u>				. <u>.</u>			100		_						Sec
18	70326	130	0.2	5.38	2	271	0.3	5	1.05	0.2	39	26	12	1574	5.91	1.39	13	20	2.19	333	5	0.11	9	0.08	2	52	0.23	197	45
19	70327	210	0.4	5.46	2	297	0.4	5	1.29	· 0.2	44	33	11	2442	6.26	1.51	13	17	2.01	407	- K. (7)	0.09	7	0.08	୍ 4	68	0.24	160	49
20	70328	130	0.8	5.17	3	273	0.4	5	0.93	0.2	38	25	12	1910	4.35	1.56	12	÷ 17	2.19	314	- 21	0.10	8	0.08	2	31	0.15	179	. 45
21	70329	100	0.8	5.54	4	346	0.6	5	1.43	0.2	36	30	11	1623	4.81	1.86	13	17	1.99	446	- 5	0.09	8	0.03	2	27	0.15	175	ି 62
22	70330	4000	8.8	4.85	2	816	0.7	5	1.54	0.5	34	38	13	1767	6.10	1.73	11	14	1.84	475	21	0.10	9	0.09	2	- 20	0.13	200	76
																		d sette			•								
23	70331	1700	6.4	5.55	4	459	0.7	5	2.20	3.3	45	39	13	1769	6.42	2.06	13	15	1.70	458	25	0.10	9	0.08	129	- 36	0.10	189	101
24	70332	240	0.4	5.95	2	421	0.4	5	1.44	0.2	42	32	10	1891	7.28	1.68	16	20	2.14	457	10	0,10	10	0.09	2	- 56	0.13	195	65
25	70133	420	0.8	4 70	3	423	0.4	ŝ	1.19	0.2	45	27	15	1673	6.99	1.36	15	17	1.73	310	8	0.11	9	0.08	6	43	0.11	182	5
26	70324	260	0.4	2.04	2	404	0.4	ŝ	1 27	0.2	19	32	14	1270	6 77	115	14	At .	1 53	295	1 Q	0.00	8	0.07	4	25	0.06	149	44
27	20225	240	17	105	5	149	0.4	5	1 29	0.2	27	20	12	1550	2 20	1 12	12	17	151	312	6	0.10	Ă	0.02		26	0.04	150	~
41	70335	500	1.4	3.75	4	540	0.4	5	1-40		, נ	27	14		1.56	1.12	14		1.51	212		0.10	U	0.00		-	0.00	1.00	
	20226		20	200		200	0.6	ç	1 41		42	10	13	1/26	\$ 61	1 29	12		1.20	242	11 - E.	0.11	e	0.00		20	0.06	121	4
28	/03.96	230	20	3.90	2	323	0.5	2	1.51	0.2	4/	19	10	1435	2.01	1.30	~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~	9 1 9	1.20	- 397	್ಷ	0.11	2	0.09		100	0.00	161	. OU
29	70337	60	0.2	4.20	4	191	0.8	2	3.14	0.2	81	12	10	ഹാ	4.20	1.90	23	्य	1.00	9/28	- -	0.09	0	0.14	·	197	0.15	106	
30	70338	170	0.4	3.75	2	774	0.6	5	1.52	02	62	21	16	666	4.81	1.24	18	12	0.98	670	- Z	0.10	8	0.11	4	149	0.15	117	<u></u>
31	70339	210	0.4	4.15	2	.575	0.4	5	0.91	02	33	28	11	806	5.21	1.29	11	13	1.35	483	5	0.10	6	0.09	2	86	0.16	115	76
32	70340	140	0.8	5.13	2	516	0.4	5	1.22	0.2	37	23	7	992	6.16	1.49	10	17	1.61	441	<u>,</u> 7	0.09	8	0.08	<u>्</u> 5	106	0.11	163	66
																					è de la c							:	a jej
33	70341	270	0.8	5.70	2	455	0.4	5	1.24	0.2	41	28	9	1827	6.59	1.65	13	19	1.76	388	ં 3	0.10	10	0.09	3	61	0.08	175	- 55
34	70342	180	0.2	4.87	2	409	0.4	5	0.91	0.2	34	22	8	1210	4.95	1.49	10	17	1.47	255	3	0.11	6	0.07	2	36	0.07	159	41
35	70343	110	0.4	5.03	2	422	0.4	5	1.03	0.2	38	28	8	1233	5.63	1.41	11	19	1.72	393	4	0.11	9	0.09	2	34	0.07	171	- 54
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100 G.P /B. C.

