REPORT ON THE KEYSTONE GROUP OF CLAIMS,

KEYSTONE MOUNTAIN AREA, NELSON MINING DIVISION,

BRITISH COLUMBIA, CANADA GEOLOGICAL BRANCH ASSESSMENT "PORT FOR

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RESOURCES CORP. RE DELAW gary, Alberta, Canada T2R 182

COVERING:

Keystone Fr. # 5137	l c.g.	Folio 4105
Cable # 5138	l c.g.	Folio 4105
Tin Brok # 5139	l c.g.	Folio 4105
General White # 5140	l c.g.	Folio 4105
Omega Fr. # 5141	l c.g.	Folio 4105
Delaware # 5136	l c.g.	Folio 41491
Del # 5137	l c.g.	Folio 41491
Clarendon # 5142	l c.g.	Folio 41556
Clarendon Fr.	l c.g.	Folio 41556

#### Record No.

84-#117-12075

3F2

Princess No. 1 #4627	l reverted c.g.	3505	(11)
Gene 1	15 units	2960	(3)
Gene 2	18 units	2961	(3)
Gene 3	5 units	2962	(3)

LOCATED :

Lat. 49°14' Long. 117°18' NTS 82F/3W Elev. 4000 ft - 5000 ft ASL

Prepared By:

P. J. Santos, P. Eng. ANGINEL RESOURCES LTD. 626-9th Ave., Castlegar, B.C., Canada

March 9, 1984

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1. SUMMARY AND CONCLUSION

The exploration program carried out on the Keystone Group of claims owned by Delaware Resources Corp. included the following:

40 Km (24.97 miles) of line cutting 2.3 Km ( 1.44 miles) of VLF-EM test profiles Km ( 0.50 mile ) of Vertical Loop EM test profiles 0.82 Km ( 1.44 miles) of Total Field magnetic tests 2.3 0.9 km (0.56 mile) of geochemical test profiles 200 hectares (494 acres) of detailed geologic mapping 30.15 Km (18.84 miles) of VLF-EM surveys Km (18.84 miles) of Total Field magnetic survey 30.15 18.5 Km (11.56 miles) of geochemical soil sampling man-days of underground sampling 13

The geologic mapping extended the known strike length of the Keystone zone for an additional 1200 feet (365 meters) to the northeast.

The combined geophysical and geochemical surveys gave positive indications that the Keystone zone probably continues for 2100 feet (650 meters) to the southwest and 2300 feet (700 meters) to the northeast right up to and beyond the border of the property.

Five coincidental geochemical and geophysical anomalies and three geochemical anomalies were found which gave credence to the

geological observation that there are multiple mineralized zones parallel to and crosscutting the main Keystone zone.

The use of the Cutler, Maine station for the VLF-EM survey to detect conductors crosscutting the Keystone trend was not effective. It is thought that conductive overburden masked the weak signals from this station. The VLF-EM data therefore did not reflect all the mineralized zones indicated by the geochemical survey and by the geologic mapping. Test profiles with the Vertical Loop EM instrument showed that it is capable of detecting the conductors crosscutting the Keystone trend.

The main Keystone zone is a prime exploration target for drilling and trenching. The underground sampling defined drill targets in the Keystone mine.

The remainder of the property is a good area to conduct an exploration program similar to that done in 1983 since most of the anomalies detected in 1983 are open.

#### 2. INTRODUCTION

Upon request of Delaware Resources Corp., owner of the Keystone Group of claims, geological, geophysical, and geochemical surveys were conducted on this mining property in 1983. The purpose of these surveys was to identify precious metal drilling and trenching targets.

### 3. LOCATION AND ACCESS

The Keystone gold-silver property is located in the Nelson Mining Division of British Columbia, Canada on the southwest side of Keystone Mountain four and one half aerial kilometers northwest of Salmo, British Columbia. It is 8 kilometers north of the settlement of Erie, 38 kilometers from the smelter in Trail, and 38 kilometers from the airport in Castlegar (see Plate 1). The property claims are plotted on NTS 82F/3W with coordinates 49°14' Lat. and 117° 18' Long. (see Plate 2).

The property is partly accessible by paved road (Highway 3) and then by an all-weather gravel road from Erie (Second Relief road) which connects with the Keystone road to the property.

The topography of the property is gentle to moderate and lies at an elevation of 4000 feet (1219 meters) to 5000 feet (1524 meters) above sea level. The property is covered with merchantable timber, in fact, the lower part of the property has been logged. The crown granted claims in the property do not have any timber rights. The

property has ample water and the company holds a water licence on Hooch Creek which runs through the property. The snow-free period in the area is from the middle of May to the middle of November.

The property is 8 kilometers from a hydro-electric power line of West Kootenay Power and Light Company.

# 4. PROPERTY DESCRIPTION AND HISTORY

The Keystone Group of Claims consists of ten crown-grants, a reverted crown-grant, and three located metric claims. Details of the property are as follows :

	(Hec.)	<u>Folio</u> No.	<u>Annya I</u> Taxes	<u>Due Date</u>
Dolormo # 5124	16 17	/1/01		
Del # 15672	19.73	41491	\$22.26	July 2
Keystone Fr. #513'	7 8.85	4105		
Cable # 5138	16.95	4105		
<b>Tin Brok # 5139</b>	16.41	4105	\$44.81	July 2
General White Fr.				·
# 5140	10.94	4105		
Omega # 5141	19.12	4105		
Clarendon # 5142 Clarendon Fr.	13.48	41556		
# 5143	13.88	41556	\$13.89	July 2
		Record No.	Annual Assessment	Due Date
Princess No. 1 # 4627 (reverted crown	20.90 grant)	3505	\$100.00	Nov. 28
Gene 1 Gene 2 Gene 3	15 units 18 units 5 units	2960 (3) 2961 (3) 2962 (3)	\$1500.00 \$1800.00 \$ 500.00	Mar. 9 Mar. 9 Mar. 14

These claims are plotted on Plate 2 (M82F/3W).

The crown-granted claims were originally staked before the turn of this century. The original Keystone group was composed of the six crown-grants; Keystone Fraction, Cable, Tin Brok, General White Fraction, and Omega. In 1900-1901, A.L. Davenport of Spokane, Washington sunk a 35-foot shaft on the Keystone vein and drove a drift 190 feet long ( No. 1 Level, see Plate 16) which was later extended to 350 feet by subsequent workers. The property was leased to various operators in 1902 to 1910. There is no record of work done in 1911 to 1931. In 1932, a small concentration mill was erected to treat the dumps from the various operations. By this time a crosscut had been driven 64 feet below Level No. 1 for 390 feet to intersect the Keystone vein and a drift was driven 70 feet southwesterly and 270 feet northeasterly (Level No. 2, see Plate 14). A raise was put up to the vein 30 feet south of where the crosscut reached it and followed the vein for 100 feet. This raise was later connected to Level No. 1 . The property was operated from 1932 to 1939 by Dufferin Gold Mines. In 1940 to 1942, the property was operated by Slocan Silver Mines who did 232 feet of drifting, 25 feet of crosscutting, and 300 feet of surface trenching. World War II put an end to the operation due to lack of miners. The property lay dormant from 1943 to 1978. In 1979, Ronald and Donald Tjader rehabilitated the portals and extended the drift (Tjader Drift) located on the Cable claim.

Between 1901 to 1981, intermittent production from the Keystone Mine amounted to 1,834 tons at 1.45 oz/ton gold and 2.93 oz/ton silver.

To the southwest and on the general strike of the Keystone vein, the Delaware and Del claims were located and crown-granted to A.L. Davenport in 1923. Numerous trenches were dug across the projected strike of the Keystone vein and a drift was driven for 50 feet (see Plate 7). No production record from these claims exists.

At the Princess No. 1 claim, the mineralized exposures were trenched. This work was done in conjunction with the work being done on the "Mjolner vein" in the Houlton crown grant by Mjolner Mines Ltd. This work was apparently done prior to 1936.

There is no record on what had been done on the Clarendon and Clarendon Fraction which was held by Slocan Silver Mines.

The Gene claims were staked in 1983 for Delaware Resources Corp. and consolidated with the rest of the property. An exploration program was carried out on the property. in 1983. Dates of Work:

P. J. Santos

April 14, 15, 21, 22, 1983 Fieldwork: 3, 9, 11, 13, 24, 27, 1983 6, 10, 13, 16, 18, 20, 1983 May June July 2, 1983 August 15, 31, 1983 Research & 10, 13, 16, 17, 1983 Report prep .: Feb. 23, 24, 25, 26, 27, 1984 1, 2, 3, 6, 7, 8, 9, 1984 Feb. Mar. David Rennie Fieldwork: May 12, 13, 17, 18, 19, 24, 1983 21, 1983 June 5, 6, 7, 8, 11, 13, 14, 15, 1983 15, 23, 31, 1983 July Aug. Sept. 1, 2, 6, 7, 8, 1983 John Murray April 21, 22, 25, 1983 May 3, 6, 9, 10, 11, 17, 18, 19, 20, 24, 25, 26, 31, 1983 June 1, 2, 3, 6, 7, 8, 9, 10, 13, 14, 15, 1983 Fieldwork: Ken Syrja 1,2, 3, 6, 7, 8, 9, 10, 13, 14, 16, 17, 20, 21, 22, 1983 1, 4, 5, 6, 7, 8, 1983 Fieldwork: June July Robert Beynon May Fieldwork: 6, 9, 10, 11, 17, 18, 19, 20, 24, 25, 26, 27, 30, 31, 1983 1, 2, 3, 6, 7, 8, 13, 14, 15, 16, 17, 20, 21, 22, 1983 4, 5, 6, 7, 8, 1983 June July

## 5. REGIONAL GEOLOGY

The general area in which the Keystone property of Delaware Resources Corp. is located lies on the eastern flank of the eugeosyncline that borders the Kootenay Arc of British Columbia on the west.

This region is underlain by the lower Cambrian units of the Kootenay Arc (Laib Formation), a succession of Permian (?) to middle Jurassic eugeosynclinal sediments and volcanics (Ymir Group and Rossland Group) and lower Cretaceous intrusives (Nelson Intrusives) as shown on Plate 3.

