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DIAMOND DRILLING REPORT  
ON THE  
**KNIPPLE LAKE PROPERTY**  
KL 1 to 4, Treaty 12 Claims

Skeena Mining Division  
N.T.S. 104 A/05 W

NORANDA EXPLORATION COMPANY, LIMITED  
(no personal liability)

**GEOLOGICAL BRANCH  
ASSESSMENT REPORT**

**22,074**

REPORT BY: MIKE SAVELL  
FRASER STEWART

OCTOBER, 1991

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## 1.0 SUMMARY

The Knipple Lake Property comprises the KL-1, KL-2, KL-3, KL-4, and TREATY-12 mineral claims in the Skeena Mining Division, approximately 60 kilometres north of Stewart, B.C. The KL claims were staked by Noranda in 1988, and the TREATY-12 claim by Ross Resources in 1989. The TREATY-12 claim was optioned by Noranda in 1990. A detailed mapping program and approximately 9 km of induced polarization was completed in 1991 that covered the most likely source areas of the mineralized boulders of quartz-calcite-sulfide vein and breccia material from which assays up to 17.9 gm/t Au, 17.1 gm/t Ag, 0.14% Cu, 3.03% Pb, and 9.02% Zn were obtained.

The property is underlain by volcanic rocks of the Jurassic Hazelton Group. Two packages of volcanics are present, an extensive sequence of interbedded andesitic flows, tuffs and agglomerates (Betty Creek Formation) and a less dominant unit of variably silicified and quartz veined felsic tuffs and flows (Mt. Dilworth Formation). Alteration is not extensive but includes silicification of felsic rocks, chloritization of andesitic rocks, and quartz-sericite-pyrite alteration of felsic porphyritic intrusive rocks found immediately southeast of the KL-3 claim.

Quartz-calcite vein and breccia material containing variable amounts of coarse sphalerite, galena, chalcopyrite and pyrite has been found in subrounded to subangular boulders up to 100 centimetres in diameter scattered over a 100 by 500 metre area at the northwest corner of the KL-3 claim. Economic gold values have been detected in a significant number of these boulders. The source has yet to be located, but is believed to be upslope and/or up ice within the grid surveyed area.

A diamond drilling program consisting of 346.6 metres in six holes tested a variety of geophysical and geochemical targets located upslope and up ice from the mineralized boulder field. The program was hampered by poor drilling conditions. Three of the weak to moderate chargeability anomalies, a geochemical anomaly, and a VLF anomaly were tested. No significant mineralization was detected.

The source of the mineralized boulder field was not located. No further work is recommended at this time and the Treaty 12 claim should be returned to the vendor.

## 2.0 INTRODUCTION

The Knipple Lake Property comprises the KL-1, KL-2, KL-3, KL-4 and the Treaty-12 mineral claims in the Skeena Mining Division in the Bowser River area, north of Stewart, B.C. The KL 1-3 claims were staked by Noranda Exploration in 1988 to secure attractive gossans believed to be associated with pyritic felsic pyroclastics. The KL-4 claim was staked in 1991 to secure a small gap to the west of the KL-1 claim. The Treaty-12 claim was staked for Ross Resources in 1989 and was subsequently optioned to Noranda Exploration in 1990. This report describes the geological, geochemical, geophysical and diamond drilling programs undertaken by Noranda in August and September of 1991 to evaluate the economic potential of the claims. Diamond drilling which was contracted to J.T. Thomas Diamond Drilling Ltd. of Smithers, B.C.

### 2.1 LOCATION AND ACCESS

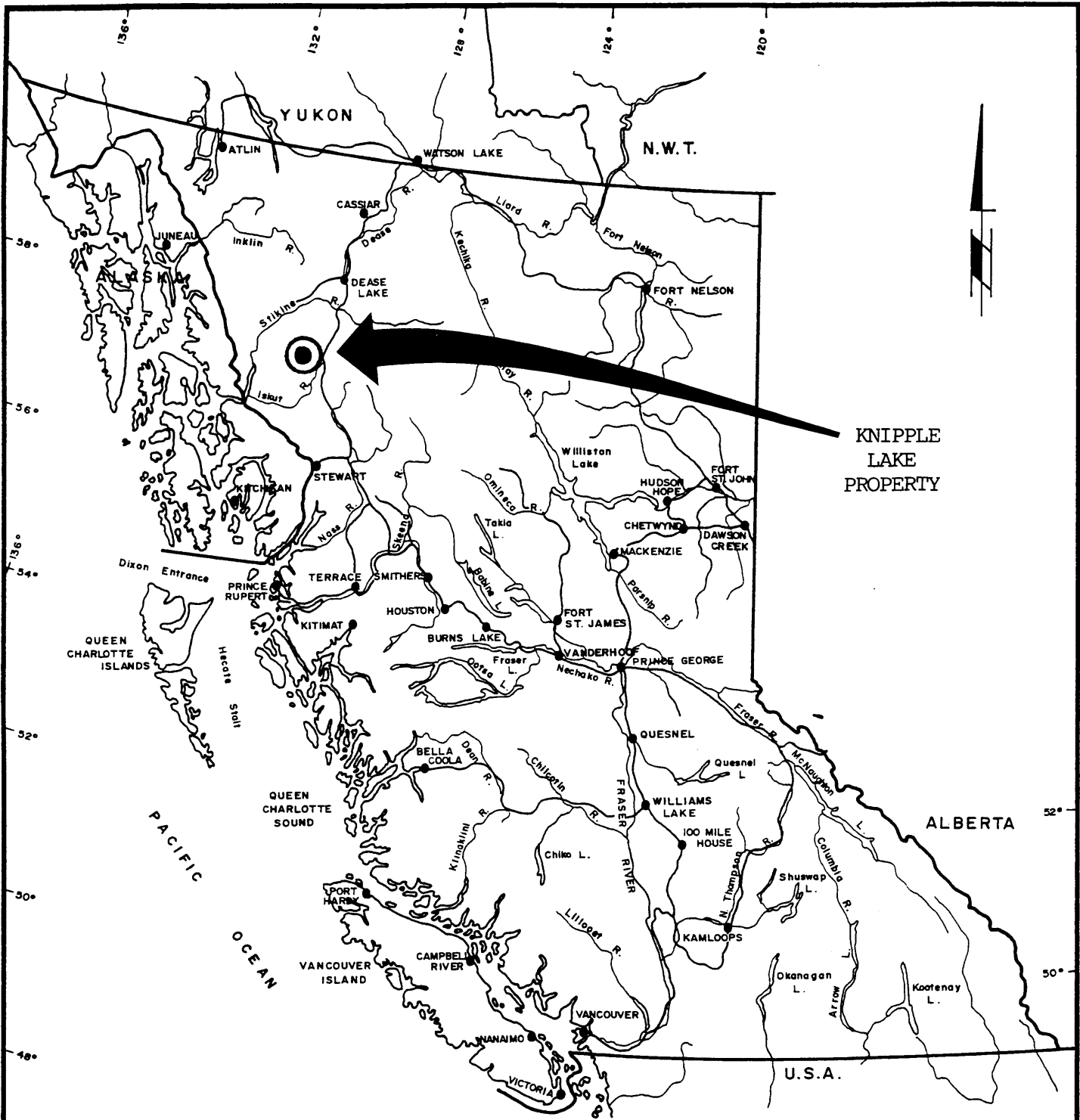
The Knipple Lake Property is located approximately 60 kilometres north of Stewart, B.C. and 25 kilometres southwest of the Stewart-Cassiar Highway (figure 1). At present, the property is accessible only by helicopter. A gravel airstrip connected by a road-barge link to the Stewart Cassiar Highway lies just 1 kilometre south of the property.

A tent camp was established on the property and was utilized for the duration of the 1991 program. A Hughes 500D helicopter chartered from Vancouver Island Helicopters was based at the camp during the drill program.

### 2.2 PHYSIOGRAPHY AND VEGETATION

The property is contained within the Boundary Ranges of the rugged Coast Mountains. Elevations range from about 450 metres in the Bowser River Valley at the south end of the property to over 1890 metres on the northwest trending ridge on the north end of the property. The east, north, and west sides of the property are bounded by thick, crevassed glaciers.

About 90% of the claims can be easily traversed. The remainder consists of steep cliffs and glaciers. Treeline lies at about 1500 metres, however receding glaciers have left large treeless areas as low as 1000 metres. Alpine areas are covered with grasses and brush typical of a cool, wet coastal alpine



KNIPPLE  
LAKE  
PROPERTY

0 100 200 KILOMETRES  
SCALE 1:8,000,000

REVISED	KNIPPLE LAKE PROPERTY	
	LOCATION MAP	
PROJ. No. _____	SURVEY BY: _____	DATE: Jan 1991
N.T.S. _____	DRAWN BY: S.K.B.	SCALE: 1:8,000,000
DWG. No.	<b>NORANDA EXPLORATION</b>	
1	OFFICE: PRINCE GEORGE, B.C.	

VANCAL 11927

environment. Below treeline mature stands of spruce, fir and hemlock dominate.

### 2.3 CLAIM DATA

The property is comprised of the following modified grid claims. Upon acceptance of this report, they will be in good standing until the indicated date. A plan of the claims is provided in figure 2.