NORANDA VANCOUVER LABORATORY Geochemical Analysis

Project Name &	No.:	KUYUL	- 148	Geol.: G.G		Date received:	FEB. 26
Material:	SMALL	55 RC	concerned in the second	Short: 1 of	2	Date completed:	MAR. 02
Remarks:	* Sampi	e screened	@ -35 MESH (0.5 mm)			_	

Organic, Δ Humus, S Sulfide Au ~ 10.0 g sample digented with aqua-regla and determined by AA. (D.L. 5 PPB)

ICP = 0.2 g sample digested with 3 ml HCIO4/HNO3 (4:1) at 203 °C for 4 bours diluted to 10 ml with water. Leeman PS3000 ICP determined elemental contenta.

N.B. The major valde elements and Ba. Be, Ce, La, Li, Ga are carely dissolved completely from geological materials with this acid dissolution method.

SAME	1.8	Au	Âr	AI	As	Ba	Be	Bi	9	Cd	Cc	Co	- G	Cu	Fe	ĸ	La	11	Me	Mo	Mo	Na	Ni	P	Pb	Sr	Ti	v	 Zn
	No.	ppb	DDm	%	pom	ppm	DOB	DOm	56	ppm	ppm	ppm	DOM	ppm	96	%	PP D	DOM	56	2000	ppm	96	POD	76	ppm	ppm	%	ppm	Dom
KL-93-1	4 - 6	5	0.2	5.35	6	318	0.5	5	3.22	0.2	63	-24	62	122	6.17	0.65	12	24	2.33	1198	2	0.15	39	0.09	2	167	0.33	219	86
	8 - 10	5	0.2	6.47	4	801	0.6	6	2.64	0.2	63	12	13	145	4.08	2.05	15	12	1.32	1254	2	0.09	9	0.10	2	160	0.33	147	93
	12 - 14	5	0.2	4.50	2	1291	1.0	5	3.44	0.2	84	12	27	50	4.32	1.50	26	12	1.50	938	1	0.09	22	0.13	7	358	0.37	141	76
	16 - 18	5	0.2	4.21	2	869	0.9	5	3.74	0.2	80	14	31	159	4.30	0.93	22	12	1.51	963	1	0.10	21	0.13	6	417	0,27	137	78
	20 - 22	190	0.2	7.42	2	745	0.6	6	2.20	0.2	64	16	11	1899	5.11	2.26	14	ß	1.50	531	. 1	0.11	5	0.11	3	159	0.26	137	51
2 · · ·	24 - 26	50	0.2	5.39	2	443	0.4	5	1.58	0.2	64	19	15	860	4.81	1.54	17	12	1.23	301	3	0.11	3	0.11	2	110	0.12	104	- 35
2 ≤ 1	28 - 30	70	0.2	5.41	2	402	0.4	7	1.47	0.2	60	17	14	900	4.85	1.61	16	12	1.14	315	3	0.11	4	0.11	2	94	0.10	105	31
	32 - 34	100	0.2	4.72	2	348	0,4	5	1.78	02	61	15	14	736	5.34	1.15	15	12	1.04	353	- 1	0.09	5	0.11	2	116	0.06	104	- 38
1, the 1	· 36 - 38	150	02	6.49	2	292	0.5	9	2.69	· 0.3	61	28	12	2056	5.66	1.25	15	21	2.21	801	-5	0.12	9	0.08	4	137	0.16	186	71
· · · ·	40 - 42	130	0.2	6.63	2	381	0.5	7	2.82	0.4	60	27	22	1415	5.12	1.66	13	19	2.39	627	3	0.13	18	0.08	4	75	0.08	175	62
- 2	44 - 46	60	0.2	6.48	2	312	0.4	8	4.03	0.2	65	25	10	1162	5.77	0.93	15	18	2.12	635	3	0.12	10	0.08	3	196	0.21	199	55
-:	48 - 50	240	0.2	6.28	2	268	0.4	6	2.93	0.4	61	21	18	1758	5.95	0.87	14	19	2.35	475	_: 2	0.11	14	0.09	: .: : 3 .	190	0.20	207	52
	52 - 54	120	02	7.46	3	590	0.5	9	2.66	0.6	61	31	13	1535	5.96	1.74	15	16	2.21	653	5	0.12	13	0.09	2	157	0.26	189	56
- · · ·	56 - 58	60	0.2	6.79	2	330	0.5	5	1.73	0.2	52	33	26	927	7.19	1.28	12 .	- 22	2.93	675	7	0.10	18	0.06	2	80	0.13	209	65