Recent work by geologists in southern British Columbia has resulted in the designation of the Rossland Group which now consists of the Archibald Formation, Elise Formation, Hall Formation, Beaver Mountain Formation, Rossland Volcanics, and the Ymir Formation. A correlation of these formations is shown on Plate 6.

Brief descriptions of the rock formations in the are presented below.

#### Laib Formation (Map Unit 7c, Lower Cambrian)

The Laib Formation is exposed on the southeast corner of Plate 3. It consists of an assemblage of argillaceous schists, phyllite, argillaceous quartzite, dolomite, and limestone. The limestone and dolomite units of this formation host the lead-zinc ore-bodies commonly associated with the Kootenay Arc. The Laib Formation is

intruded by granitic to granodioritic phases of the Nelson Intrusives and is at fault contact with the argillites of the Ymir Group. The Laib is one of the formations that make up the Kootenay Arc.

#### <u>Ymir Formation (Map Unit 11, Lower Jurassic)</u>

The Ymir Formation occurs in the southeast corner of Plate 3 and is in fault contact with the Laib Formation. Normally, it overlies the conformably the the Laib. It is composed of dark-colored argillaceous quartzite, argillites, and slates. This formation hosts the better known gold deposits of the Ymir Gold Camp. It is equivalent to or probably identical to the Hall Formation.

# Archibald Formation (Map Unit 12, Triassic ? and Jurassic)

The Archibald Formation is exposed on Archibald Creek and Erie Creek as shown on Plate 3. It consists of thin-bedded, grey sandstone intercallated with argillites, massive sandstone, and some banded siltstone and bedded tuff. At Erie Creek, the sandstone is pebbly and interbedded with tuff. According to H.W. Little (1960) this formation is Sinemurian (Lower Jurassic) and calls these rock units Sinemurian Beds. This formation is now classified as a member of the Rossland Group.

## Elise Formation (Map Unit 13, Lower Jurassic)

The Elise Formation occurs at Erie Creek, Hooch Creek, and Hell-Roaring Creek as shown on Plate 3. The formation essentially

consists of greenstones of andesitic to basaltic composition. In some instances the texture is coarse enough to be called gabbro. Normally, the volcanics exhibit flow breccia textures. These rocks are usually dark gray green, often black-brown or light green due to alteration, commonly serpentinization. It is frequently intruded by granite and granodiorite of the Nelson Intrusives. The Elise is the host rock of several important gold deposits in the area such as the Second Relief Mine, Porto Rico, and the Clubine-Comstock mine. The Elise Formation is now considered as the lowest subdivision of the Rossland Group. The Elise forms an anticline trending northeast south of Keystone Mountain. This formation is probably the equivalent of the Rossland Volcanics and represents the less altered portions of the Rossland Volcanics.

#### Hall Formation (Map Unit 14, Lower to Middle Jurassic)

The Hall Formation forms a north-south trending belt from Salmo to Hall Creek. The rock units of the formation are commonly grey to black and carbonaceous and when sheared graphite is formed. It consists of argillites, banded siltstones, graywacke-sandstone, quartzite, and conglomerate. Some beds are pyritic and some are calcareous. The formation is invaded by granitic to granodioritic phases of the Nelson Intrusives. Adjacent to these intrusives, hornfelsed zones are developed.

This is the most important formation in the area. It hosts the more important gold deposits such as the Arlington, Keystone, and

Silver Dollar, and Hall Creek mines. The Hall formation is equivalent if not identical to the Ymir Formation. Both formations are now classified under the Rossland Group.

# Rossland Volcanics (Map Unit 16. Middle to Upper Jurassic)

The Rossland Volcanics are exposed south of Hell-Roaring Creek on the south side of Plate 3. The formation consists of lava flows of altered andesite and basalt. This formation is commonly altered to greenstone with the development of serpentine and clivine. This formation is now part of a larger group of eugeosynclinal deposits called the Rossland Group.

# Nelson Intrusives (Map Unit 17, Lower Cretaceous)

The Nelson Intrusives occur extensively west and north of Salmo. This intrusive exhibits various phases. From Nelson northward, it is usually granitic, towards the south of Nelson it is granodioritic. Away from the main batholith body, the Nelson Formation form numerous stocks that intrude all the older rocks in the area. Sills of rhyollite and dikes of lamprophyre cuts the Hall Formation and are included in the Nelson Intrusives. The Nelson hosts some of the small but high grade gold vein deposits in the area.

## 6. LOCAL GEOLOGY

The Keystone property is underlain by rock formations belonging to the Rossland Group and the Nelson Intrusives. The members of

the Rossland Group encountered in the Keystone property are the Elise Formation (volcanics) and the Hall Formation (sediments) which are intruded by the Nelson Intrusive (granodiorite, rhyolite, and lamprophyre).

The Elise Formation consists of basalt and andesite flows intercalated with andesite tuff and agglomerate. The volcanics also occur as flow breccias. The volcanics is usually porphyritic with augite and/or plagioclase as phenocrysts. The gray-green color of the rocks indicates a low-grade type of alteration. In places the andesite tuff contain disseminated pyrite.

The Hall Formation at the Keystone property is essentially dark-colored, gray to black, carbonaceous argillites and shale which if sheared forms graphitic zones. Intercalated with the argillites are thin-bedded quartzites, sandstones, and siltstones.

The Hall Formation forms a syncline in the adjoining Arlington property to the west, while at the Keystone property, a granodiorite stock caused a doming of the sediments around the stock. An annular zone of carbonaceous hornfels is developed around this stock. A complex system of quartz veining with associated sulfides occur as sills and veins cutting the argillites.

Sills and dikes of granitic composition (rhyolite) occur with the argillite beds. Quartz veins are developed above and/or below the sills and on either or both sides of the dikes, Sulfides composed of pyrite, sphalerite, galena, tetrahedrite, and rarely

pyrrhotite occur with the quartz veins (see Fig. 4). In places the sulfides form swarms of veinlets in brecciated argillite in the absence of quartz (see Fig. 3). More details on the mineralization are found on the section on mineralization of this report.

Thin bands of pyrite are interbedded with the argillites (see Fig. 1). Disseminations of fine pyrite are also concentrated along certain thin horizons of the argillite (see Fig. 2). These pyrite occurrences are syngenetic which were deposited under reducing conditions that are usually developed in eugeosynclinal environments. Further details on this type of mineralization are presented in Section 7 of this report.

#### 7. MINERALIZATION

Sulfide mineralization in the Keystone property consists of pyrite, sphalerite, galena, minor tetrahedrite, and rare pyrrhotite. This mineralization forms three systems: as syngenetic bands in the argillites, as hydrothermal veins with quartz parallel to the bedding of the argillites, and as veins with or without quartz cutting across the bedding of the argillites (also hydrothermal). These systems have associate geophysical and geochemical trends (see Plates 14, 15, and 22).

The syngenetic mineralization is megascopically observable as thin distinct bands of pyrite parallel to the bedding of the black

argillites (see Fig. 1) and as disseminated pyrite concentrated along certain layers of the black argillites (see Fig. 2). A sample of this material assayed 48 ppb Au, 1.10 ppm Ag, 9.1 ppm As, 4.7 ppm Sb, 91 ppm Cu, 15 ppm Pb, and 36 ppm Zn which indicate an elevated content of gold and silver.

Hydrothermal mineralization consists of gold and silver bearing sulfides forming massive to semi-massive veins and disseminations usually in association with quartz (see Fig. 3 and 4). These quartz-sulfide veins occur between the bedding planes of the argillite forming superimposed multiple zones varying in thickness from a few inches to as much as twelve feet. At the adjacent Arlington mine, the Big Bullpen stope was 60 feet thick due to the multiple zones encountered.

Fracturing across the bedding (cross-fracturing) is also mineralized and the underhand stope at the Keystone mine was driven into one of these cross-fracturing. In places the sulfides, mainly pyrite act as the cementing material (mattrix) of brecciated argillite (see Fig. 3).

Gold and silver is directly associated with the sulfides, and the grade is directly proportional to the amount of sulfides present, particularly the sphalerite. The precious metals are however not confined to the sulfides and quartz but are also diffused into the wall rocks. No free gold has been identified except occasionally in boxworks left over from oxidized sulfides found

near the surface.

Surface and underground sampling was recently done on the property. Sampling was undertaken by W. Sirola, P. Eng., a consulting engineer and the assays ranged from .067 oz/ton Au to 2.93 Oz/ton Au. These assays are plotted on Plate 13. Some samples were also taken by R. Netolitzky, President of Delaware Resources Corp. and the assays are found in the Appendix of this report. A total of 46 channel samples were taken from the acessible underground workings of the Keystone Mine. The assays ranged from .CO3 oz/ton Au to .62 oz/ton Au. These assays are plotted on Plate 16 and the assay certificates are found in the Appendix of this report.

According to the Ministry of Mines reports and the smelter settlement sheets, the production from the Keystone ore-body from 1901 to 1981 amounted to 1,834 tons of 1.45 oz/ton Au and 9.93 oz/ ton Ag with values in gold, silver, and silica.

The Keystone ore-body is essentially identical to the adjacent Arlington ore-body. It is considered the up-dip extension of the Arlington ore-body. The average strike of the Keystone vein is N35°E and dips 20° NW. Mineralized cross-fractures and lamprophyre dikes cutting the Keystone vein trend to the northwest. Past mining and exploration activities has traced the mineralization for 600 feet (183 meters) along the strike. Recent detailed surface mapping has traced the Keystone vein along

strike for an additional length of 1200 feet (366 meters). Past explorations by drifting has traced for 1000 feet (304 meters) what appears to be the down-dip extension of the Keystone zone (see Plate 7).

A statistical study by G. Addie (1980) indicates that the Arlington and the Keystone ore-bodies are genetically related, both being deposited during the second phase of mineralization in the area (see Plate 12).