Table 1. Claim Data

Name	Units	Record #	Record Date	Expiry Date
KL-1	18	6934	10/25/88	10/25/94
KL-2	18	6935	10/25/88	10/25/94
KL-3	18	6936	10/25/88	10/25/94
KL-4	3	304308	09/13/91	09/13/92
TREATY-12	20	7824	08/26/89	08/26/94

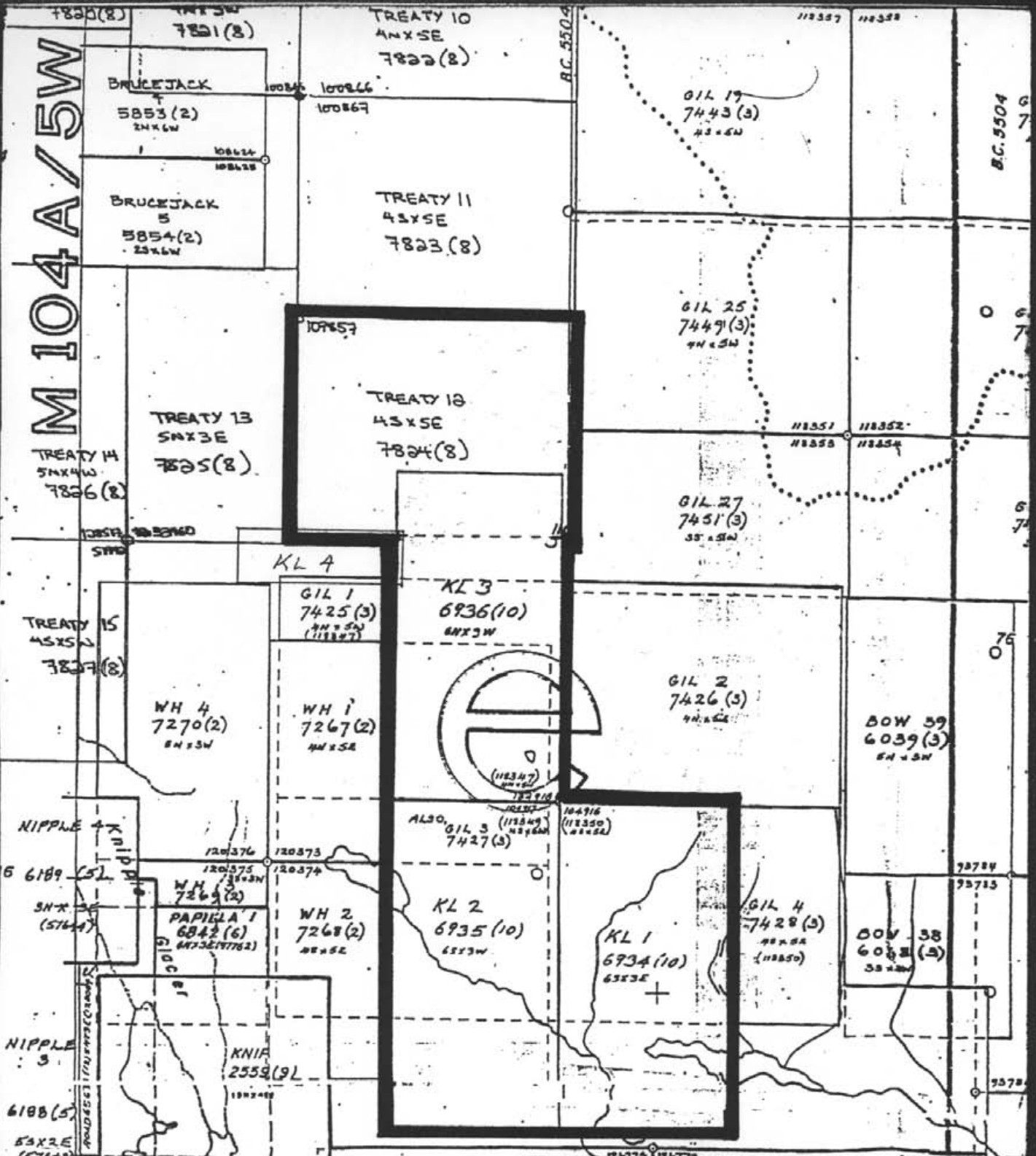
### 2.4 PREVIOUS WORK

The property lies within the Stewart-Iskut belt of mineralization which has seen considerable exploration activity in recent years. The area of the claims has been staked several times by different interests previous to Noranda however there is no record of work in the public domain and no evidence of any serious exploration was observed during the 1990 program.

1989 : Discovery of numerous quartz-calcite-sulfide vein and breccia float boulders with abundant galena, sphalerite, chalcopryrite and pyrite. Values up to 0.521 oz/t Au, 0.50 oz/t Ag, 0.14% Cu, 3.03% Pb, and 9.02% Zn were obtained. The source area was presumed to be up slope on either the KL-3 or TREATY-12 claim.

1990 : Noranda established a grid concentrating on the areas thought to be the most likely source areas of the mineralized float. This grid was soil sampled at 25 m spacing, mapped and prospected. Three moderate strength multi-element geochem anomalies were delineated and a variably silicified and quartz veined felsic volcanic unit was mapped out. This unit is evidence of intense hydrothermal activity in the area of the mineralized boulders. The source of the mineralized boulders was not discovered.

M 104A/5W



REVISED	KNIPPLE LAKE PROPERTY	
	CKAIM SKETCH	
PROJ. No. 293	SURVEY BY: _____	DATE: NOV 90
N.T.S. 104A/5W	DRAWN BY: _____	SCALE: 1:50,000
DWG. No. 2	<b>NORANDA EXPLORATION</b>	
	OFFICE: PG	

B-8-E

### 3.0 GEOLOGY

#### 3.1 REGIONAL GEOLOGY

The area lies near the western edge of the Intermontane Belt of the Canadian Cordillera, where it parallels the Coast Plutonic Complex. Recent work by both the Geological Survey of Canada and the Geological Services Branch of British Columbia provides a useful framework of the complex geology of this rugged area. The area includes four, unconformity bounded, tectonostratigraphic assemblages: 1) Paleozoic Stikine Assemblage; 2) Triassic-Jurassic volcano-plutonic complexes of Stikinia; 3) Middle and Upper Jurassic Bowser overlap assemblage; and 4) Tertiary Coast Plutonic Complex. (Anderson, 1989) This section of the Intermontane Belt forms the west limb of the "Stikine Arch," a roughly horseshoe shaped area of Upper Triassic to Jurassic stratigraphy that hosts most of the significant mineral deposits in northwest B.C. and also the Toodogone gold camp.

#### Paleozoic Stikine Assemblage

This is the oldest assemblage and it is comprised of three distinct, mainly volcanic-carbonate mappable divisions: Early Devonian limestones and intermediate to felsic volcanics, Mississippian bioclastic limestones, and Permian fragmental volcanics and limestone. These rocks are generally highly metamorphosed and deformed.

#### Triassic-Jurassic volcano-plutonic complexes (Stewart Complex)

Comprises both the Triassic Stuhini Group and the Jurassic Hazleton Group. The Stuhini Group consist of limestones and bimodal to mafic volcanics deposited in island arc environments. These rocks host the Snip and Johnny Mountain deposits. Hazleton Group rocks comprise andesitic breccias and lavas, felsic tuffs and breccias, maroon-green volcanic siltstone, greywacke, conglomerate, calcareous siltstone, and black shale also with island arc affinities. These rocks host the Eskay Creek deposits.

#### Middle and Upper Jurassic Bowser Overlap Assemblage

These are predominantly turbidite black clastics deposited in the Bowser Basin, a result of uplift to the west due to emplacement of the Coast Range Intrusives.



### Tertiary Coast Plutonic Complex

Tertiary, post-tectonic, felsic plutons characterize the Coast Plutonic Complex. Eastward younging of strata from west to east and local zones of high strain attest to intrusion and uplift of the complex.

Locally Tertiary to Recent subaerial volcanics cover low lying areas of the above strata.

Sub-volcanic intrusions accompany most of the volcanic centres of the Mesozoic island arc complexes. These range from Alaskan type ultramafics to felsic dykes. Distinctive porphyritic dykes link Upper Triassic and Lower Jurassic volcanics with their plutonic equivalents. Many of the significant mineral deposits in the Stewart Complex are found to have a close association with volcanic centres.

### 3.2 PROPERTY GEOLOGY

Geological mapping was concentrated in the area thought to be the most likely source area of the mineralization found in float described above. A 1:2500 scale geological plan of this area is presented on figure 3. Topographical control was provided by the survey grid described in the next section.

In 1991 detailed mapping was undertaken mainly in the area extending from the mineralized boulder field in the grid west area centered about the 11000W tieline to the north end of the grid. This is the area containing the soil geochem anomalies outlined last year and covered with I.P. surveys this year. The results of this work are shown on figure 3.

### Surficial Geology

Most of the bedrock exposures are found along the top and west sides of the north to northwest trending ridge that dominates the Treaty-12 and KL-3 claim. North and northeast of this ridge the slope abruptly drops forming inaccessible cliffs that end at glacial ice. The mapped area forms a relatively consistent west facing slope averaging about 30 degrees that ends at a wide south flowing receding glacier. The slope is covered with a thin to moderately thick apron of fine talus, minor moraine outwash and poorly developed, slumping alpine soil. Vegetation has stabilized most of the area at lower elevations. The glacier to the west of this slope has receded approximately

50 metres vertically from its maximum size and has left a 100 to 200 metre wide strip underlain with a semi-consolidated basal till which is in places covered with coarse and loose lateral and medial moraine and talus that is still unvegetated. The edge of this strip is indicated on figure 3. The east-west trending 600 to 700 metre wide zone shown in the south grid area is dominated by more recent coarse moraine and outwash from the small receding icefields at the southeast corner of the map.

### Bedrock Lithologies

The mapped area is underlain by two packages of volcanic rocks: an extensive sequence of interbedded andesitic flows and tuffs (Unit # 1), agglomerates (Unit # 2), and a less dominant unit of felsic tuffs and flows (Unit # 3).

Unit 1 consists of dark brown, maroon, grey and green vaguely layered, subrounded to subangular clasts of andesitic rocks of unit 2 in a darker very fine matrix. In areas of sufficient outcrop mappable beds in the order of 25 to 100 metres thick can be traced over several hundred metres. The unit was further subdivided to include two mappable beds of grey-brown agglomerate (unit 1b) and maroon tuffs (unit 1c).

Unit 2 has been subdivided into four mappable divisions:  
2a) Massive, dark varicoloured, feldspar porphyritic flows.  
2b) Vaguely layered, fine lithic tuff +/- feldspar phenocrysts.  
2c) Massive, dark amygdaloidal feldspar porphyritic flows.  
2d) Massive, dark greenish brown chloritic flows. Minor dark grey dolomitic sediments are shown as subunit 2e.

Unit 3 has been subdivided into two mappable divisions:  
3a) Massive to vaguely layered, variably silicified and quartz veined pale grey to brown, feldspar porphyritic lapilli tuff.  
3b) Massive pale grey to buff, aphyric siliceous rhyolite.

These rocks are interpreted to have been rapidly deposited in dominantly submarine environment proximal to a volcanic centre. The felsic volcanics may indicate the youngest eruptive event. They are considered part of the Lower Jurassic Hazleton Group. Units 1 and 2 may correlate with the Betty Creek Formation and unit 3 with the Mt. Dilworth Formation.