1	60 - 62	30	0.2	6.13	2	348	0.4	9	1.69	0.3	54	30	14	911	5.42	1.40	14	17	2.19	515	14	0.12	11	0.09	999-9 3 0 2010-008	107	0.13	182	48
	64 - 66	40	0.2	5.41	3	381	0.4	5	1.81	0.4	57	34	17	701	5.44	1.25	14	14	1.66	540	7	0.11	12	0.08	2	130	0.14	175	50
	68 - 70	60	0.2	5.64	2	298	0.4	8	1.45	0.3	54	48	14	1583	6.10	1.18	13	15	2.06	S02	19	0.11	12	0.08	2	111	0.18	268	- 50
5	72 - 74	120	0.2	5.89	2	326	0.4	5	2.27	0.4	58	32	16	1734	4.75	1.42	14	14	1.75	520	8	0.15	11	0.07	÷. 3.	123	0.18	179	49
	76 - 78	80	0.2	4.14	2	279	03	5	1.58	0.2	52	13	19	773	3.21	1.02	12 -	11	1.14	373	3	0.11	8	0.06	. 2	107	0.12	127	34
97.	80 - 82	120	0.2	5.08	2	234	0.4	5	3.00 (0.2	65	22	13	1848	4.95	0.63	14	14	1.61	451	6	0,10	6	0.09	2	221	0.26	181	49
" KL-93-1	84 - 86	300	0.2	3.58	3	337	0.4	5	2.47	02	57	21	20	1393	5.49	0.93	13	12	1.49	488	217	0.10	4	0.07	2	58	0.15	154	- 58
·*KL-93-2	6 - 8	150	0.2	5.63	2	661	0.6	5	1.40	0.2	- 58	16	23	1281	4.69	1.65	14	19	1.75	701	9	0.12	15	0.11	14	166	0.33	187	73
A D	10 - 12	230	0.2	5.84	2	304	0.4	5	0.88	03	48	16	9	3579	5.92	1.51	12	- 24	2.04	369	2	0.10	5	0.09	2	66	0.30	168	- 44
-12.8	14 - 16	300	1.4	6.28	2	386	0.6	8	1.10	0.4	53	17	9	4780	5.15	1.97	13	20	157	519	4	0.10	5	0.10	5	92	0.19	164	- 52
<2 C	18 - 20	180	0.2	4.86	2	329.	0.4	7	1.26	02	54	18	13	2774	4.64	1.23	14	18	1.48	286	2	0.10	6	0.11	4	128	0.20	156	38
			÷.			·			:				:				:												14
	22 - 24	670	0.2	4.93	4	338	0.4	5	1.93	02	69	13	15	3753	5,70	1.09	18	18	1.42	541	2	0.09	5	0.10	2	193	0.22	175	52
:7	26 - 28	2.50	0.4	4.32	3	378	0.4	8	1.18	0.3	55	14	16	3044	6.02	1.13	13	16	1.32	364	4	0.08	3	0.11	2	126	0.19	172	49
10	30 - 32	160	0.4	4.46	8	493	0.6	11	2.35	0.8	78	14	15	1848	5.24	1.27	20 :	17	1.32	729	18	0.10	6	0.13	8	146	0.11	154	74
5 2 -	32 - 34	220	0.2	4.71	2	494	د ہ	5	1.91	0.3	58	19	16	956	4.82	1.26	15	18	1.42	557	8	0.11	7	0.09	29	93	0.09	ហ	70
1 X	34 - 36	110	0.2	4.21	2	328	0.4	5	1.40	1.2	49	17	15	766	4,60	1.06	12	17	1.40	342	7	0.11	5	0.07	6	45	0.07	147	- 56
:2	36 - 18	80	02	4 27	2	109	0.4	5	1.55	02	54	25	15	1068	4.77	0.93	13	117	1.67	317	÷ 7	0.09	6	0.07	4	89	0.11	158	42
13	18 - 40	40	02	4.39	2	351	0.3	5	1.51	02	52	34	17	667	4.95	1.06	14	15	1.52	358	8	0.10	7	0.07	4	87	0.13	156	៍រាំ
	40 - 47	110	0.2	4.26	5	301	กัจ		1.66	0.2	57	28	16:	1468	5.33	0.77	14	17	1.62	330	7	0 11	÷	0.08	8 (3)	123	0.12	168	ं दर्द
5	47 - 44	20	0.7	4 08	2	70	ñ.	š	1 28	0.2	52	277	18	1012	5.91	0.88	11	16	1.51	277	6	0.17	Å	0.07	7	84	0.00	165	47
· KL-93-2	44 - 46	80	0.2	4.23	2	327	03	5	1.45	0.2	55	29	16	1286	6.00	0.89	14	16	1.54	234	ġ	0.12	š	0.08	2	101	0.09	183	43