The large majority of the gold mines in the Trail Creek and Nelson Mining divisions are hosted by rock formations belonging to the Rossland Group (see Plate 6). In the Salmo-Ymir area, the host rocks of the gold deposits are the argillites in the Hall and Ymir formations. In the Rover Creek, Erie Creek, and Rossareas, the greenstones of the Rossland Volcanics and the land Elise Formation are the host rocks. The Rossland Group therefore is analogous to the Archean greenstones of Ontario, to the Milford Series of the Tillicum area and the Slocan sediments of the Slocan area of British Columbia. The Hall Formation, the Elise Formation, and greenstone formations of Ontario were deposited under reducing conditions in a volcano-sedimentary sequence as evidenced by the carbonaceous contents of the formations and the presence of syngenetic pyrite. Under such depositional conditions, gold was deposited syngenetically. The gold was then remobilized by hydrothermal fluids supplied by later intrusions

(such as the granodiorite stock, Map Unit 17c, Plate 3) and redeposited along fractured zones, bedding planes, and other structures with favorable porosity and permeability. This explains the the predominance of gold bearing orebodies within the formational units of the Rossland Group in areas where intrusions and shearing occur.

## 8. GEOPHYSICS

In the spring and early summer of 1983, 60.3 kilometers of geophysical surveys were conducted on the Keystone Group of claims. These surveys were comprised of 30.15 kilometers of VLF-EM (Very Low Frequency - Electromagnetic) survey and 30.15 kilometers of Total Field magnetic survey which were conducted by personnel of Interpretex Resources Ltd. under the direction of the author. The objectives of these surveys were to trace the extention of the main Keystone zone beyond its currently explored limits, to detect the presence of additional conductors parallel to the Keystone trend, and to delineate the cross-cutting mineralized structures known to occur in the property.

Prior to the above-mentioned surveys, preliminary test profiles were conducted to determine the best parameters for the surveys. These tests were conducted on lines 14a, 15, 15b, and 16 as shown on Plate 5 totaling 2.3 kilometers. These lines were cut across and over the known ore-body at the Keystone mine. These tests

involved VLF-EM using a Geonics EM-16 instrument, Vertical Loop E.M. using McPhar VHEM Electromagnetic System instruments, and Total Field magnetics using a Geometrics G-816 magnetometer. The results of these preliminary test profiles are shown on Plate 17, Plate 18, and Plate 19. The VLF data (Plate 17) and the Vertical Loop E.M. data (Plate 18) showed good response over the Keystone ore-body. However, poor coupling with the Hawaii station and masking by conductive overburden apparently prevented the strong vertical loop conductor at the "XO" positions over Line 15 and Line 15b from showing up on the VLF profiles. The low frequencies used by the VHEM Vertical Loop E.M. system will be useful in penetrating below the conductive overburden where VLF response will be masked and weak (Rockel, May 1983).

The Total Field magnetic tests (Plate 19) did not correlate directly with conductivity. It was felt however that the magnetic information on the survey area may be useful in outlining structural features.

In the subsequent electromagnetic surveys it was decided to use the VLF-EM instrument only. In addition to using the Seattle/ Hawaii stations for conductors striking northeast (Keystone Trend), the Cutler, Maine station was used instead of the VHEM Vertical Loop E.M. instrument to detect conductors crosscutting the Keystone trend. It was also decided that a base station magnetometer will be used to facilitatae the magnetic survey.

The parameters of the geophysical surveys are as follows:

Line separation	122 meters (400 feet)
Station spacing	25 meters ( 82 feet)
Horizontal Control	compass and chain, slope corrected
Base line azimuth	035°
Instrumentation:	
VLF-EM	Geonics EM-16 Cutler, Seattle, Hawaii crystals
Magnetics	Geometrics G-816 magnetometer Geometrics G-856 Memory Auto- recording Digital Base Sta tion magnetometer

The results of the geophysical surveys are shown on Plate 20, Plate 21, Plate 22, and Plate 23. The majority of the conductive zones trend azimuth 030° to 060°.

The strongest and most consistent conductive feature is the main Keystone zone. This multiple conductive zone appears to terminate south of Line 10 (outside of property) and continues northward off the northeast border of the property. The geochemical data and geologic mapping support these findings.

A strong VLF-EM response on Line 15 at 1350W is thought to be caused by a buried metal pipe. However, it is coincident with a magnetic high typical of bedrock magnetism (Rockel, July 1983) and it is also coincident with a gold soil geochemical anomaly (see Plate 14). A VLF conductor on Line 15 at 875 W coincides partly with a geochemical anomaly that trends to the northwest. This trend is not well reflected by the Fraser Filtered VLF data since this method is not effective in delineating northwest trends as was indicated by the preliminary test profiles.

A strong VLF response on Line 17 at 500 E is coincidental with a Au-Ag geochemical trend which may either be part of a zone parallel to the Keystone zone or part of a zone crosscuting the Keystone zone.

A conductive trend on Line 19 at 700 W is coincidental with the intersection of two geochemical anomalies.

Three conductive trends occur east of the baseline near Hooch Creek more or less parallel to the Keystone trend. The geochemical data, although incomplete in this area appears to follow a similar trend.

Fault trends interpreted from the geophysical data (Rockel, 1983) were not supported by the geological mapping. However, this part of the area is close to the contact of the granodiorite stock on Keystone Mountain and the the lithological change may have affected the geophysical charecteristics of the area.

In conjunction with the VLF survey, a magnetometer survey was conducted using the same grid lines. A total of 30.15 kilometers were surveyed. A repeatability of plus or minus 3 gammas was mantained. The results of the Total Field magnetic survey are plotted and contoured on Plate 23. To simplify the data presentation,

50,000 gammas were subtracted from each reading. The magnetic data do not correlate with the electromagnetic and geochemical data which means that the Keystone mineralization is not related to any significant amount of pyrrhotite. The data do show a very active magnetic environment in the northeastern part of the property in proximity to the granodiorite stock that intrudes the Hall Formation at Keystone Mountain. At 1350 W on Line 15 where there is thought to be a "buried pipe" causing a distinct VLF response, the magnetic data only indicata typical bedrock magnetism.

#### 9. GEOCHEMISTRY

The 1983 exploration program on the Keystone property included 18.5 kilometers of geochemical soil sampling. The objectives of this geochemical program are to delineate the extensions of the Keystone zone, to detect mineralized zones parallel to Keystone zone, and to delineate mineralized zones crosscutting the Keystone zone.

Prior to the soil sampling program, two profile lines (Line 14 and Line 15a, total length .9 km ) which ran across the Keystone ore zone were soil sampled to determine the best possible parameters for the survey. The A-horizon (humus) was sampled and analyzed for gold. The B-horizon was also sampled and analyzed for gold, silver, copper, lead, and zinc. The results of these tests are shown as profiles on Plate 8, Plate 9, Plate 10, and Plate 11. The statistical analyses and the assays are included in the Appendix of this report.

The gold in the A and B horizons, the silver and particularly the lead in the B-horizon are bi-modal, probably due to the presence of syngenetic and hydrothermal mineralization and the fact that the test lines went through old trenches and the nugget effect is considerable. Nevertheless, the gold silver, copper, and zinc profiles closely parallel each other. The lead profile is markedly "wavy" due to the higher standard deviation. This is thought to be due to the more severe nugget effect from relatively large lead (galena) "nuggets". The profile of the gold in the A-horizon reflects more accurately the Keystone ore zone. This is probably due to the better control that is possible when sampling the A-horizon (humus).

It was decided in the subsequent geochemical soil survey to sample only the B-horizon and assay for gold and silver only since their profiles closely parallel each other. The parameters of this soil survey are as follows:

Line spacing	122 meters (400 feet)
Sample spacing	25 meters ( 82 feet)
Survey control	Compass and chain, slope corrected
Base line azimuth	035°
Analytical method	Au & Ag by fire assay

# Statistical analyses Number of samples

By Osborne-1 Computer

Tost profiles

Test	profiles	74
Soil	survey	798

The contoured results of this soil sampling program are presented on Plate 14 and Plate 15. The statistical analyses of the geochemical data are found in the Appendix of this report.

The value of 15 ppb Au in the B-horizon is considered definitely anomalous in undisturbed areas (where no trenching or mining have been done). This value is greater than the mean plus one standard deviation of the log gold values in the profile tests and equal to the mean plus one standard deviation of the log gold values in the actual survey. This value is also twice the gold value over the Keystone ore zone.

The value of 1.00 ppm in the B-horizon is taken as anomalous for silver in undisturbed areas. This value is the mean plus two standard deviations of the log silver values in the profiling tests and equal to the mean plus one standard deviation of the actual survey. This is about twice the actual value of the silver over the Keystone ore zone.

The gold geochemical map show two sets of trends, one parallel to the Keystone zone and the other crosscutting the Keystone. One strong gold anomaly on the edge of the edge of the survey area (Line 16 and Line 18) has a trend different from the rest of the anomzlies. All the northeast trending geochemical anomalies have distinct geophysical signatures but the crosscutting geochemical trends have vague or none at all. However, this is probably due to the limitations of the VLF survey.

The most obvious geochemical anomaly is over the Keystone ore zone. Like the geophysical survey, the geochemical survey has traced this zone for 2100 feet (650 meters) southwest of the known zone and 2300 feet (700 meters) northeast of the main zone right up to and beyond the border of the property.

The strong VLF anomaly on Line 15 at 1350 W that is thought to be a "buried pipe" is part of a gold geochemical anomaly 750 meters (2460 feet) long and open to the southwest and trends parallel to the Keystone zone.

On Line 19 at 700 W and Line 20, 500 W is a gold geochemical anomaly 250 meters (820 feet) long trending parallel to the Keystone zone and open to the northeast. It appears to connect with a gold anomaly that extends all the way to Line 10 and the main baseline.

At 500 E on Line 15, there appears to be a junction of two gold anomalies, one of which trends parallel to the Keystone zone and the other crosscutting it. The geochemical data in this area is incomplete but these anomalies have associated geophysical conductive trends.

On Line 11, 200 W to Line 15, 800 W is a gold geochemical anomaly 800 meters (2624 feet) long and still open to the south. There is no distinct geophysical trend associated with this gold anomaly except at Line 15, 875 W and at Line 11, 200 W where it crosses the Kystone zone. However, this may be due to the limitation of the VLF survey.

On Line 10 at the base line to Line 17, 250 W is a gold geochemical zone that cut across the Keystone trend. This geochemical trend is 1100 meters (3608 feet) long and appears to connect with the gold geochemical anomaly at Line 19, 700 W which is parallel to the Keystone trend. This anomaly is open to the south. The profile tests with the VHEM Vertical Loop E.M. picked up a conductor on Line 15 and Line 15b (see Plate 18) that coincide with this anomaly but the subsequent VLF surveys did not pick up this conductor due to the conductive overburden.