### Structure

Where layering is discernable two general attitudes were measured. In the central east portion of the mapped area, attitudes are found to trend from about 060 to 090 degrees and dip moderately to steeply southward. Several hundred metres south of this area the dominant trend is found to be around 180 degrees with a variable westward dip.

Two 070 degree trending faults are interpreted to cross the gridded area as shown on figure 3. The interpretation is based on strong contrasts in magnetic relief indicated from a ground magnetic survey completed later in the program and on the presence of strongly foliated and brecciated rocks in the vicinity of Line 10100 N, 10350 W.

### Alteration

There is very little alteration in the gridded area other than Unit 3a which is variably silicified and laced with a pervasive network of white quartz stringers and veinlets. Disseminated pyrite is common. Another notable area of alteration is around the galena barite showing located at 9600N / 11125W. There is silicification, quartz veining and 1-2 % pyrite associated with this showing. Weakly chloritized andesitic rocks are found scattered throughout the east-central grid area. In the moraine fields on the grid south area numerous boulders of strongly quartz-sericite-pyrite altered felsic porphyritic intrusive are derived from exposures on the main ridge immediately east of the KL-3 claim. This intrusive is believed to be the source of the hydrothermal system that produced the observed alteration.

### Mineralization

A boulder field at the southwest grid corner from 11000W, 8700N to 11250W, 9225N contains over one hundred boulders of quartz-calcite breccia and vein material containing coarse sphalerite, galena, pyrite, and chalcopyrite (figure 3). There are often recrystallized and silicified wall rock material within and adjacent to the quartz-calcite-sulphide mineralization. The boulders are subrounded to subangular and range in size from about 10 to 100 centimetres in diameter. Of forty-eight samples analyzed, all have anomalous gold values, and 18 have values greater than 4 gm/t Au and 8 greater than 8.5 gm/t Au. Base metal values average 0.18% Cu, 0.89% Pb, and 3.14% Zn; silver averages 9.3 gm/t. Intense prospecting failed to locate the

source of the boulders on the largely overburden covered slopes above, however the presence of minor sulphide bearing clots and quartz veins with anomalous gold values (up to 1260 ppb) suggests the source is within the grid area.

Prospecting in 1991 extended the boulder field as far north as 9220N, 11310W where similar float contained Au values to 7930 ppb Au. Narrow quartz-carbonate veins (less than 1 metre wide) found in outcrop near 9400N, 11150W and 9600N, 11050W contain spotty sulphides and low Au values (to 240 ppb). Silicified felsic tuffs laced with a network of fine quartz veins are common in the central grid area however chip samples returned a maximum of 52 ppb Au and very low base metals.

A subangular boulder approximately 0.5 metres in diameter located at 10290N, 11365W of coarse, milky white quartz contains irregular patches of coarse sulphides and returned values of 4,780 ppb Au, 36,688 ppm Pb and 69,518 ppm Zn (#113856). This is the most significant result obtained from outside the boulder field, which lies about 2 km to the south. Only one mineralized boulder was found even though the area is covered with thousands of moraine and felsensmeer boulders and very little vegetation. The style of mineralization in the boulder field to the south is noticeably different and it is thought that the source areas are distinct.

#### 4.0 DIAMOND DRILLING

The 1991 diamond drilling program consisted of 346.5 m of BDBGM core in six holes. The first three holes were drilled to test weak to moderate strength IP chargeability anomalies located upslope and up ice from the mineralized boulder field. The first hole was also coincident with a galena-carbonate showing at the upslope edge of a multielement geochem anomaly. The fifth hole was drilled to test the central area of the main soil geochem anomaly and the sixth hole was drilled to test a VLF anomaly. The drill hole locations are plotted on Figure 3 and the drill logs are in appendix VII. Cross sections displaying geology, analytical results and topography are shown on Figures 22-26.

A unitized, helicopter transportable, hydraulic wireline drill using thin wall BDBGM rods was utilized. The core is currently stored at the campsite on the property.

KN-91-01 : Located at 9600N / 11050W, drilled at a dip of -45' and bearing of 290'. This hole tested a moderate IP chargeability anomaly coincident with a galena-carbonate showing at the upslope edge of the largest multielement soil geochem

anomaly. It was drilled to a depth of 84.12 m and intersected 10.97 m of overburden and 73.15 m of maroon lapilli tuffs and agglomerates. No significant mineralization was encountered. The I.P. response is attributed to intensely fractured zones lined with clayey selvages and seams.

KN-91-02 : Located at 9400N / 10800W, drilled at a dip of -60° and a bearing of 290°. This hole was drilled to test a moderate IP chargeability anomaly. It was drilled to a depth of 93.27 m and intersected 7.92 m of overburden and 83.35 m of interbedded green andesitic volcanics and maroon agglomerate. No significant mineralization was encountered. The I.P. response is attributed to intensely fractured zones lined with clayey selvages and seams.

KN-91-03 : Located at 9000N / 10600W, drilled at dip of -45° and a bearing of 110°. This hole was abandoned at 15.84 m due to seizing rods in fractured rocks.

KN-91-03a : Located at 9000N / 10600W, drilled at a dip of -80° and a bearing of 110°. This hole was drilled to test a moderate IP chargeability anomaly. It was drilled to a depth of 67.06 m and intersected 6.70 m of overburden and 60.36 m of interbedded green andesitic volcanics and maroon agglomerate. No significant mineralization was encountered. The I.P. response is attributed to intensely fractured zones lined with clayey selvages and seams.

KN-91-04 : Located at 9600N / 11275W, drilled at dip of -45° and a bearing of 110°. This hole was drilled to test the center of the multielement soil geochemistry anomaly north and up ice of the mineralized boulder field. It was drilled to a depth of 57.91 m and intersected 1.52 m of overburden and 56.39 m of maroon agglomerate. The hole had to be abandoned at 57.91 m due to poor ground conditions. No significant mineralization was encountered up to this point.

KN-91-05 : Located at 9080N / 11140W, drilled at dip of -45° and a bearing of 080°. This hole was drilled to test the VLF anomaly coincident with the mineralized boulder field. It was drilled to a depth of 29.87 m and was unable to penetrate further due to seizing rods and casing in poor ground conditions. This hole intersected 29.87 m of badly broken green andesitic volcanics with abundant sand and clay layers. The target depth was never reached. No significant mineralization was encountered up to this point.

Geochem analyses on selected intervals returned a maximum of 100 ppb Au from a zone of brecciation and quartz flooding in hole

#1. No significant base metal values were received.

The drilling program was stopped prematurely due to cost overruns resulting from poor drilling conditions. Numerous open, sand and gravel filled fractures inhibiting water return and lubrication led to excessive bit wear, downhole equipment destruction, and unbudgeted servicing costs.

### 5.0 CONCLUSIONS

The program was unsuccessful in locating the source of the mineralized boulder field. It is still believed that they have not been transported very far and their source is somewhere on the KL-3 or Treaty 12 mineral claims.

### 6.0 RECOMMENDATIONS

No further work is recommended at this time. The Treaty 12 claim should be returned to the vendor before the next payment is due (Aug. 31, 1992).

APPENDIX I

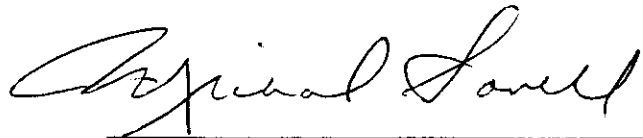
STATEMENT OF QUALIFICATIONS

APPENDIX I

STATEMENT OF QUALIFICATIONS

I, Michael Savell, of the City of Prince George, Province of British Columbia, do certify that:

1. I am a geologist residing at 3507 Rosia Road, Prince George, British Columbia.
2. I am a graduate of Dalhousie University, Halifax, Nova Scotia with a Bachelor's of Science (Honours) degree in Geology.
3. I am a member in good standing of the Geological Association of Canada, the Prospector's and Developer's Association and the B.C.-Yukon Chamber of Mines.
4. I presently hold the position of Sr. Project Geologist with Noranda Exploration Company, Limited and have been in their employ since 1980.



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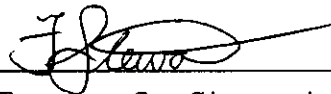
Michael Savell  
Sr. Project Geologist  
Noranda Exploration Co., Ltd.  
(no personal liability)



STATEMENT OF QUALIFICATIONS

I, FRASER J. STEWART, hereby certify that:

1. I am a geologist residing at 302 - 1910 Renwick Crescent, Prince George, B. C.
2. I graduated from the University of Alberta in April 1989, with the degree of Bachelor of Science in Geology.
3. I have been employed by Noranda Exploration Company, Limited as a geologist since May 1989.
4. I personally took part in the surveys described in this report and that this report is based upon a personal knowledge of the property.



---

Fraser J. Stewart, (B.Sc.)

APPENDIX II

ANALYTICAL PROCEDURE

Soils, Silts, Rocks

The samples are dried and screened to -80 mesh. Rock samples are pulverized to -120 mesh. A 0.2 gram sample is digested with 3 ml of  $\text{HClO}_4/\text{HNO}_3$  (4 to 1 ratio) at 203° C for four hours, and diluted to 11 ml with water. A Leeman PS 3000 is used to determine elemental contents by I.C.P. Note that the major oxide elements and Ba, Be, Ce, Ga, La and Li are rarely dissolved completely from geological materials with this acid dissolution method.

For Au analyses, a 10.0 gram sample of -80 mesh material is digested with aqua regia and determination made by A.A.

Heavy Mineral Concentrates

The entire concentrate is digested in aqua regia solution, and elemental concentrations of Au, Ag, Cu, Pb, and Zn are determined by A.A.