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NORANDA DELTA

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E. .. Frencherry

Г.Т.	SAMPLE	Aш	Ag	A	_Λs	Ba	Be	Bi	- Ca	Cd	Ce	Co	C C T	Ct	Fc	K	La	Li	Mg	Min	Mo	Na	Ni	P	Ръ	Sr	Т	- V	Za 9903-00
No.	No.	ppb	pp m.	- %	ppm	ppm	ppm	ррш	%	ррд	ppm	рра	ppm	ppm	<u>%</u>		ppm	ppn	<u>%</u>	ppm	ppm	- 96	ppm	<u> </u>	ppm	ppm	<u>%</u>	ppm	ppm Pg. 2 of 2
- 38'7 K	L-93-2 46-48	60	0.2	4.63	2	491	0.4	5	1.54	0.2	51	26	18	1167	4.62	1.27	13		1 <i>5</i> 7	192	: :12	0.14	6	0.08	2	61	0.09	186	37
79 15	48 - 50	30	0.2	4.26	2	340	0.3	6	1.43	0.2	50	24	18	897	4.49	0.92	13	16	1.60	270	10	0.12	6	0.08	2	94	0.09	1.50	45
10 17	50 - 52	60	0.2	5.10	2	380	0.4	5	1.29	0.2	45	19	19	935	3.99	1.51	11	15	1.58	310	1	0.11	6	0.09	4	60	0.11	132	16
11 24	52 - 54	100	0.2	5.21	2	352	0.4	5	1.51	0.2	52	17	16	885	4.72	1.34	15	17	1.62	335	4	0.11	8	0.09	88 X	97	0.13	155	59
(2.1	54 - 56	150	0.2	5.12	2	313	0.4	5	1.06	0.4	47	17	14	945	4.79	1.48	11	16	1.76	292	ंड	0.12	6	0.07	<u> </u>	48	0.15	177	44
	• • • •				-			-						-12				지겠							문화				
13 2 /	56 - 58	70	02	6 39	7	731	0.5	5	2.81	0.4	62	24	19	7733	5.98	1.58	14	20	2.11	658	3	0.09	14	0.09	3	103	0.29	233	62
14	58 - 60	- en	02	5 45		7 90	04	š	2.00		62	26	- 10	1357	5.85	1.55	14	े 1 7	195	622	- q	0.11	8	0.10	5	83	0.30	208	74
15 - 2	60 - 62	ŝ	07	1 22	7	212	0.4	ž	1.25	0.2	<u>.</u>	31	16	097	\$ 20	1 28	13	21	1 54	481	20	0 10	7	0.09	- K	51	013	149	n
16	62 64	170	0.0	5 05	,	207	0.4	5	1.75	0.5	50	27	12	7779	A 00	1 51	14	19	1 64	354	10	0 17	ģ	0.04	6	31	0.09	155	
17 - 2	64	120	0.2	5.05	2	207	0.4	2	1.00	0.0	12	24		1200	4.57	1 20	12	20	2.04	225	4	0.12	, ,	0.00		49	01.0	198	