A very pronounced gold geochemical anomaly was found on Line 17 and Line 16 at 1250 E. It is at least 250 meters long and is open to the south. There is no associated geophysical trend but this is because it is at the edge of the surveyed area.

The geochemical data east of the base line from Line 5 to Line 15 is incomplete. Nevertheless, the available data suggest several trends more or less parallel to the geophysical conductive trends.

The silver geochemical data reflect to a lesser extent the gold geochemical data. The Keystone ore zone was detected to a limited extent.

The gold anomaly on Line 17 and Line 16 has an associated

silver geochemical reflection but it is comparatively vague and narrow.

Two silver anomalies loosely match the two gold anomalies that crosscuts the Keystone trend at the southwestern part of the surveyed area.

Two silver anomalies trend parallel to the Keystone trend on Line 18, 250 W and Line 16, 500 W. There appears to be no gold geochemical anomalies associated with them but there is a weak geophysical trend associated with them.

East of the base line from Line 10 to Line 16 are several silver trends that appear to correlate with several geophysical conductive trends. The geochemical data is quite incomplete in this area to make more detailed observation than this.

#### 10. RECOMMENDATIONS

Further exploration of the property is recommended. Part of the Keystone Zone can now be drilled without further detailing. There are several underground targets that are ready for drilling without further detailing.

Detailing using closer line spacing should be done on the various geophysical and geochemical anomalies detected in 1983.

Several lines not soil sampled in 1983 should be sampled to define the geochemical anomalies better.

The northwest trending anomalies crosscutting the Keystone

trend should be surveyed with a Vertical Loop E.M. survey to delineate them further prior to trenching/drilling.

Contingent on the results of the detailing work, the delineated anomalies should be trenced and/or drilled.

The line cutting and exploration should be extended to the rest of the property.

The Keystone road should be extended to provide access to the norhtern part of the property.

Loring McIsaac	
Fieldwork:	May 17, 18, 19, 20, 24, 25, 26, 27, 30, 31, 1983 June 1, 2, 3, 6, 7, 8, 9, 10, 13, 14, 15, 16, 17, 20, 21, 22, 1983 July 4, 5, 6, 7, 1983
P <b>ster</b> Dorey	
Fieldwork:	May 17, 18, 19, 20, 24, 25, 26, 27, 30, 31, 1983
Lloyd Addie	,
Fieldwork:	May 17, 18, 19, 20, 24, 25, 26, 27, 30, 31, 1983 June 1, 2, 3, 6, 7, 8, 9, 10, 13, 14, 15, 17, 20, 21, 22, 1983 July 4, 5, 6, 7, 8, 1983
Manolito Cillo	
Drafting:	June 9, 15, 17, 18, 22, 1983 July 19, 20, 21, 22, 25, 1983 Aug. 6, 19, 20, 1983
Henry Huser	
Fieldwork:	May 3, 1983
Ed Rockel	
Fieldwork:	May 12, 13, 1983 June 8, 9, 10, 11, 12, 13, 14, 15, 16, 17, 18, 19, 20, 21, 22, 23, 24, 25, 26, 27, 29, 1983
	July 1, 3, 1983
Hallie Rockel	
Fieldwork:	May 12, 13, 1983 June 8, 9, 10, 11, 12, 13, 14, 15, 16, 17, 18, 19, 20, 21, 22, 23, 24, 25, 26, 27, 29, 1983
	July 1, 3, 1983

Secretarial:	May	17,	19, 21, 24, 27, 28, 29, 1983
	June	16,	17, 20, 21, 22, 23, 24,
	July	-20, -4, 30	5, 7, 8, 12, 13, 1983
	Sept.	2,	9, 13, 14, 15, 28, 29, 1983

Wages:

Geologists \$ Drafting and secretarial	13,600.00 2,557.50	
Line cutting	6 <b>,791.</b> 50	
Trenching	24 <b>7.</b> 50	
Soil sampling, 37 man-days @ \$90.00/day	4,320.00	
Underground sampling	2,070.00	
Assays		
Acme Analytical \$5,481.32   Terra-Min Lab. 719.20   \$6,200.32	6,200.32	
Geophysics		
Interpretex Resources Ltd.	9,958.07	
Truck Rental (Includes gas, oil diesel fuel)	3,250.00	
Equipment and supplies		
Field equipment (includes underground safety equip.) 2776.87 Disposable supplies <u>459.15</u> \$3236.00	3,236.02	
Accommodations		
Groceries for camp	717.50	

Maxine Lukey

Freight and Mail	283.25
B.C. Telephone	118.64
Printing, Reproduction, Color proces	sing 857.66
Preparation of Reports	2,043.70
M <b>isc</b> ellaneous	<b>299.</b> 50
TOTAL	\$56,551.16

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Written and Submitted By:

1. Aant

Perfecto J. Santos, P. Eng. March 9, 1984

Allen, G. 1939

1941

1982

Cominco Ltd.

Doelle, H. E.

1938

1932

1970

Little, H. W.

1960

1965

1982

Addie, G.G.	- Gold in the Nelson area; Ministry of
1980	Energy, Mines, & Pet. Res. research paper, 22pp

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Sirola, W. M. 1983	- Report on the property of Delaware Res- ources Corp. Keystone gold-silver pro- ject, Salmo area, Nelson M.D.; Engineer's report, 19 pp.
Walker, J. F. . 1934	- Geology and mineral deposits of the Salmo map area, B.C.; GSC Memoir 172, p.22

,
13. STATEMENT OF QUALIFICATIONS

I, Perfecto J. Santos, hereby certify:

That I am a Consulting Geological Engineer residing at 626 - 9th Avenue, Castlegar, B.C., Canada,

That I am a registered Professional Engineer in the Province of British Columbia, Canada,

That I am a graduate of the College of Engineering, University of the Philippines with a Bachelor of Science degree in Mining Engineering (Geology Option),

That I have been practicing my profession for the past twenty three years,

That I personally supervized the work as this described in this report on the Keystone Group of Claims owned by Delaware Resources Corp. of Calgary, Alberta, Canada.

P.J. Santos, P. Eng.

Consulting Geological Engineer

			PAGE
14.	APPENDIX	· · · · · · · · · · · · · · · · · · ·	34 - 90
	<b>(</b> a)	Maps and Illustrations	34 - 47
	<b>(</b> b)	Assays of Test Profiles Soil Samples	48 - 52
	(c)	Statistical Analyses of Profile Samples	53 <b>-</b> 60
	(d)	Assays of Soil Sampling Program	61 - 82
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	(f)	Description and Assays of Delaware Samples	86 - 87
	(g)	Assays of Underground Samples	88 - 90

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r		·		<b>*</b>			38	
E	RA, PERIOD, EPOCH	N	NELSON (WEST H	ALF)	NELSON (EAST HALF) RICE, 1941	ROVER-EN	RIE CREEKS	GOLD DEPOSITS (EXAMPLES)
		WEST PART	SOUTHEAST PART	NORTHEAST PART	WEST PART			
	PLIOCENE		2					
R Y	MIOCENE		CONGLOMERATE					
TA	OLIGOCENE					2		
2	EOCENE							
μ	PALEOCENE	2						
	UPPER CRETACEOUS	SOPHIE MOUNTAIN FORMATION	-					
	LOWER CRETACEOUS	· · · · · · · · · · · · · · · · · · ·				NELS INTRU	ON ISIVES	ROZAN BLACKCOCK, ROANOAKE, WII COX, DAYRBEAK
ZOIC.	UPPER JURASSIC		??			ROSSLA	ND GROUP	GRANITE - POORMAN ATHABASCA PATTI(Norcross Res.)
MESO	MIDDLE JURASSIC		HALL FORMATION				[]	CONVOR (Waybo Res.) ROZAN (Patrick Res.) ROOT (Noramex Min.) HUNGARY MAN (Waybo)
	LOWER JURASSIC	ROSSLAND F.M.	ROSSLAND F.M. SINEMURIAN BEDS	(ROSSLAND FORMATION TO NORTHWEST)		HALL-YMIR	?	YMIR MINE KEYSTONE (Delaware) SILVER DOLLAR ARLINGTON (Erie Gold)
	TRIASSIC		YMIR GROUP	SLOCAN GROUP ? KASLO GROUP	SLOCAN GROUP ? KASLO GROUP ??	(UNIT B)	ROSSLAND FORMATION (ELISE - BEAVER	CLUBINE COMSTOCK TILLICUM AREA SECOND RELIEF MAY & JENNIE PERRIER ROSSLAND GOLD-
	PERMIAN	?	?				MOUNTAIN FORMATION)	CAMP BEE (Silver Dart Min.) Bulldog (Silver Dart Min.)
ZOIC	PENNSYLVANIAN	MOUNT ROBERTS FORMATION			MILFORD GROUP			
NLÆ0.	MISSISSIPPIAN	· · · · · · · · · · · · · · · · · · ·		,	??		??	
ДЧ	DEVONIAN					<u> </u>		
	PLATE 6 Correlati Creeks	ion of Formations Area. ( modified fro	and Golds Deposits om H.W. Little's C	in The Nelson-Ca SSC memoir 308	stlegar Area and (	other area	as includin	g The Rover-Erie

To Accompany report by P.J. SANTOS P.Eng.



RJ. Sontos , P. Eng. Fcb. 1984



BAg in B-Morizon

PLATE 9 Au E Ag Geochemical Soil Profiles Line 14

P.J. Santos, P.Eng Feb. 1984



PLATE 10 Cu-Pb-Zn Geochemical Soil Profiles Line 15(q)

> P.J. Santos P.Eng Feb. 1984



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PLATE 11 Cu-Pb-Zn Geochemical Soil Profiles Line 14



"Gold in the Nelson Area"



PLATE 12 Statistical correlation of the Arlington and Keystone properties using production data.



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NOD BY. DATE. SA JRAA, P.Eng. III	2400 (1'=200') PLATEND. 13

-



### Fig. 1

Photograph of syngenetic pyrite in black argillites of the Hall Formation forming distinct interbeds. Shown also hydrothermal quartz veins with pyrite cutting the argillite.