APPENDIX III  
ANALYTICAL RESULTS

GEOCHEMICAL ANALYSIS CERTIFICATE

Knipple (B) DDH-91-02, 03, 04

Noranda Exploration Co. Ltd. PROJECT 9109-091 293

File # 91-4632

1050 Davie St., Vancouver BC V6E 1M4



SAMPLE#	Mo	Cu	Pb	Zn	Ag	Ni	Co	Mn	Fe	As	U	Au	Th	Sr	Cd	Sb	Bi	V	Ca	P	La	Cr	Mg	Ba	Ti	B	Al	Na	K	V	Au*
	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	%	ppm	ppm	%	ppm	%	ppm	%	%	%	ppm	ppb
040826 DR	1	5	2	113	.1	4	18 1540	5.62	3	5	ND	1	24	.2	2	2	104	1.98	.090	2	21	1.62	455	.14	5	2.92	.04	.43	1	7	
040827 DR	1	6	2	101	.2	3	16 1249	4.77	4	5	ND	1	104	.5	2	2	77	3.36	.090	6	11	1.68	696	.14	2	2.32	.05	.34	1	2	
040828 DR	1	1	2	99	.1	3	17 961	4.38	2	5	ND	1	113	.8	2	2	42	2.35	.081	4	9	1.52	1842	.10	3	2.29	.05	.37	1	2	
040829 DR	1	36	5	104	.5	3	14 1711	3.81	9	5	ND	1	205	.2	2	2	46	6.68	.075	8	16	1.55	1988	.06	4	2.65	.02	.56	1	6	
RE 040834 DR	1	13	2	104	.3	3	18 1315	5.30	2	5	ND	1	110	.9	2	2	76	3.69	.077	4	12	1.53	988	.13	2	2.71	.06	.34	1	2	
040830 DR	1	14	3	94	.3	3	14 1236	3.78	2	5	ND	1	99	1.3	2	2	44	4.17	.084	7	6	1.47	407	.08	2	2.13	.03	.31	1	3	
040831 DR	1	4	2	79	.1	2	14 927	4.27	2	5	ND	1	128	.2	2	2	44	3.10	.074	5	11	1.14	647	.11	3	1.96	.04	.43	1	4	
040832 DR	1	4	2	95	.1	3	16 1116	4.79	2	5	ND	1	121	.4	2	2	55	3.71	.076	3	14	1.42	765	.15	3	2.26	.05	.36	1	2	
040833 DR	1	8	3	99	.2	3	16 1097	5.26	2	5	ND	1	91	.5	2	2	63	3.34	.081	2	14	1.41	394	.17	2	2.39	.06	.38	1	4	
040834 DR	1	11	2	105	.1	3	18 1298	5.28	2	5	ND	1	107	1.1	2	2	73	3.66	.076	5	13	1.53	943	.13	2	2.65	.06	.35	1	1	
040835 DR	1	36	8	112	.2	4	20 1273	5.50	3	5	ND	1	57	.5	2	2	74	1.68	.078	4	16	1.76	908	.14	2	2.91	.06	.33	1	6	
040836 DR	1	21	6	115	.4	4	19 1188	5.89	6	5	ND	1	48	.8	2	2	82	1.58	.087	5	9	1.90	542	.16	3	2.89	.05	.27	1	7	
040837 DR	1	31	6	104	.4	2	16 1585	5.13	4	5	ND	1	120	.3	2	2	80	5.48	.071	6	5	1.90	709	.15	2	2.72	.04	.22	1	5	
040838 DR	1	8	4	135	.7	3	14 1381	4.91	3	5	ND	1	68	.5	2	2	70	1.94	.092	12	12	.96	509	.08	3	1.86	.01	.46	1	3	
040839 DR	1	7	7	76	.5	2	15 1317	3.39	17	5	ND	1	73	.3	2	2	36	4.58	.102	12	8	.30	2036	.03	5	1.30	.01	.60	1	5	
040840 DR	1	6	5	71	.3	2	11 2471	2.30	26	5	ND	1	141	.3	2	2	28	11.48	.086	13	6	.47	1095	.02	2	1.19	.01	.44	1	2	
040841 DR	1	5	10	72	.2	1	10 3026	3.18	14	5	ND	1	197	.4	2	2	36	11.51	.096	15	5	.54	1029	.06	3	1.30	.02	.42	1	4	
040842 DR	1	9	6	91	.4	2	12 1556	3.41	59	5	ND	1	114	.5	4	2	43	5.99	.100	13	3	.58	1169	.04	3	1.30	.01	.38	1	4	
040843 DR	2	15	25	125	.5	3	17 1862	4.66	5	5	ND	1	136	.2	2	2	62	7.44	.080	11	10	.94	295	.06	4	1.90	.01	.52	1	3	
040844 DR	1	4	13	85	.4	4	13 1904	3.89	4	5	ND	1	188	.2	3	2	64	8.67	.070	12	10	.63	1197	.07	4	1.45	.01	.46	1	6	
040845 DR	1	172	5	184	3.1	19	13 1416	3.51	4	5	ND	1	81	3.3	2	2	33	4.15	.075	11	22	.64	685	.03	4	1.32	.01	.35	164	8	
040846 DR	1	30	43	112	3.4	4	18 1296	3.27	56	5	ND	1	139	1.2	3	2	16	5.67	.085	12	7	.34	402	.07	4	.97	.03	.39	5	2	
040847 DR	1	65	19	35	.8	4	7 1374	2.51	29	5	ND	2	132	.3	2	2	16	7.34	.086	11	8	.16	191	.07	3	.72	.02	.46	2	2	
040848 DR	1	4	8	57	.5	3	10 1418	3.06	10	5	ND	3	174	.7	2	2	51	7.25	.080	11	7	.50	588	.05	2	1.07	.03	.41	1	2	
040849 DR	1	9	8	56	.2	3	11 1146	2.77	7	5	ND	2	152	.4	2	2	42	5.77	.093	11	4	.57	656	.05	2	1.11	.02	.39	1	1	
040850 DR	1	17	11	100	.1	5	16 1282	3.48	3	5	ND	1	99	.5	2	2	47	4.95	.076	11	6	1.06	144	.06	2	1.42	.01	.25	1	1	
040901 DR	1	11	13	66	.3	7	10 1306	3.55	11	5	ND	2	40	.3	4	2	46	1.53	.082	17	17	.58	216	.07	2	1.03	.02	.26	1	2	
040902 DR	1	10	5	70	.7	7	13 1417	3.46	6	5	ND	3	147	.2	4	2	56	3.17	.087	16	22	1.04	1634	.05	2	1.43	.04	.24	1	2	
040903 DR	1	7	2	72	.2	10	11 1329	3.56	9	5	ND	1	99	.6	6	2	31	3.28	.086	15	25	.60	279	.06	2	1.08	.01	.36	1	1	
040904 DR	1	8	9	69	.1	8	12 1228	3.28	9	5	ND	1	106	.2	6	2	53	3.90	.092	18	25	.97	580	.08	2	1.35	.02	.37	1	3	
040905 DR	1	26	31	40	1.5	4	13 1066	3.45	74	5	ND	1	68	.2	13	2	61	3.00	.084	13	7	.17	463	.07	4	.75	.01	.39	2	27	
040906 DR	1	46	17	50	1.4	4	12 1063	3.68	69	5	ND	1	65	.3	12	2	60	3.58	.089	15	7	.17	392	.05	2	.78	.01	.39	3	100	
040907 DR	1	43	10	85	.8	5	16 1148	4.17	12	5	ND	1	85	.2	3	2	67	3.99	.088	13	8	.46	509	.05	2	1.19	.01	.42	1	7	
040908 DR	1	132	5	105	.5	4	17 1567	3.68	31	5	ND	1	202	.7	5	2	54	6.07	.082	12	7	.97	1257	.06	2	1.61	.01	.43	1	20	
040909 DR	1	22	16	71	.4	4	12 1303	3.43	51	5	ND	1	124	.7	7	2	43	4.28	.104	15	10	.60	548	.05	5	1.37	.01	.57	1	1	
STANDARD C/AU-R	19	61	39	130	7.3	69	32 1029	3.92	43	19	6	39	53	18.6	17	20	60	.47	.087	38	59	.87	175	.08	33	1.91	.06	.14	13	460	

SEP 27 1991  
 DATE RECEIVED

ICP - .500 GRAM SAMPLE IS DIGESTED WITH 3ML 3-1-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR AND IS DILUTED TO 10 ML WITH WATER. THIS LEACH IS PARTIAL FOR MN FE SR CA P LA CR MG BA TI B W AND LIMITED FOR NA K AND AL. AU DETECTION LIMIT BY ICP IS 3 PPM. ASSAY RECOMMENDED FOR ROCK AND CORE SAMPLES IF CU PB ZN AS > 1%, AG > 30 PPM & AU > 1000 PPB. SAMPLE TYPE: CORE AU\* ANALYSIS BY ACID LEACH/AA FROM 10 GM SAMPLE. Samples beginning 'RE' are duplicate samples.  
 DATE REPORT MAILED: Sept 25/91 SIGNED BY: [Signature] D. TOYE, C. LEONG, J. WANG; CERTIFIED B.C. ASSAYERS

APPENDIX IV

DRILL LOGS



NORANDA EXPLORATION CO. LTD.  
DIAMOND DRILL LOG

PROPERTY : KNIPPLE LAKE  
HOLE No. : KN-91-01

PAGE : 2

INTERVAL (m)		MAJOR/MINOR UNITS	DESCRIPTION	SAMPLE NUMBER	INTERVAL (m)		SAMPLE WIDTH	GEOCHEMICAL SAMPLES								
FROM	TO				FROM	TO		Au ppb	Ag ppm	Cu ppm	Pb ppm	Zn ppm	As ppm	Se ppm		
26.82	28.00	qtzstr	- common carb +/- Qtz stringers @ 30-45 cm * - maroon feld. lapilli tuff w common Qtz stringers @ 10-15 to cm and others at random orientations *													
28.00	84.12	AGBLDM	MARCON AGGLOMERATE - greyish maroon silic. carb. rich agglomerate comprised primarily of subrounded maroon tuff clasts - wkly precciated and subsequently carb +/- Qtz flocced so that carbonate comprises about 5% of rock - averages about 70% recovery *	40903	35.66	39.32	3.66	1	.2	7	2	72	9	6		
39.40	39.90	bleach	- washed out to a lit grey color (dueteric) around a small fault (1 cm wide) that consists of coarse sand and gouge *													
43.28	43.40	fault	- fault gouge with maroon volc. frags *													
44.98	45.00	fault	- fault gouge w small maroon volc. frags *	40904	45.05	50.00	5.75	3	.1	8	9	69	9	6		
47.00	47.05	fault	- fault gouge w maroon volc. frags *													
47.24	47.40	fault	- fault gouge w maroon volc. frags *													
47.70	47.90	fault	- fault gouge w maroon volc. frags *													
48.77	56.99	brx'd	Breccia Zone - this zone is intensely brecciated w v strong to int. Qtz flooding comprising 20-30% of rock with large fragments of maroon volcanics (5-10 cm) * 52.42 - 52.50 : fault gouge w maroon volc. fragments * 53.95 - 55.78 : sand and gouge w common very rounded fragments (due to	40905 40906	50.00 54.50	54.50 57.50	3.70 3.00	27 100	1.5 1.4	26 46	31 17	40 50	74 69	13 12		