,,	04 - 00	120	<u>~</u>	5.40	5	ω,	05	2	1.05	0.4	40	24	Б	1.550	0.01	1.04	15		203	صفد		0.12	0	0.00	- -		0.17	1.00	
	(4 - 68	240	02	< n<	5	and	01	٤.	1 17	67	57	40	12	2508	5.64	1 41	14	16	1 70	357	10	0.09	g	0.06		60	0.20	141	20
-0 13 / A	00 - 08 (8 - 70	190	02	5.00	5	210	0.0	, e	1.17	0.2	20	- 2 2	13	10/2	4 4 2	1.0	12	16	2.01	200	10	0.07	0 0	0.007	- T	า๊ง	016	160	29
	08 - 70	100	0.4	3.43	-	200	0.4	2	1 20	0.2	40	20	10	1761	4.41	1.00	11	10	201	200		0.11	,	0.07	- 7	- 26	0.10	140	
14 7.5	70 - 72	200	10.4	5.06	4	200	0.4	2	1.17	0.3		21	12	1/01	4,43	1.04		20	2.02	377		0.11	, ,	0.07	- -	20	0.15	190	47
1.5	12 - 14	2800	144	4.99	4	500	V .5	2	1.44	0.5	00	40	14	1000	0.34	1.00	12	10	208	332		0.12	10	0.06	· · · · · · · · · · · · · · · · · · ·	24	0.17	100	100
-4	74 - 76	1200	0.0	3.32	2	.934	0.6	2	1.97	4.4	22	16	12	1003	2,44	1.94	12	12	1.79	401	- 44	0.11	10	0.07	93	۹	U.11	191	100
					-		~ 4					46	10				• •	•••		400		•••		0.00			0.12	176	
20 C	76 - 78	250	0.2	3.31	2	396	0.4	2	1.30	0.2	24	- 20	10	144.5	0.12	1.38	14	18	1.92	407	12	0.11	y	0.05	4	49	0.12	1/5	
×0	78 - 80	270	02	3.68	2	364	03	2	1.08	0.2	49	- 24	15	1568	6.08	1.07	12	13	1.31	220	8	0.12	8	0.06	4	31	0.09	134	59
11	80 - 82	230	0.4	4.17	2	437	0.4	5	1.21	0.2	53	- 30	15	1237	5.79	1.23	14	10	1.35	238	. 11	0.12	2	0.07		25	0.07	121	9.3
8	82 - 84	310	0.2	4.04	2	300	0.4	5	1.13	0.2	53	25	12	1447	7.66	1.09	12	17	1.56	321	. 4	0.10	7	0.08	. 2	24	0.06	159	· 26
9 KI	-93-2 84-86	170	0.8	3.90	3	385	0.5	5	1.50	0.3	54	20	14	1110	4.99	1.27	13	- 13	1.26	367	6	0.11	6	0.09	- 5	32	0.06	117	58