Photograph of black argillites (Hall Formation) containing syngenetic pyrite disseminated preferentially along certain horizons.









Photograph of typical high grade ore consisting of quartz vein material with pyrite, galena, sphalerite, and tetrahedrite which is gold and silver bearing.



#### Fig. 5

Photograph of the Keystone ore-body exposed in a trench consisting of superimposed parallel sills of sulfide-bearing quartz in argillites and quartzite.



#### Fig. 6

Photograph of a piece of argillite that depicts in miniature two systems of mineralization, one following the bedding, the other cross-cutting the bedding of the argillite.





Job # 83-118

Delaware Resources

Client Project DEL-1

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Sample No.	Au	Ag	Cu	Pb	Zn	
,	ppb	ppb	ppm	ppm	ppm	
B Horizon Soil						
Line 14 250 W	-2	500	22	230	450	
225	-2	550	20	187	750	
200	-2 .	360	18	310	470	
175	-2	980	22	300	630	
150	8	360	26	330	670	
125	2	390	31	310	<b>7</b> 00	
100	12	360	61	115	880	
75	4	610	25	310	470	
50	6	350	23	260	410	
25	-2	430	20	390	380	
BL	-2	520	19	340	300	
<b>25</b> E	-2	500	64	20	500	
50	-2	520	34	21	500	
75	4	670	67	23	520	
100	2	460	49	25	760	
125	10	190	230	13	1810	





Job # 83-118

Client Project DEL-1

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Sample No.	Au	
Sample 140.	na	
	PLN	
A Horizon Humus		
Line 15 A 125 W	-2	
100	-2	
75	4	
50	-2	
25	-2	
BL	-2	
<b>25</b> E	-2	
50	-2	
75	· -2	•
100	-2	
125	-2	
150	-2	
175	2	
200	6	
225	8 <sup>.</sup>	
250	-2	





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Client Project DEL-2

Date

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Sample No.	Au	Ag	Cu	Pb	Zn	
	ppb	ppb	ppm	ppm	ppm	
				<b></b> ,		
<u>B Horizon Soil</u>						
Line 15 A 125 W	4	740	17	290	230	
100	2	760	18	280	260	
75	204	830	9	173	115	
50	-2	650	16	250	230	
25	2	620	12	128	154	
BL	2	660	20	280	350	
25 E	4	720	26	180	400	
50	6	<b>37</b> 0	22	310	320	
75	-2	280	15	210	137	
100	2	280	44	188	440	
125	-2	300	27	250	390	
150	22	720	53	260	400	
175	14	420	84	174	710	
200	6	340	33	260	420	
225	14	240	29	330	540	
250	6	230	28	<b>33</b> 0	480	





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Date

Client Project DEL-2

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Sample No.	Au	
	ppb	
A Horizon Humus		
Line 14 250 W	-2	
225	-2	
200	-2	
175	-2	
150	8	
125	-2	
100	-2	
75	16	
50	-2	
25	2	
BL	8	
25 E	6	
50	4	
75	2	
100	26	
125	-2	





Job # 83-121

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Delaware Resources

Client Project DEL-3

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Sa	mple No.	Au	Ag	Cu	Pb	Zn	
		ppb	ppb	ppm	ppm	ppm	
B Horizon So	bils						
Line 14	150 E	-2	180	53	16	1560	
	175	-2	260	93	23	1750	
	200	32	430	100	21	960	
	225	4	250	60	22	790	
	250	2	320	46	24	510	
A Horizon Hu	mus						
Line 14	150 F	-2					
Dine 14	130 E	_2					
	225	4					
	250						
	230						
	200	Not	received				
	200	Not	recerved				
			•				

Project: Delaware Date run: March 2/84 Operator: DWR Units for Au and Ag are log pob. All others are log pom. B-Horizon samples

IElement				Mean	1	SD	ł
ł			- } -		• • •		1
1	log	Au	1	0.50	ł	0.54	į
ţ	log	Ag	ł	2.63	ļ	2.19	
ļ	log	Сu	1	1.51	j	0.29	i
ł	log	₽Ъ	i	2.10	1	0.50	1
!	log	Zre	ł	2.68	i	0.27	ŀ

A-Horizon samples

ł	Eleme	nt	ł	Mean	ł	SD	j
}			1		ł		ł
ł	$1 \circ \underline{\circ}$	Au	}	Ø.28	1	0.41	ł

Project: Delaware Date run: March 2/84 **D**perator: DWR Line 14 / 250 W --> 250E B-Horizon samples |Element | Mean | SD | 

 log Au
 0.38
 0.46

 log Ag
 2.61
 0.18

 log Cu
 1.60
 0.30

 log Pb
 1.90
 0.58

 log Zn
 2.82
 0.21

 Line 15 / 125W --> 250E B-Horizon samples |Element | Mean | SD | 

 l log Au I
 0.67 I
 0.61 I

 l log Ag I
 2.67 I
 0.20 I

 l log Cu I
 1.38 I
 0.25 I

 l log Pb I
 2.37 I
 0.12 I

 l log Zn I
 2.49 I
 0.22 I

 Line 14 / 250W --> 250E A-Horizon samples IElement | Mean | SD | |-----|-----|-----| 1 log Au 1 0.36 1 0.46 1 Line 15 / 125W --> 250E A-Horizon samples |Element | Mean | SD | |-----|-----|------| 1 log Au 1 0.15 1 0.30 1

Units for Au and Ag are log pob. All others are log pom.

HISTOGRAM FOR ELEMENT log Au (A-Horizon) PROJECT: Delaware OPERATOR: DWR; DATE March 2/84

CELL NO. OF SAMPLES CUM. FR. %

-0.20	-	0.00	Ø	0.00
0.00	-	0.20	22	61.11
Ø.20	-	Ø.40	Э	69.44
0.40		0.60	Ø	69.44
0.60	-	0.80	6	86.11
0.80		i.00	З	94.44
1.00		1.20	0	94.44
1.20		1.40	i	97.22
1.42		1.60	1	100.00

Histogram for log gold - DELAWARE PROJECT

FREQ CELL .00-.20 22 \* .20-. 40 3 \*\*\* **.** 4121-- 60 Ø .60-. 80 5 \*\*\*\*\* .80- 1.00 3 \*\*\* 1.00- 1.20 Ø 1.20- 1.40 1 \* 1.40- 1.60 i \*

HISTOGRAM FOR ELEMENT log Au (B-Horizon) PROJECT: Delaware OPERATOR: DWR DATE March 2/84

CELL		NO. OF	SAMPLES	CUM.	FR. X	
*******	******	******	******	******	*****	¥
a or				-		
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0.00 -	0.25	13	;	35	. 14	
0.25 -	0.50	7	,	54	.05	
0.50 -	Ø.75		i	67	.57	
0.75 -	1.00	5	i	81	.08	
1.00 -	1.25	4	•	91	.89	
1.25 -	1.50	1		94	.59	
1.50 -	1.75	1		97	.30	
1.75 -	2.00	0	ł	97	.30	
2.00 -	2.25	Q	ł	97	.30	
2.25 -	2.50	1		100	. 2121	

Histogram for loggold (B-Horizon) - DELAWARE PROJECT

CELL FREQ

.00-	.25	13	********
.25-	. 50	7	****
.50-	.75	5	****
.75-	i.00	5	****
1.00-	1.25	4	****
1.25-	1.50	1	*
1.50-	1.75	1	×
1.75-	2.00	Ø	
2.00-	2.25	Ø	
2.25-	2.50	1	×

HISTOGRAM FOR ELEMENT log Ag (B-Horizon) PROJECT: Delaware DPERATOR: DWR DATE March 2/84

CELL NO. OF SAMPLES CUM. FR. %

2.17		2.26	0	0.00
2.26	-	2.35	2	5.41
2.35		2.44	4	16.22
2.44	-	2.53	4	27.03
2.53	-	2.62	7	45.95
2.62	-	2.71	6	62.16
2.71		2.80	5	75.68
2.80	-	2.89	7	94.59
2.89	-	2.98	1	97.30
2.98	-	3.07	1	100.00

Histogram for log silver (B-Horizon) - DELWARE PROJECT

CELL FREQ 2.17~ 2.26 Ø 2.26~ 2.35 2 \*\*

c.co-	န္းမှသံမ	<u> </u>	**
2.35~	2.44	4	****
2.44-	2.53	4	***
2.53~	2.62	7	******
2.62~	2.71	6	*****
2.71-	2.80	5	****
2.80-	2.89	7	******
2.89~	2.98	1	*
2.98~	3.07	1	¥

HISTOGRAM FOR ELEMENT log Cu (B-Horizon) PROJECT: Delaware OPERATOR: DWR DATE March 2/84

CELL NO. OF SAMPLES CUM. FR. %

0.80 -	0.95	2	0.00
0.95 -	1.10	2	5.41
1.10 -	1.25	3	13.51
1.25 -	1.40	11	43.24
1.40 -	1.55	8	64.86
1.55 -	1.70	3	72.97
1.70 -	1.85	6	89.19
1.85 -	2.00	3	97.30
2.00 -	2.15	Ø	97.30
2.15 -	2.30	Ø	97.30
2.30 -	2.45	1	100.00

Histogram for log copper (B-Horizon) - DELAWARE PROJECT

CEL	L F	REQ	
=========	=====	===	
.80-	.95	21	
.95-	1.10	2	**
1.10-	1.25	З	***
1.25-	1.40	11	*****
1.40-	1.55	8	****
1.55-	1.70	3	***
1.70-	1.85	6	****
1.85-	2.00	З	***
2.00-	2.15	Ø	
2.15-	2.30	Ø	
2.30-	2.45	1	*

HISTOGRAM FOR ELEMENT log Pb (B-Horizon) PROJECT: Delaware OPERATOR: DWR DATE March 2/84