PROPERTY : KNIPPLE LAKE  
HOLE No. : KN-51-02

INTERVAL (m)		MAJOR/MINOR UNITS	DESCRIPTION	SAMPLE NUMBER	INTERVAL (m)		SAMPLE WIDTH	Au	Ag	Cu	Pb	Zn	GEOCHEMICAL SAMPLES	
FROM	TO				FROM	TO							ppm	ppm
			2.0 cm. subrounded *											
31.10	31.50	fault	- fault zone w/ly brecciated w fault gouge and coarse sand *											
35.50	35.70	fault	- fault zone with gouge, coarse sand and angular fragments of green volcanics *											
35.00	35.30	fault	- fault zone w gouge and angular fragments *											
41.05	42.20	agglom	- zone containing large (10 cm) clasts of maroon feld. lapilli tuff *											
42.30	42.35	fault	- fault w angular clasts and gouge *	40828	43.00	45.00	2.00	2	.1	1	2	99	2	2
43.40	43.50	fault	- fault w gouge and angular fragments *											
43.90	44.90	fault	- fault w gouge and angular fragments *											
45.50	45.70	fault	- fault zone w gouge and carb vn (2 cm) @ 20 to CA *											
47.10	47.15	fault	- fault w gouge and angular fragments *											
49.20	49.30	fault	- fault w gouge and angular fragments *											
51.00	52.00	fault	- fault zone w gouge and angular fragments and some wk ep alt in a sheared zone 5 cm wide @ 40 to CA *	40829	53.00	56.50	3.50	6	.5	36	5	104	9	2
53.10	53.50	fault	- fault zone w gouge and subangular frags - some evidence of shearing (gashes) - wk carbonate alt *											
54.10	54.35	shear	- shear zone w perv carb alt *											
57.00	56.00	carovn	- large carb veins @ 40-50 to CA *	40830	62.90	65.53	2.63	3	.3	14	3	94	2	2
63.70	64.10	fault	- fault zone w gouge and angular frags *											
64.00	65.60	fault	- fault w gouge and some wk ep alt *	40831	68.00	72.85	4.85	4	.1	4	2	79	2	2





NGRANDA EXPLORATION CO. LTD.  
DIAMOND DRILL LOG

PROPERTY : KNIPPLE LAKE  
HOLE No. : KN-91-04  
Grid System : KNIPPLE  
Collar Eastings : 11275.000  
Collar Northings : 5600.000  
Collar Elevations : 1304.000  
Collar Bearing : 116.00  
Grid Baseline : 20.00

Collar Inclination : -45.00  
Grid Bearing : 110.00  
Final Depth : 57.91  
Claim No. :

PAGE : 1

Logged by : FRASER STEWART  
Date : SEPT. 14/91 - SEPT. 15/91  
Downhole Survey : NDNE  
Drilled By : J.T. THOMAS  
Core Size : 80

INTERVAL (m)		MAJOR/MINOR UNITS	DESCRIPTION	SAMPLE NUMBER	INTERVAL (m)		SAMPLE WIDTH	GEOCHEMICAL SAMPLES								
FROM	TO				FROM	TO		Au ppm	Ag ppm	Cu ppm	Pb ppm	Zn ppm	As ppm	Sb ppm		
0.00	1.52	OVRBDN	OVERBURDEN *													
1.52	57.91	AGGLOM	MARDON AGGLOMERATE - deep maroon colored subrounded to round clasts in a darker maroon f.g.d. matrix - less common lt green tuff clasts (round) - sizes vary from 0.25-10.00 cm - common zones of carbonate +/- Qtz flooding (Qtz rare) - approx. 100 % recovery *	40846	9.60	12.60	3.00	2	3.1	172	5	184	4			
12.40	12.45	qtzcarb	- Qtz-carb vein with maroon fragments *													
13.60	13.90	dtralt	- washed out to a lt grey - diueteric alt ? *													
14.60	15.40	carb	- carbonate rich matrix *	40847	14.63	18.20	3.57	2	.8	65	19	35	29			
17.60	17.65	fault	- fault with grey brown gouge *													
26.30	26.31	qtzvn	- quartz vein @ 90 to CA *	40848	28.60	31.60	3.00	2	.5	4	8	57	10			
29.85	29.86	qtzclvn	- quartz-chlorite vein @ 90 to CA *	40849	35.90	39.60	3.70	1	.2	9	8	56	7			
40.00	46.00	agglom	- medium grey green feldspar lapilli tuff clasts and frags become very abundant * 43.45 - 43.50 : Qtz-carb flooding *													
46.00	51.20	fltzn	46.00 - 57.91 : rock becomes very badly broken with what appears to be fault splays (?) * 46.25 - 46.30 : fault 46.94 - 46.95 : fault	40850	48.30	51.50	3.20	1	.1	17	11	100	3			













Abbreviations:

alt - altered  
approx - approximate  
CA - core axis  
carb - carbonate  
chl - chloritic  
cm - centimetre  
ep - epidote  
f - fine  
feld - feldspar  
frags - fragments  
gd - grained  
int - intense  
lt - light  
med - medium  
orient - orientated  
perv - pervasive  
porph - porphyritic  
pyrx - pyroxene  
qtz - quartz  
silic - silicified  
tr - trace  
v - very  
vnltts - veinlets  
volc - volcanics  
w - with  
wk - weak  
wkly - weakly  
w/o - without

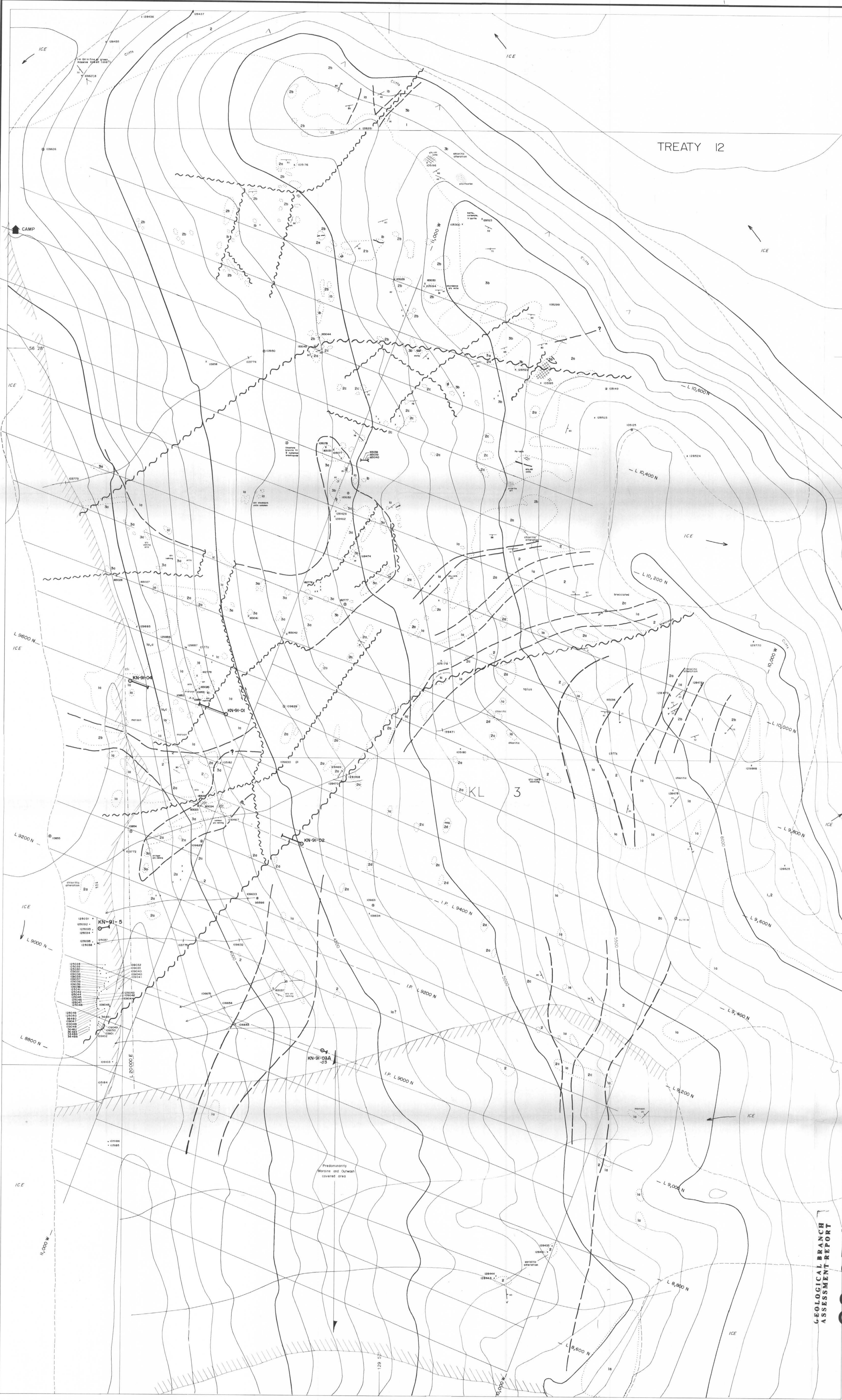
APPENDIX V

STATEMENT OF COSTS

CLAIMS : KL-1, KL-2, KL-3, TREATY-12  
DATES : SEPTEMBER 1 TO OCTOBER 1, 1991  
TYPE OF REPORT : DIAMOND DRILLING

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1) WAGES	
Rate per day : \$153.58	
No. of days : 54	
Dates : 09/01/91 to 10/01/91	
TOTAL	\$ 8,293.59
2) FOOD, ACCOMMODATION, AND SUPPLIES	
Rate per day : \$49.72	
No. of days : 106	
Dates : 09/01/91 to 10/01/91	
TOTAL	\$ 5,270.32
3) TRANSPORTATION	
Rate per day : \$298.89	
No. of days : 106	
Dates : 09/01/91 to 10/01/91	
TOTAL	\$31,682.50
4) ANALYSES	
34 core samples analyzed by ICP (30 elements) and AA (Au) @ \$15.00 each :	\$ 510.00
5) CONTRACTORS	
346.5 metres diamond drilling @ \$64.47 per metre	\$18,646.80
Drill pad construction	\$ 5,380.00
6) COST OF PREPARATION OF REPORT	
Author	\$ 500.00
Drafting	\$ 200.00
Typing	\$ 75.00
TOTAL	\$ 775.00
<b>TOTAL COST</b>	<b>\$ 70,558.21</b>