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APPENDIX V

STATEMENT OF COSTS

NORANDA EXPLORATION COMPANY, LIMITED STATEMENT OF COSTS

PROJECT: KLIYUL DATE: MARCH, 1993 TYPE OF REPORT: DRILLING a) Wages: No. of Mandays : 77 Rate per Manday: \$219.11 Dates From : February 1 - March 9, 1993 Total Wages : 77 x \$219.11 \$ 16,871.47 b) Food & Accommodations: No. of Mandays : 77 Rate per Manday: \$53.43 Dates From : February 1 - March 9, 1993 Total Costs : 77 x \$53.43 \$ 4,114.11 C) Transportation: No. of Mandays : 77 Rate per Manday: \$71.01 Dates From : February 1 - March 9, 1993 Total Costs : 77 x \$71.01 \$ 5,467.77 d) Instrument Rental: Type of Instrument: No. of Mandays : Rate per Manday: Dates From : Total Costs : Type of Instrument: No. of Mandays : Rate per Manday: Dates From : Total Costs :

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e)	Analysis: 254 samples x \$13.80 (See attached schedule)	\$ 3	3,505.20
f)	Cost of preparation of Report: Author : 2 mandays x \$247/manday Drafting: 1 manday x \$247/manday Typing : 1 manday x \$120/manday	\$ \$ \$	494.00 247.00 120.00
g)	Other:		
	Contractor:		
<u>Paci</u>	fic Western Helicopters Ltd.:		
	13.9 hrs x \$750/hr including fuel and oil	\$ 1 0	0,425.00
<u>Midn</u>	ight Sun Drilling Ltd.:		
	560 meters of reverse circulation drilling at \$112.21/meter	\$6	2,837.29
<u>Payc</u>	o Contracting Ltd.:		
	65 hours of cat time at \$110/hour and 18 hours of Lowbed hauling at \$100/hour.	\$ E	8,950.00
<u>Tena</u>	<u>ki Enterprises Ltd.</u> :		
	56 hours of cat time at \$135/hour	\$ 7	,560.00
<u>Soni</u>	c Concrete & Aggregate Ltd.:		
	32 hours of Lowbed hauling at \$100/hour	\$ 3	3,200.00
	TOTAL COST	\$12	3,791.84
h)	Unit Costs for Drilling No. of Mandays: 77 No. of Units : 560 meters Unit Costs : \$221.06/meter Total Cost : 560 meters x \$221.06	\$12	3,791.84

NORANDA EXPLORATION COMPANY, LIMITED (CORDILLERA DIVISION)

DETAILS OF ANALYSES COSTS

PROJECT: KLIYUL

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ELEMENT	NO.	OF	DETERMINATIONS	Cost	PER	DETERMINATION	TOTAL CO	osts
ICP (30 Elemen + Geochem Au	nt) 1		254		#13.	.80/sample	\$3,505	5.20

APPENDIX VI

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STATEMENT OF QUALIFICATIONS

I, D. Graham Gill of the City of Vancouver, Province of British Columbia, hereby certify that:

I am a geologist residing at 5442 - 7th Avenue, Delta, B.C.

I have graduated from the University of British Columbia in 1983 with a BSc in geology.

I have worked in mineral exploration since 1979.

I have been a temporary employee with Noranda Exploration Company, Limited since May, 1979 and a permanent employee since November, 1987.

O.M. Hell

D. Graham Gill



ple No.	1	ANDESITIC TUFFS (FLOWS (Propylitized)
Trail	2	FERROCRETE
us	3	QTZ - SERICITE - CLAY - PYRITE ALTN.
ee	4	MAINLY SERICITE ALTN.
ression	5	DIORITE, QTZ DIORITE GABBRO
anda Soil Line m Sample Interval)	6	LIMESTONE
ff / Bluff	7	BLACK, PYRITIC SHALES, ARGILLITES LOCALLY CALCAREOUS
	8	CALCAREOUS ANDESITE TUFF / AGGLOMERATE
	9	MODERATE - INTENSE PYRITE/SILICA ALTERATION
	10	LISTWANITE



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