NO. OF SAMPLES CUM. FR. % CELL \*\*\*\*\* 1.01 -1.11 Ø 0.00 1.11 -1.21 2 5.41 1.21 -1.31 1 8.11 1.31 -1.41 7 27.03 1.41 -1.51 Ø 27.03 1.51 -1.61 Ø 27.03 1.61 -1.71 Ø 27.03 1.71 -2 1.81 27.03 1.81 -1.91 0 27.03 1.91 -2.01 Ø 27.03 2.01 -2.11 3 32.43 2.11 -2.21 Ø 32.43 2.21 -2.31 5 45.95 2.31 -2.41 4 56.76 2.41 -2.51 11 86.49 2.51 -2.61 5 100.00

Histogram for log lead (B-Horizon) - DELAWARE PROJECT

CELL FREQ 1.01- 1.11 Ø 1.11- 1.21 2 \*\* 1.21- 1.31 1 \* 1.31- 1.41 7 \*\*\*\*\*\* 1.41- 1.51 Ω1 1.51- 1.61 Ø 1.61- 1.71 Ø 1.71- 1.81 Ū. 1.91 1.81-0 1.91- 2.01 Ø 2.01- 2.11 2 \*\* 2.11- 2.21 Ø 2.21- 2.31 5 \*\*\*\*\* 2.31- 2.41 4 **\*\*\***\* 2.41- 2.51 11 \*\*\*\*\*\*\*\*\*\* 2.51- 2.61 5 \*\*\*\*\*

HISTOGRAM FOR ELEMENT log Zn (B-Horizon) PROJECT: Delaware OPERATOR: DWR DATE March 2/84

CELL NO. OF SAMPLES CUM. FR. % \*\*\*\*\*\*\*\*\*\*\*\*\*\*\*\*\* 1.93 -2.06 12 0.00 2.06 -2.19 З 8.11 2.19 -2.32 Ø 8.11 16.22 2.32 -2.45 З 2.45 -2.58 4 27.03 2.58 -2.71 13 62.16 2.71 -2.84 4 72.97 2.84 -2.97 6 89.19 1 2.97 -3.10 91.89 3.10 -3.23 94.59 1 3.23 -3.36 2 100.00

Histogram for log zinc (B-Horizon) - DELAWARE PROJECT

CELL FREQ 1.93- 2.06 ŧ2۱ 2.06- 2.19 3 \*\*\* 2.19~ 2.32 Ø 2.32~ 2.45 \*\*\* E 2.45~ 2.58 4 \*\*\*\* 2.58- 2.71 13 \*\*\*\*\*\*\*\*\* 2.71- 2.84 4 **\*\*\***\* 2.84- 2.97 6 \*\*\*\*\* 2.97~ 3.10 1 \* 3.10- 3.23 1 \* 3.23- 3.36 2 \*\*

ACME ANALYTICAL LABORATORIES LTD. 852 E. HASTINGS, VANCOUVER B.C. PH:253-3158 TELEX:04-53124 DATE RECEIVED JULY 15 1983 DATE REPORTS MAILED -

#### ASSAY CERTIFICATE GEOCHEMICAL

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 $J_{\rm u}$   $2^2$ 

A .500 GH SAMPLE IS DIGESTED WITH 3 NL OF 3:1:3 HCL TO HNO3 TO H2O AT 90 DEG.C. FOR 1 HOUR. THE SAMPLE IS DILUTED TO 10 MLS WITH WATER. ELEMENTS ANALYSED BY AA : AG, SAMPLE TYPE : SUIL - DRIED AT 60 DEG C., -80 MESH. AU .. - INCLUDING PD, PT 10 GN, FIRE ASSAY CONCENTRATION, HNO3 LEACH OFF AG.

AQUA REGIA DIGESTION, GRAPHITE AA ANALYSIS.

ASSAYER	A CALLE DEAN T	OYE, CERTIFIED	B.C. ASSAYER
ANGINEL RESOU	RCES / FILE # 83-1	PROJECT:D	ELEWARE PAGE# 1
	SAMPLE	AG AU★★ PPM PPB	
	10 750W 10 725W 10 700W 10 675W 10 650W	.7 3 .5 4 .5 45 .6 5 .4 6	
	io 625W 10 600W 10 575W 10 550W 10 525W	$ \begin{array}{cccc} .7 & 14 \\ .6 & 7 \\ .8 & 8 \\ .9 & 11 \\ .5 & 4 \end{array} $	
	10 25E 10 50E 10 75E 10 100E 10 125E	.6 95 .4 1 .6 18 .6 3 .4 3	
	10 150E 10 175E 10 200E 10 225E 10 250E	.6 4 .4 3 .4 2 .6 6 .3 4	
	10 275E 10 300E 10 325E 10 350E 10 374E	$ \begin{array}{cccccccccccccccccccccccccccccccccccc$	
	10 400E 10 425E 10 450E 10 475E 10 500E	$ \begin{array}{cccccccccccccccccccccccccccccccccccc$	
	10 525E 10 550E 10 575E 10 600E 10 625E	.5       13         .4       24         .4       14         .6       15         .7       32	
	10 650E 10 675E	.4 11 .8 23	

SAMPLE	AG PPM	AU×¥ PPB
10 700E	.5	64
10 750E	1.3	7
10 775E	.5	15
10 800E	.9	5
10 825E	.9	7
10 850E	1.1	6
10 875E	1.2	7
10 900E	.6	15
10 950E	.7	5
10 975E	.7	13
10 1000E 10 1025E 10 1050E 10 1075E 10 1100E	.5 .6 .6 1.1 .9	13 6 8 10
10 1125E	- 7	57
10 1150E	- 6	4
10 1175E	- 7	9
10 1200E	- 4	4
10 1209E	- 7	4
11 725W 11 700W 11 675W 11 650W 11 625W	.5 .6 .7 .8	245 255
11 600W	1.2	3
11 575W	1.1	26
11 550W	.1	8
11 525W	.2	4
11 500W	1.2	9
11 475W	1.4	5
11 450W	1.0	4
11 425W	2.1	3
11 400W	.5	19
11 375W	.9	, 3
11 350W	.8	9
11 325W	.7	4

SAMPLE	AG PPM	AU** FPB
11 300W	- 4	6
11 275W	- 4	2
11 250W	- 7	32
11 225W	- 4	4
11 200W	- 4	90
11 175W 11 150W 11 125W 11 100W 11 75W	. 4 . 2 . 4 . 2 . 3	70 2 1 39
11 50W 11 25W 11 0W 12 725W 12 700W	. 2 . 4 . 8 . 3 . 4	71 3 5 4
12 675W	.5	13
12 650W	.7	16
12 625W	.7	39
12 600W	.6	12
12 575W	1.1	20
12 550W	.4	24
12 525W	.4	42
12 500W	2.4	8
12 475W	2.4	8
12 450W	2.4	8
12 425W	2.2	32
12 400W	.6	210
12 375W	.3	19
12 350W	.1	13
12 325W	.2	11
12 300W	.3	15
12 275W	.1	11
12 250W	.1	10
12 225W	.2	10
12 200W	.4	11
12 175W	.4	7
12 150W	.2	7

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SAMPLE	AG PPM	AU∗≁ PPB
12 125W 12 100W 12 75W 12 50W 12 50W	. 7 . 5 . 5 . 7 . 6	12 10 11 13 11
12 OW 13 725W 13 700W 13 650W 13 625W	.7 .5 .5 .4 .3	7 1 1 1 2
13 600W 13 575W 13 550W 13 525W 13 500W	1.2 .7 .6 .7 .5	5 25 20 3
13 475W 13 450W 13 425W 13 400W 13 375W	- 8 - 5 - 5 - 6 - 4	24453
13 350W 13 325W 13 300W 13 275W 13 250W	.3 .6 .5 .4 .5	31 3 2 2 29
13 225W 13 200W 13 175W 13 150W 13 125W	- 6 - 8 - 8 - 4 - 5	33327
13 100W 13 75W 13 50W 13 25W 13 600E	- 8 - 7 - 8 - 4 - 6	1 1 1 4
13 625E 13 650E	.6 1.8	4 8

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SAMPLE	AG PPM	AU★★ ₽₽₽
13 675E	.8	5
13 700E	.7	4
13 725E	3.4	2
13 750E	1.2	10
13 775E	.7	5
13 800E	1.2	4
13 825E	.6	2
13 850E	.7	4
13 875E	1.2	10
13 900E	.8	9
13 925E	.7	10
13 950E	1.4	4
13 975E	1.1	4
13 1000E	1.4	1
13 <del>1050E</del> /025 <i>E</i>	.8	5
13 <del>1075E</del> /05%E	1 . 4	3
13 <del>1100E</del> /075E	. 8	3
13 <del>1125E</del> //00E	. 7	5
13 <del>1150E</del> //25E	. 7	5
13 <del>1175E</del> //50E	. 9	1
13 <del>-1200E</del> // <i>75E</i>	.5	1
13 <del>-1225E</del> /2 <i>00E</i>	1.0	1
13 <del>-1250E</del> /225E	.8	3
14 750W	.7	2
14 725W	.9	3
14 700W	.7	25
14 675W	.5	16
14 650W	2.2	5
14 625W	.9	2
14 600W	.7	3
14 575W 14 550W 14 525W 14 500W 14 500W	.8 .7 .7 .8 .5	3 2 9 3 2 9 3 2
14 450W	.7	1
14 425W	.5	2

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SAMPLE	AG PPM	AU★★ PPB
14 400W	.4	3
14 375W	.1	25
14 350W	.6	4
14 325W	.3	3
14 300W	.4	7
14 275W	.3	15
14 275E	.5	8
14 300E	.2	14
14 325E	.3	24
14 330E	.4	25
14 375E	.5	9
14 400E	.3	21
14 425E	.3	19
14 450E	.4	41
14 475E	.3	13
14 500E	.4	3
14 525E	.4	10
14 550E	.5	5
14 575E	.1	10
14 600E	.9	9
14 625E	.3	16
14 650E	.2	58
15 1700W	.8	20
15 1675W	1.6	6
15 1650W	.8	5
15 1625W 15 1600W 15 1575W 15 1550W 15 1525W	1.0 .6 .5 .5 .6	5 8 6 13
15 1500W	.4	3
15 1475W	1.3	13
15 1450W	.9	6
15 1425W	.8	31
15 1400W	.4	62
15 1375W	. 8	24
15 1350W	. 7	6