**ANALYTICAL RESULTS**

**SILTS**

SAMPLE NO.	Au ppm	Ag ppm	Cu ppm	Pb ppm	Zn ppm	As ppm
103772	20	0.8	27	10	102	8
103773	3	0.2	21	2	94	17
103774	10	0.6	28	15	91	5
103775	10	0.2	29	7	103	8
104653	10	0.6	24	4	107	11
104654	10	0.6	29	2	92	27
104655	20	0.4	22	2	95	17
104656	5	0.2	21	2	102	16
104657	85	0.4	22	2	82	11
104658	85	0.4	22	2	82	11
104659	85	0.4	22	2	82	11
104660	85	0.4	22	2	82	11
104661	85	0.4	22	2	82	11
104662	85	0.4	22	2	82	11
104663	85	0.4	22	2	82	11
104664	85	0.4	22	2	82	11
104665	85	0.4	22	2	82	11
104666	85	0.4	22	2	82	11
104667	85	0.4	22	2	82	11
104668	85	0.4	22	2	82	11
104669	85	0.4	22	2	82	11
104670	85	0.4	22	2	82	11
104671	85	0.4	22	2	82	11
104672	85	0.4	22	2	82	11
104673	85	0.4	22	2	82	11
104674	85	0.4	22	2	82	11
104675	85	0.4	22	2	82	11
104676	85	0.4	22	2	82	11
104677	85	0.4	22	2	82	11
104678	85	0.4	22	2	82	11
104679	85	0.4	22	2	82	11
104680	85	0.4	22	2	82	11
104681	85	0.4	22	2	82	11
104682	85	0.4	22	2	82	11
104683	85	0.4	22	2	82	11
104684	85	0.4	22	2	82	11
104685	85	0.4	22	2	82	11
104686	85	0.4	22	2	82	11
104687	85	0.4	22	2	82	11
104688	85	0.4	22	2	82	11
104689	85	0.4	22	2	82	11
104690	85	0.4	22	2	82	11
104691	85	0.4	22	2	82	11
104692	85	0.4	22	2	82	11
104693	85	0.4	22	2	82	11
104694	85	0.4	22	2	82	11
104695	85	0.4	22	2	82	11
104696	85	0.4	22	2	82	11
104697	85	0.4	22	2	82	11
104698	85	0.4	22	2	82	11
104699	85	0.4	22	2	82	11
104700	85	0.4	22	2	82	11

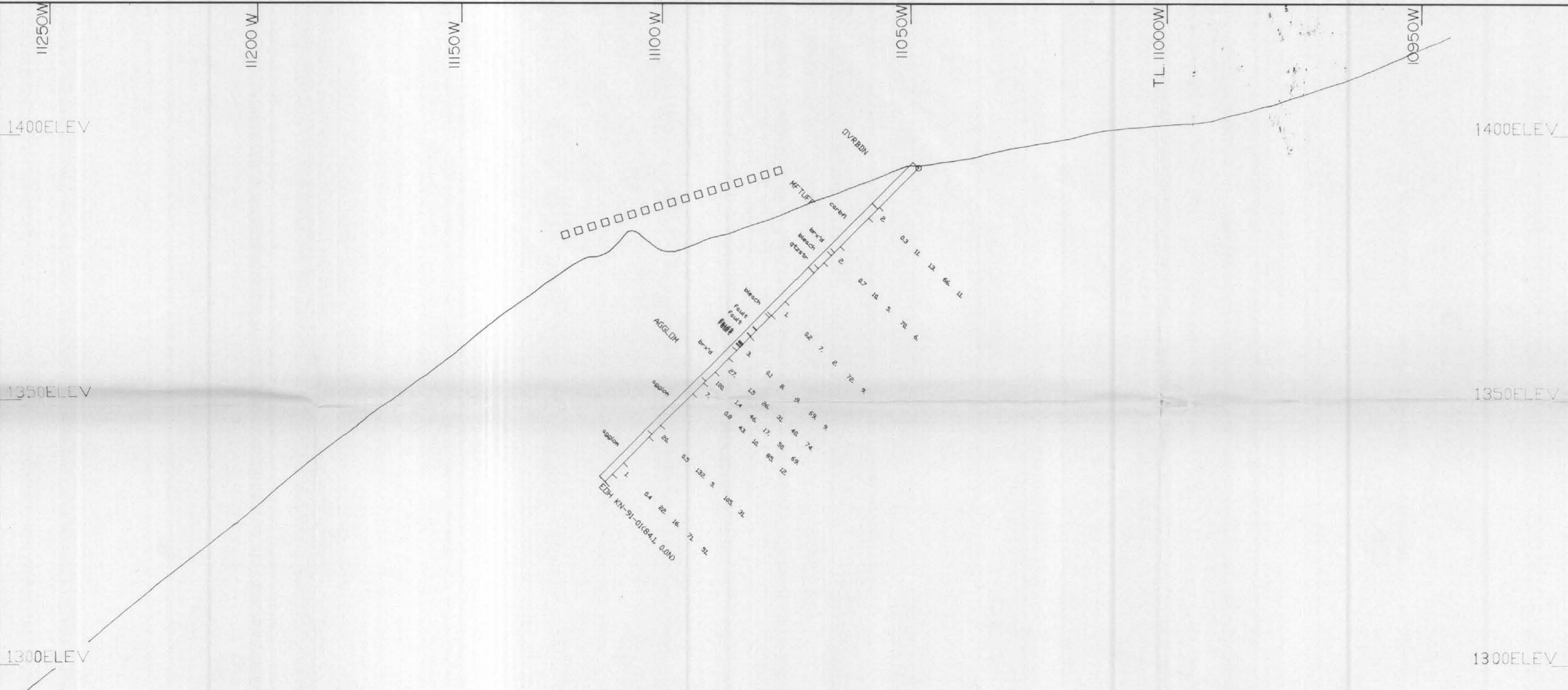
**ROCKS**

SAMPLE NO.	Au ppm	Ag ppm	Cu ppm	Pb ppm	Zn ppm	As ppm	Sb ppm
34480	2100	1.4	508	142	5335	28	2
34481	420	0.2	236	42	2125	106	11
34482	84	2.4	1026	250	87	90	3
34483	1680	2.4	1026	250	87	90	3
34484	42700	11.2	288	619	2487	191	4
34485	3750	2.4	141	281	4739	11	2
95598	390	1.7	91	204	431	163	6
105102	280	0.4	8	7	315	4	2
105103	84	0.4	23	54	35	327	15
105104	49	0.2	12	17	45	48	4
105105	35	0.3	3	11	126	4	2
105106	20	0.4	19	25	23	33	2
105107	70	1.4	19	28	42	11	2
105108	35	0.3	3	11	126	4	2
105109	20	0.4	19	25	23	33	2
105110	70	1.4	19	28	42	11	2
105111	35	0.3	3	11	126	4	2
105112	20	0.4	19	25	23	33	2
105113	70	1.4	19	28	42	11	2
105114	35	0.3	3	11	126	4	2
105115	20	0.4	19	25	23	33	2
105116	70	1.4	19	28	42	11	2
105117	35	0.3	3	11	126	4	2
105118	20	0.4	19	25	23	33	2
105119	70	1.4	19	28	42	11	2
105120	35	0.3	3	11	126	4	2
105121	20	0.4	19	25	23	33	2
105122	70	1.4	19	28	42	11	2
105123	35	0.3	3	11	126	4	2
105124	20	0.4	19	25	23	33	2
105125	70	1.4	19	28	42	11	2
105126	35	0.3	3	11	126	4	2
105127	20	0.4	19	25	23	33	2
105128	70	1.4	19	28	42	11	2
105129	35	0.3	3	11	126	4	2
105130	20	0.4	19	25	23	33	2
105131	70	1.4	19	28	42	11	2
105132	35	0.3	3	11	126	4	2
105133	20	0.4	19	25	23	33	2
105134	70	1.4	19	28	42	11	2
105135	35	0.3	3	11	126	4	2
105136	20	0.4	19	25	23	33	2
105137	70	1.4	19	28	42	11	2
105138	35	0.3	3	11	126	4	2
105139	20	0.4	19	25	23	33	2
105140	70	1.4	19	28	42	11	2
105141	35	0.3	3	11	126	4	2
105142	20	0.4	19	25	23	33	2
105143	70	1.4	19	28	42	11	2
105144	35	0.3	3	11	126	4	2
105145	20	0.4	19	25	23	33	2
105146	70	1.4	19	28	42	11	2
105147	35	0.3	3	11	126	4	2
105148	20	0.4	19	25	23	33	2
105149	70	1.4	19	28	42	11	2
105150	35	0.3	3	11	126	4	2
105151	20	0.4	19	25	23	33	2
105152	70	1.4	19	28	42	11	2
105153	35	0.3	3	11	126	4	2
105154	20	0.4	19	25	23	33	2
105155	70	1.4	19	28	42	11	2
105156	35	0.3	3	11	126	4	2
105157	20	0.4	19	25	23	33	2
105158	70	1.4	19	28	42	11	2
105159	35	0.3	3	11	126	4	2
105160	20	0.4	19	25	23	33	2
105161	70	1.4	19	28	42	11	2
105162	35	0.3	3	11	126	4	2
105163	20	0.4	19	25	23	33	2
105164	70	1.4	19	28	42	11	2
105165	35	0.3	3	11	126	4	2
105166	20	0.4	19	25	23	33	2
105167	70	1.4	19	28	42	11	2
105168	35	0.3	3	11	126	4	2
105169	20	0.4	19	25	23	33	2
105170	70	1.4	19	28	42	11	2
105171	35	0.3	3	11	126	4	2
105172	20	0.4	19	25	23	33	2
105173	70	1.4	19	28	42	11	2
105174	35	0.3	3	11	126	4	2
105175	20	0.4	19	25	23	33	2
105176	70	1.4	19	28	42	11	2
105177	35	0.3	3	11	126	4	2
105178	20	0.4	19	25	23	33	2
105179	70	1.4	19	28	42	11	2
105180	35	0.3	3	11	126	4	2
105181	20	0.4	19	25	23	33	2
105182	70	1.4	19	28	42	11	2
105183	35	0.3	3	11	126	4	2
105184	20	0.4	19	25	23	33	2
105185	70	1.4	19	28	42	11	2
105186	35	0.3	3	11	126	4	2
105187	20	0.4	19	25	23	33	2
105188	70	1.4	19	28	42	11	2
105189	35	0.3	3	11	126	4	2
105190	20	0.4	19	25	23	33	2
105191	70	1.4	19	28	42	11	2
105192	35	0.3	3	11	126	4	2
105193	20	0.4	19	25	23	33	2
105194	70	1.4	19	28	42	11	2
105195	35	0.3	3	11	126	4	2
105196	20	0.4	19	25	23	33	2
105197	70	1.4	19	28	42	11	2
105198	35	0.3	3	11	126	4	2
105199	20	0.4	19	25	23	33	2
105200	70	1.4	19	28	42	11	2
105201	35	0.3	3	11	126	4	2
105202	20	0.4	19	25	23	33	2
105203	70	1.4	19	28	42	11	2
105204	35	0.3	3	11	126	4	2
105205	20	0.4	19	25	23	33	2
105206	70	1.4	19	28	42	11	2
105207	35	0.3	3	11	126	4	2
105208	20	0.4	19	25	23	33	2
105209	70	1.4	19	28	42	11	2
105210	35	0.3	3	11	126	4	2
105211	20	0.4	19	25	23	33	2
105212	70	1.4	19	28	42	11	2
105213	35	0.3	3	11	126	4	2
105214	20	0.4	19	25	23	33	2
105215	70	1.4	19	28	42	11	2
105216	35	0.3	3	11	126	4	2
105217	20	0.4	19	25	23	33	2
105218	70	1.4	19	28	42	11	2
105219	35	0.3	3	11	126	4	2
105220	20	0.4	19	25	23	33	2
105221	70	1.4	19	28	42	11	2
105222	35	0.3	3	11	126	4	2
105223	20	0.4	19	25	23	33	2
105224	70	1.4	19	28	42		