SAMPLE	AG FPM	AU** PPB
15 1325W 15 1300W 15 1275W 15 1250W 15 1225W	. 4 . 8 . 4 . 5	12 5 7 13 9
15 1200W 15 1175W 15 1150W 15 1125W 15 1125W	.3 .5 .4 .3	3 8 4 3 6
15 1075W	.3	8
15 1050W	.4	19
15 1025W	.4	4
15 1000W	.5	5
15 975W	.6	2
15 950W	.6	2
15 925W	.8	5
15 900W	.4	5
15 875W	.3	5
15 850W	.6	9
15 825W	.5	13
15 800W	.6	8
15 775W	.4	3
15 750W	.4	15
15 725W	.6	3
15 700W	- 8	10
15 675W	- 6	7
15 650W	1 - 2	3
15 625W	- 4	6
15 600W	- 8	8
15 575W 15 550W 15 525W 15 500W 15 475W	1.1 .8 .5 .4	5 3 3 10
15 450W	- 4	1
15 425W	- 6	3

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SAMFLE	AG FFM	AU+* PPB
15 400W	.8	4
15 375W	.6	4
15 350W	.7	16
15 325W	.7	3
15 300W	.7	3
15 275W 15 250W 15 225W 15 200W 15 175W	.7 .6 .4 .6 .7	7 3 2 4
15 150W	-8	1
15 125W	-6	1
15 100W	1.3	1
15 75W	-9	1
15 50W	-7	1
15 25W	.7	10
15 25E	.8	26
15 50E	.7	41
15 150E	.4	61
15 175E	.8	4
15 200E	.4	4
15 225E	.5	16
15 250E	.5	3
15 275E	.3	6
15 300E	.3	4
15 325E	- 5	14
15 350E	- 7	1
15 375E	- 4	4
15 400E	- 4	5
15 425E	- 5	19
15 450E	.4	6
15 475E	.7	12
15 500E	.7	4
15 525E	.6	33
15 550E	.6	24
15 575E	.5	3
15 600E	.6	40
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SAMPLE	AG PPM	AU×× PP9
15 625E	.5	7
15 650E	.5	14
15 675E	.7	11
15 700E	.4	4
16 1525W	1.0	4
16 1500W	.9	2
16 1475W	1.6	5
16 1450W	.9	3
16 1425W	1.2	4
16 1425W	1.4	5
16 1375W	-9	32
16 1350W	1-5	30
16 1325W	1.2	20
16 1300W	1.0	6
16 1300W	.6	5
16 1250W	- 7	2
16 1225W	- 7	4
16 1200W	- 8	13
16 1175W	- 9	5
16 1175W	- 6	2
16 1125W 16 1100W 16 1075W 16 1075W 16 1025W	- 8 - 6 - 7 - 6 - 5	5 19 6 2
16 1000W	.6	2
16 975W	.8	2
16 925W	.6	4
16 900W	.7	4
16 875W	.5	4
16 850W 16 825W 16 800W 16 775W 16 750W	.7 1.2 .5 .4 .4	N 15 15 4 13
16 725W	.5	11
16 700W	.2	4

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SAMFLE	AG PPM	AU¥¥ PPB
16 675W	. 4	5
16 650W	. 5	4
16 625W	. 4	7
16 600W	. 5	3
16 575W	. 9	4
16 550W	1.6	3
16 525W	2.5	3
16 500W	1.5	2
16 475W	1.8	5
16 450W	.9	3
16 425W	- 7	120
16 400W	- 6	1
16 375W	- 6	3
16 350W	- 5	2
16 325W	- 4	3
16 300W	.7	1
16 275W	.6	2
16 250W	.5	1
16 225W	.7	3
16 200W	.7	4
16 175W 16 150W 16 125W 16 100W 16 75W	- 6 - 6 - 9 - 8 - 8	3 5 3 12
16 50W	.6	1
16 25W	.5	3
16 0W	.6	5
16 25E	1.2	2
16 50E	.7	4
16 75E	- 4	2
16 100E	- 8	3
16 125E	- 3	2
16 150E	- 4	4
16 175E	- 6	5
16 200E	.5	2
16 225E	.4	1

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SAMPLE	AG PPM	АU** РРВ
16 250E	. 6	9
16 275E	. 2	7
16 300E	. 6	15
16 325E	. 6	22
16 350E	. 5	11
16 375E 16 400E 16 425E 16 450E 16 475E	.5 .7 .6 .7 .6	2 23 12 14
16 500E	. 7	14
16 525E	. 6	3
16 550E	. 4	7
16 575E	. 6	10
16 600E	. 7	5
16 625E	- 6	3
16 650E	2.1	5
16 675E	- 8	4
16 700E	- 6	9
16 725E	- 8	4
16 750E	- 6	18
16 775E	- 7	21
16 800E	- 8	3
16 825E	- 5	12
16 850E	- 8	4
16 875E	.9	2
16 900E	.7	7
16 925E	.9	4
16 950E	2.4	7
16 975E	.7	3
16 1000E	.9	17
16 1025E	1.2	5
16 1050E	1.0	5
16 1075E	1.0	6
16 1100E	.6	9
16 1125E	.9	4
16 1150E	.8	5

ANGINEL	RESOURCES	FILE # 83-1209	PROJ	ECT:DELEWARE	PAGE#	12
	SAMPL	E	AG FPM	AU** PPB		
	16 11	75E	.5	22		
	16 12	00E	. 6	17		
	16 12	25E	1.4	470		
	16 12	50E	1.3	20		
	17 13	50W	.5	6		
	17 <del>13</del>	<del>25W</del> /275 VV	. 4	2		
	17 <del>13</del>	<del>00W</del> 1250	.5	360		
	17 <del>12</del>	<del>75W</del> /275	.6	64		
	17 <del>12</del>	<del>50W</del> /200	.5	56		
	17 <del>12</del>	<del>25W</del> //75	.6	20		
	17 <del>12</del>	<del>00W</del> //50	.7	34		
	17 <del>11</del>	<del>75W</del> 1125	.6	9		
	17 <del>11</del>	50W //00	1.1	7		
	17 <del>11</del>	<del>25W</del> 1075	.8	1		
	17 <del>11</del>	00W / 050	.5	4		
	17 <del>10</del>	75W /025	.6	3		
	17 <del>10</del>	50W /000	. 7	4		
	17 -10	<del>00W</del> 975	.8	10		
	17 97	SW 950	.5	3		
	17 <del>95</del>	<del>0W</del> - 925	.7	4		
	17 -97	<u>511</u> 900	. 6	4		
	17 -90	CHL 875	- 6	4		
	17 - 87	5W 850	.5	1		
	17 85	CHL 825 W	.7	1		
	17 80	OW	.8	4		
	17 77	Shi	6	2		
	17 75	ola	.0	2		
	17 73		.0	<u>र</u> र		
	17 74			с С		
	17 67	5W	.5	2		
				-		
	17 65	OW	• _	2		
	17 62	5W	.5	a		
	17 60	<u>OM</u>	1.0	6		
	17 57	5.4	• 7	5		
	17 55	OW	.6	12		
	17 52	5W	.9	42		
	17 50	0W	1.2	110		

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SAMPLE	AG PPM	AU** PPB
17 475W 17 450W 17 425W 17 400W 17 375W	- 4 - 4 - 5 - 4 1.0	4 3 5 5
17 350W 17 325W 17 300W 17 275W 17 250W	. 3 . 6 . 5 . 6 . 4	3 6 3 3 3 3
17 225W 17 <del>220W</del> 200W 17 175W 17 150W 17 150W 17 125W	.5 .5 .4 .4	3 4 3 5 1
17 100W 17 75W 17 50W 17 25W 17 0W	. 1 . 3 . 5 . 4 . 4	2 4 3 2 4
17 25E 17 50E 17 75E 17 100E 17 125E	.8 .6 .5 .5	9 2 10 2 19
17 150E 17 175E 17 200E 17 200A 17 225A	1.1 .6 .4 .4 .5	5 21 2 2 18
17 250A 17 275A 17 300A 17 325A 17 350A	- 5 - 5 - 4 - 1 - 4	21 43 8 57 8
17 375A 17 400E	.3 .2	6 41

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SAMPLE	AG FFM	AU** PPB
17 425E	.4	19
17 450E	.3	5
17 475E	.4	4
17 500E	.6	8
17 525E	.3	9
17 550E	. 6	2
17 575E	. 4	2
17 600E	. 8	28
17 625E	. 4	2
17 650E	. 4	3
17 675E 17 700E 17 725E 17 750E 17 775E	345 	10 2 2 1 2
17 800E 17 825E 17 850E 17 875E 17 900E	.8 .5 .6 .5 .7	6 17 9 15
17 925E	- 9	3
17 950E	- 4	17
17 975E	1 - 6	4
17 1000E	1 - 1	5
17 1025E	- 5	9
17 1050E	.6	38
17 1075E	.2	11
17 1100E	.9	22
17 1125E	1.1	5
17 1150E	1.0	85
17 1175E	1.1	7
17 1200E	1.6	5
17 1225E	.5	4
18 1250W	.9	5
18 1225W	.6	4
18 1200W 18 1175W	.5	9 2

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SAMPLE	AG PPM	AU++ PF®
18 1150W	. 4	9
18 1125W	. 7	4
18 1100W	. 6	9
18 1075W	. 7	6
18 1050W	. 4	16
18 1025W	.6	8
18 1000W	1.1	7
18 975W	.7	13
18 950W	.5	5
18 925W	.6	6
18 900W	.7	7
18 875W	.6	2
18 850W	1.6	2
18 825W	.6	17
18 800W	.8	5
18 775W	.6	4
18 750W	.7	4
18 725W	.5	8
18 700W	.8	4
18 675W	.7	15
18 650W 18 625W 18 600W 18 575W 18 550W	.5 .5 .5 .7	334 23 23
18 525W 18 500W 18 475W 18 450W 18 425W	- 4 - 8 - 7 - 7 - 7 - 9	2 13 4 3 1
18 400W	- 8	14
18 375W	- 7	3
18 350W	- 8	2
18 325W	- 6	5
18 300W	1 - 4	3
18 275W 18 250W	$3.2 \\ 1.1$	3 11