9600N

9600N

KN-91-01(1394.0)



1400ELEV

1400ELEV

1350ELEV

1350ELEV

1300ELEV

1300ELEV

1250ELEV

1250ELEV

**LEGEND**

**Major Units:**

- AGGLDM - Karoon Agglomerate
- GRTUFF - Green tuff
- MFUFF - Karoon Feldspar Phytic tuff
- OVRBDN - Overburden

**Minor Units:**

- agglom - agglomerate
- bleach - bleached
- br'd - brecciated
- carb - carbonate
- carbfl - carbonate flooded
- carbstr - carbonate stringers
- carbvn - carbonate vein
- cave - clay and sand filled fractures
- clysnd - clay and sand filled fractures
- drtuff - deuterite alteration
- epalt - epidote alteration
- fault - fault
- fltzn - fault zone
- qzcbvn - quartz, carbonate vein
- qzclvn - quartz, chlorite vein
- qzstr - quartz stringers
- qzvn - quartz vein
- sand - sand filled fracture
- tuff - tuff

□□□□ ZONE OF INCREASED CHARGEABILITY

**GEOLOGICAL BRANCH  
ASSESSMENT REPORT**

**22,074**

KNIPPLE LAKE PROPERTY

Vertical Section Looking North  
DDH KN-91-01  
Geology and Geochem Analyses

SECTION 9600N

SCALE: 1:500M

27 NOV 1991

4

NORANDA EXPL. PRINCE GEORGE, B.C.

9400N

KN-91-02(1392.0)

9400N

10950W

10900W

10850W

10800W

10750W

1400ELEV

1400ELEV

1350ELEV

1350ELEV

1300ELEV

1300ELEV

1250ELEV

1250ELEV

**LEGEND**

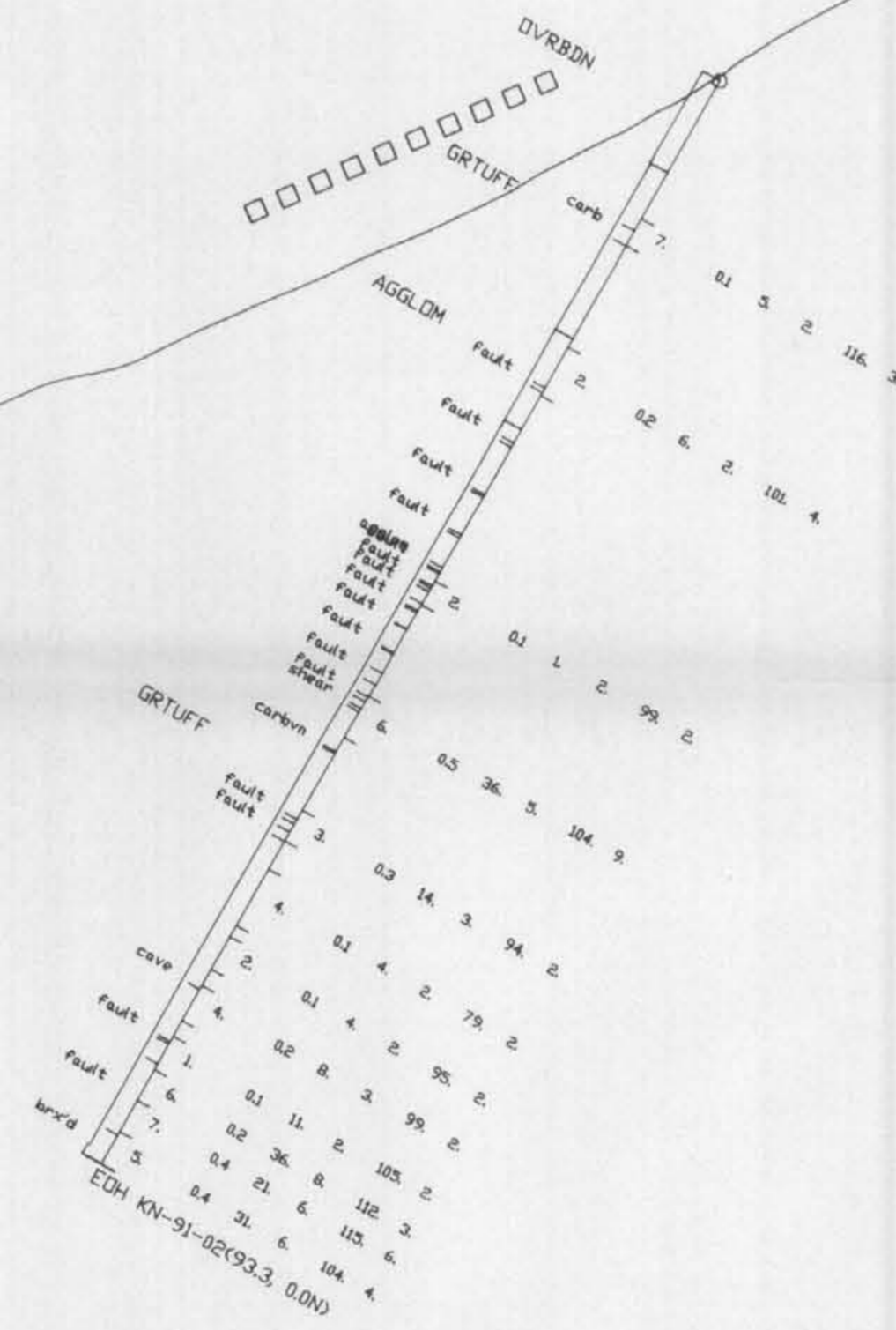
**Major Units:**

- AGGLDM - Karoon Agglomerate
- GRTUFF - Green Tuff
- MFTUFF - Karoon Feldspar Phytic Tuff
- OVRBDN - Overburden

**Minor Units:**

- agglom - agglomerate
- bleach - bleached
- brx'd - brecciated
- carb - carbonate
- carbfl - carbonate flooded
- carbst - carbonate stringers
- carbvn - carbonate vein
- cave - cave
- clysnd - clay and sand filled fractures
- ditalt - deuteric alteration
- epalt - epidote alteration
- fault - fault
- flzn - fault zone
- qzcbvn - quartz, carbonate vein
- qzclvn - quartz, chlorite vein
- qzstr - quartz stringers
- qzvn - quartz vein
- sand - sand filled fracture
- tuff - tuff

□□□□ ZONE OF INCREASED CHARGEABILITY



**GEOLOGICAL BRANCH  
ASSESSMENT REPORT**

**22,074**

KNIPPLE LAKE PROPERTY

Vertical Section Looking North  
DDH KN-91-02  
Geology and Geochem Analyses

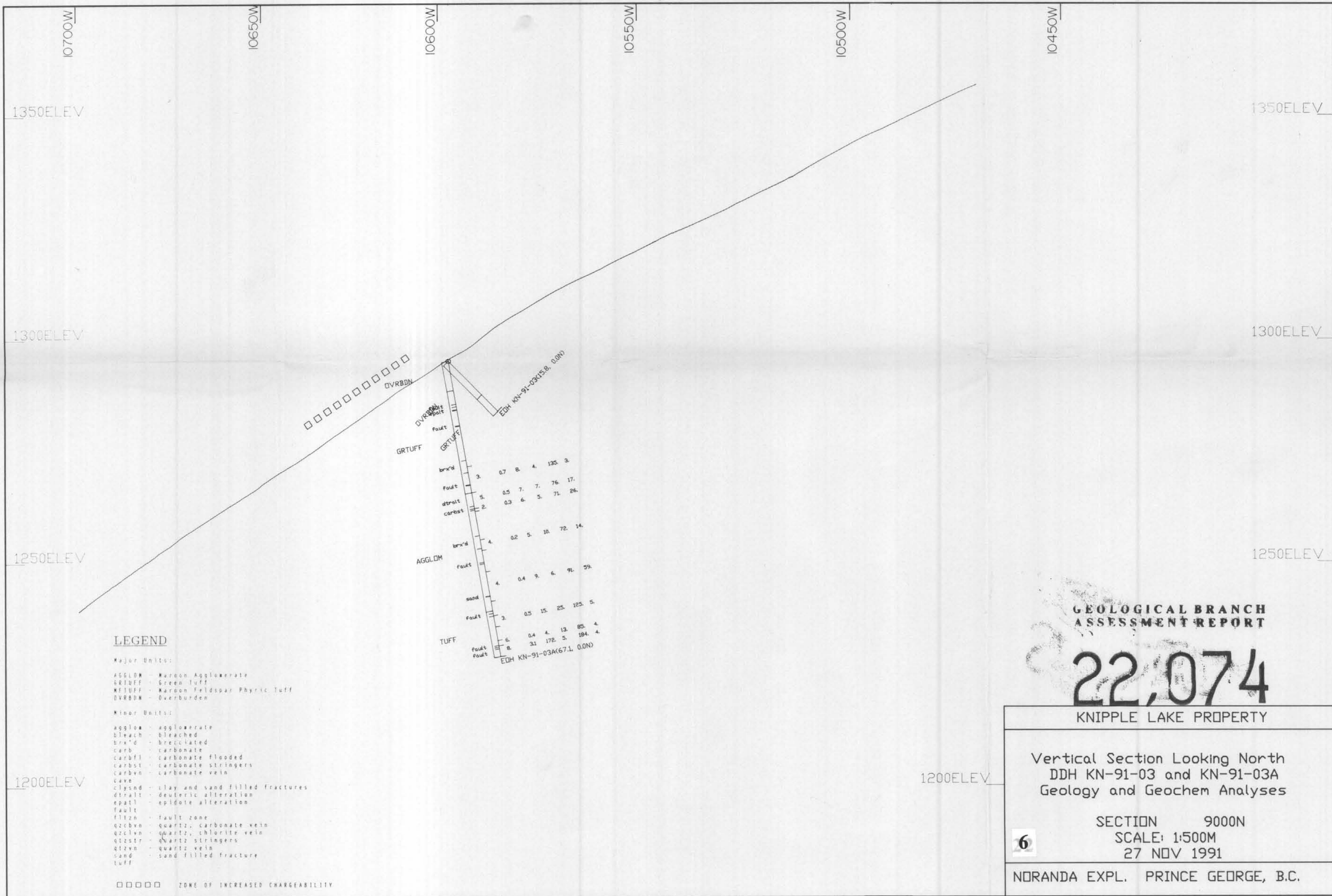
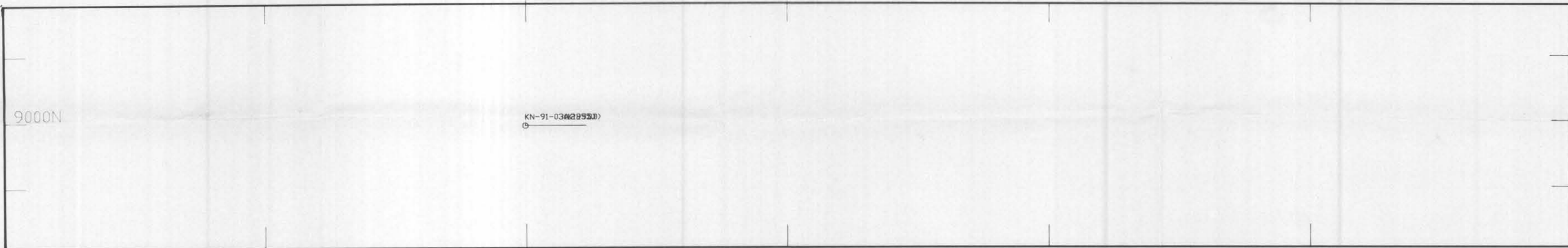
SECTION 9400N

SCALE: 1:500M

27 NOV 1991

5

NORANDA EXPL. PRINCE GEORGE, B.C.



**GEOLOGICAL BRANCH  
ASSESSMENT REPORT**

**22,074**

KNIPPLE LAKE PROPERTY

Vertical Section Looking North  
DDH KN-91-03 and KN-91-03A  
Geology and Geochem Analyses

**6**

SECTION 9000N  
SCALE: 1:500M  
27 NOV 1991

NORANDA EXPL. PRINCE GEORGE, B.C.



KN-91-04(1304.0)

9600N

11350 W

11300 W

11250 W

11200 W

11150 W

1350ELEV

1350ELEV

1300ELEV

1300ELEV

1250ELEV

1250ELEV

**LEGEND**

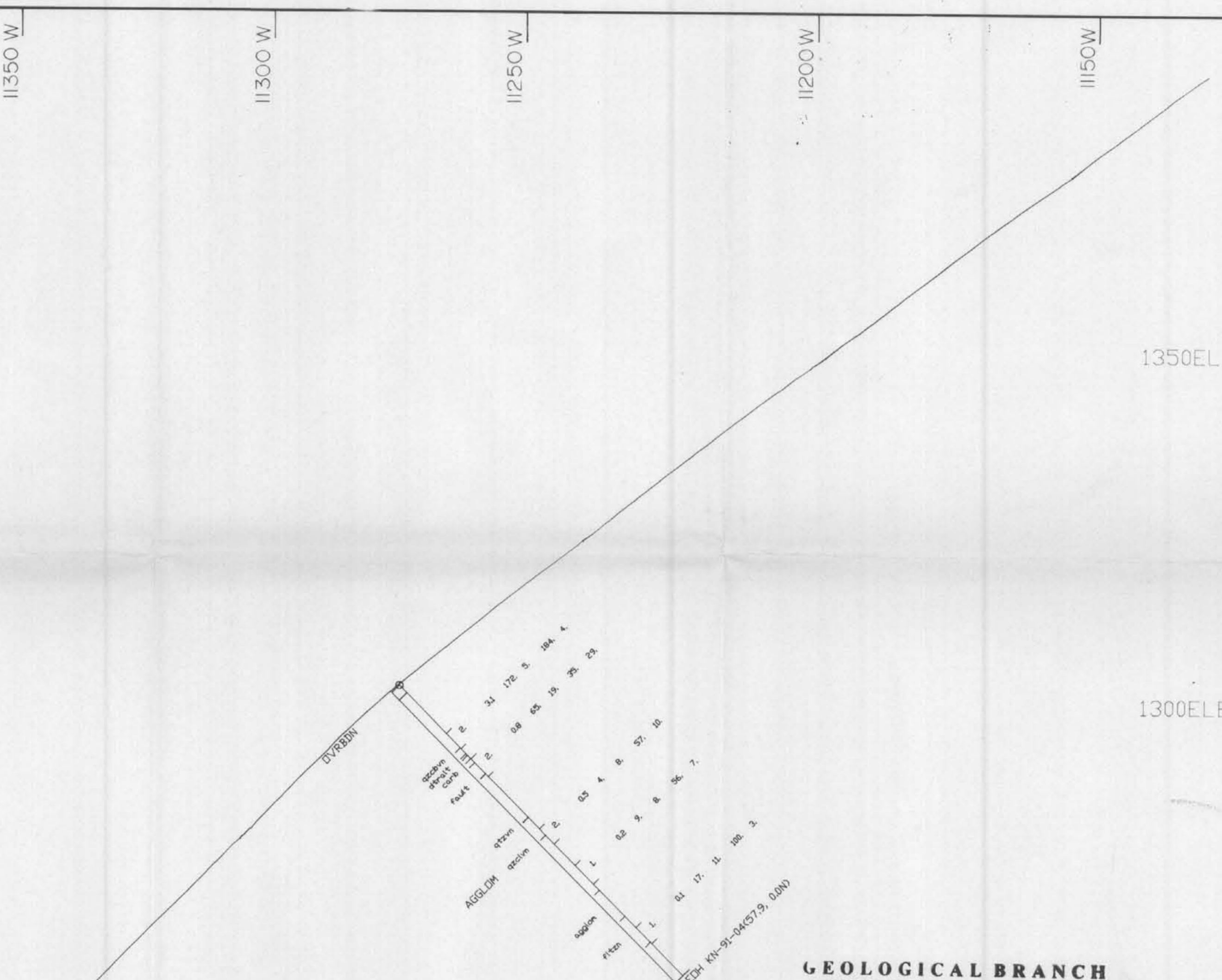
**Major Units:**

AGGLDM - Maroon Agglomerate  
GRITUFF - Green Tuff  
MFTUFF - Maroon Feldspar Phyric Tuff  
OVRBDN - Overburden

**Minor Units:**

agglom - agglomerate  
bleach - bleached  
brx'd - brecciated  
carb - carbonate  
carbfl - carbonate flooded  
carbstr - carbonate stringers  
carbvn - carbonate vein  
cav - cave  
clysnd - clay and sand filled fractures  
dtralt - deuteritic alteration  
epalt - epidote alteration  
fault - fault  
flz - fault zone  
qzcbvn - quartz, carbonate vein  
qzclvn - quartz, chlorite vein  
qzstr - quartz stringers  
qzvn - quartz vein  
sand - sand filled fracture  
tuff - tuff

□□□□ ZONE OF INCREASED CHARGEABILITY



**GEOLOGICAL BRANCH  
ASSESSMENT REPORT**

**22,074**

KNIPPLE LAKE PROPERTY

Vertical Section Looking North  
DDH KN-91-04  
Geology and Geochem Analyses

SECTION 9600N

SCALE: 1:500M

27 NOV 1991

**7**

NORANDA EXPL. PRINCE GEORGE, B.C.

9080N

KN-91-05(1143.0)

JN

11250W

11200W

11150W

11100W

11050W

**LEGEND**

Major Units:

- AGLOM - Maroon Agglomerate
- GR TUFF - Green Tuff
- RT TUFF - Maroon Felsic Rhyolite Tuff
- OVRBDM - Overburden

Minor Units:

- aglom - agglomerate
- bleach - bleached
- brack - brecciated
- carb - carbonate
- carbft - carbonate flooded
- carbst - carbonate stringers
- carbvn - carbonate vein
- cave - cave
- clysnd - clay and sand filled fractures
- dsalt - deuterite alteration
- epalt - epidote alteration
- fault - fault
- fract - fault zone
- qtzcarb - quartz, carbonate vein
- qtzchl - quartz, chlorite vein
- qtzstr - quartz stringers
- qtzvn - quartz vein
- sand - sand filled fracture
- tuff - tuff

- ZONE OF INCREASED CHARGEABILITY
- AXIS OF V.P. ANOMALY

1200ELEV

1150ELEV

1100ELEV

**GEOLOGICAL BRANCH  
ASSESSMENT REPORT**

**22,074**

KNIPPLE LAKE PROPERTY

Vertical Section Looking North  
DDH KN-91-05  
Geology and Geochem Analyses

SECTION 9080N  
SCALE: 1:500M  
27 NOV 1991

**8**

NORANDA EXPL. PRINCE GEORGE, B.C.