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SAMPLE	AG PPM	AU** PPB
18 225W 18 200W 18 175W 18 150W 18 125W	.8 .7 .9 .6 .8	4 3 4 2
18 100W 18 75W 18 50W 18 25W 18 0W	- ජ - ජ - ජ - 4 - 7	4 4 2 2
18 25E	1.0	9
18 50E	.5	6
18 75E	1.0	1
18 100E	2.0	2
18 125E	.7	5
18 150E	- 6	125
18 175E	- 8	6
18 200E	- 7	3
18 225E	- 7	9
18 250E	- 7	8
18 275E	.8	8
18 300E	.7	39
18 325E	.7	1050
18 350E	1.2	30
18 <del>350A</del> <i>375E</i>	1.1	15
18 <del>375A</del> 400 <i>E</i>	.8	29
18 <del>400A</del> 425	1.0	11
18 <del>425A</del> 450	.9	31
18 <del>450A</del> 475	.5	5
18 <del>475A</del> 500	.4	4
18 <del>500A</del> 525 18 <del>525A</del> 550 18 <del>550A</del> 575 18 <del>575A</del> 600 18 <del>600A</del> 625	-5 -7 -8 1-0 1-1	1 1 1 3
18 <del>625A</del> 650	- 8	1
18 <del>650E</del> 675E	- 7	2

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PAGE# 17

SAMPLE	AG FPM	AU★★ PPB
18 <del>675E</del> 700E	.6	8
18 <del>700E</del> 725	.6	7
18 <del>725E</del> 750	.5	6
18 <del>750E</del> 775E	.6	9
18 800E	.7	8
18 825E	. 6	5
18 850E	. 7	5
18 875E	. 8	8
18 900E	. 7	2
18 925E	. 8	10
18 950E	1.6	11
18 975E	1.5	7
18 1000E	.8	9
18 1025E	.9	12
18 1050E	1.0	17
18 1075E	.6	9
18 1100E	.9	17
18 1125E	.6	1
18 1150E	.8	6
18 1175E	1.0	2
18 1200E	. 7	1
18 1225E	. 9	90
18 1250E	. 9	2
19 1032W	. 7	4
19 1000W	. 4	3
19 975W	.7	2
19 950W	.6	36
19 925W	.7	4
19 900W	.7	3
19 875W	.8	3
19 850W 19 825W 19 800W 19 775W 19 750W	.8 .6 .7 .7	9 4 9 7 11
19 725W	-4	6
19 700W	.5	70

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SAMFLE	AG FFM	AU** PPB
19 675W 19 650W 19 625W 19 600W 19 575W	.3 .7 .6 .3 .3	20 6 4 7
19 550W 19 525W 19 500W 19 475W 19 450W	.4 .5 .3 .4 .7	5 4 1 1 1
19 425W 19 400W 19 375W 19 350W 19 325W	.5 .4 .5 1.0 .6	555 54 4
19 300W 19 275W 19 250W 19 225W 19 225W	.6 .3 .4 .6 .5	2 2 1 2 2
19 175W 19 150W 19 125W 19 100W 19 75W	.4 .5 .4 .3 .3	34553 553
19 50W 19 25W 19 25E 19 50E 19 75E	.3 .5 .5 .5 .3 .3	<b>6</b> 34 ភូម
19 100E 19 125E 19 150E 19 175E 19 200E	.4 .3 .5 .3 .4	8 2 2 4 4
19 225E 19 250E	- 4 - 4	4 2

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SAMPLE	AG PPM	AU* 🖌 PPB
19 275E 19 300E 19 325E 19 350E 19 375E	. 4 . 4 . 3 . 2	11 6 7 4 7
19 400E 19 425E 19 <del>475E</del> <i>450E</i> 19 <del>500E</del> <i>475</i> 19 <del>525E</del> <i>500</i>	.7 .5 .6 .4 .6	3 4 3 5
19 <del>550E</del> <i>\$25</i> 19 <del>575E</del> <i>\$50</i> 19 <del>600E</del> <i>\$75</i> 19 <del>625E</del> 600 19 <del>650E</del> 625	- 4 - 5 - 8 - 9 1.0	2 8 8 8 8 8 2
19 <del>675E</del> 650	.5	3
19 <del>700E</del> 675	.3	3
19 <del>725E</del> 700	1.3	11
19 <del>750E</del> 725	.7	7
19 <del>775E</del> 750	.4	13
19 <del>800E</del> 7 <i>75</i>	- 5	4
19 <del>825E</del> 800	- 6	3
19 <del>850E</del> 825	- 4	4
19 <del>875E</del> 850	- 5	3
19 <del>900E</del> 875	- 6	2
19 <del>925E</del> 900	1.6	3
19 <del>950E</del> 925	1.0	4
19 <del>975E</del> 950	.7	10
19 <del>1000E</del> 975	1.3	9
19 <del>1025E</del> /000	1.1	8
19 <del>1050E</del> /025	.4	7
19 <del>1075E</del> /050	.5	6
19 <del>1100E</del> /075	.6	7
19 <del>1125E</del> //00	.5	2
19 <del>1150E</del> //25	.3	2
19 <del>1175E</del> // <i>50</i>	.9	4
19 <del>1200E</del> // <i>75</i> E	.5	2

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SAM	PLE	AG PPM	AU** PPB
19 19 19 19 20	<del>1225E</del> /200E <del>1250E</del> /225 <del>1275E</del> /25 <sup>0</sup> <del>1274E</del> /274E DE	- 9 - 8 1 - 1 - 7 1 - 2	3 3 2 4
20 20 20 20 20	25E 50E 75E 100E 125E	- 4 - 6 - 8 - 6 - 6 - 9	3 4 9 5
20 20 20 20 20	150E 175E 200E 225E 250E	.7 .9 1.0 .9 .8	10 7 7 6 7
20 20 20 20 20	275E 300E 325E 350E 375E	.8 1.0 .7 1.0 1.1	6 13 11 7 9
20 20 20 20 20	400E 425E 450E 475E 500E	- 6 - 5 - 7 - 8 - 7	9 6 7 8
20 20 20 20 20	525E 550E 575E 600E 625E	- 7 - 9 - 7 - 7 - 5	6 5729
20 20 20 20 20	650E 675E 700E 725E 750E	- 6 - 6 - 8 - 6 - 6	3 2 4 6 4
20 20	775E 800E	- 4 - 9	12 9

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SAMPLE	AG	AU++
	PPM	PPB
20 825E	.8	9
20 850E	.5	6
20 875E	.8	6
20 900E	.3	7
20 925E	. 4	1Ŭ
20 950E	.3	4
20 9756	.6	9
20 1000E	.5	10
20 1025E	. 6	5
20 1050E	.5	10
20 1075E	. 4	4
20 1100E	.6	3
20 1125E	.5	3
20 1150E	.3	4
20 1175E	. 4	2
20 1200E	.3	5
20 1225E	.5	4
20 1250E	.5	3

ACME ANALYTICAL LABORATORIES LTD. 852 E. HASTINGS, VANCOUVER B.C. PH:253-3158 TELEX:04-53124

DATE RECEIVED JULY 18 1983

Huaz DATE REPORTS MAILED

#### ASSAY CERTIFICATÉ GEOCHEMICAL

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A .SOU ON SAMPLE IS DIGESTED WITH 3 ML OF 3:1:3 HCL TO HNO3 TO H20 AT 90 DEG.C. FOR 1 HOUR. THE SAMPLE IS DILUTED TO 10 MLS WITH WATER. ELEMENTS ANALYSED BY AA : AG. SAMPLE TYPE : SOIL - DRIED AT 60 DEG C., -BO MESH. AU\*\* - INCLUDING PD, PT 10 GM, FIRE ASSAY CONCENTRATION, HN03 LEACH OFF AG,

AQUA REGIA DIGESTION, GRAPHITE AA ANALYSIS.

ASSAYER

FILE # 83-1243

My DEAN TOYE, CERTIFIED B.C. ASSAYER PROJECT: DELEWARE RES. PAGE# 1

ANGINEL RESOURCES

SÀÌ	MF'LE	AG PPM	AU** PPB
20	950W	- 7	4
20	925W	- 8	3
20	900W	- 7	4
20	875W	- 3	2
20	850W	- 4	5
20	825W	. 7	4 N N A &
20	800W	. 4	
20	775W	. 5	
20	750W	. 5	
20	725W	. 5	
20 20 20 20 20	700W 675W 650W 625W 600W	.5 .5 .4 .3	2 2 7 5 15
20 20 20 20 20	575W 550W 525W 500W 475W	.3 .3 .5 .4 .4	5 16 33 9
20 20 20 20 20	450W 425W 400W 375W 350W	.3 .4 .3 .4 .5	2 2 2 5 3 6
20	325W	.3	11
20	300W	.3	11
20	275W	.2	3
20	250W	.5	17
20	225W	.2	3
20	200W	.1	4
20	175W	.4	3
20	150W	.4	5
20	125W	.2	9
20	100W	.2	4
20	75W	.4	10
20	50W	.3	17
20	25W	.5	39
20	0W	1.5	3

HISTOGRAM FOR ELEMENT LOG GOLD PROJECT: DELAWARE OPERATOR: DWR DATE FEB. 7/84

CELL		NO. OF	SAMPLES	CUM. FR. %
*******	*******	<del>****</del> **	********	******
3.02 -	2.81	1		0.13
2.81 -	2.60	1		0.25
2.60 -	2.39	1		0.38
2.39 -	2.18	1		0.50
2.18 -	1.97	4		1.00
1.97 -	1.76	11		2.38
1.76 -	1.55	17	,	4.52
1.55 -	1.34	34		8,78
1.34 -	1.13	61		16.44
1.13 -	0.92	106	,	29.74
0.92 -	0.71	119	ţ	44.67
Ø.71 -	0.50	179	)	67.13
0.50 -	0.29	218		94.48
Ø.29 -	0.08	Z	I	94.48
0.08 -	-0.13	44		100.00

DELAWARE Project Mean and standard deviation for LOG GOLD

Mean value= 0.750 Standard deviation= 0.424 Number of samples= 798 HISTOGRAM FOR ELEMENT LOG SILVER PROJECT: DELAWARE OPERATOR: DWR DATE FEB. 7/84

CELL NO. OF SAMPLES CUM. FR. %

0.53	-	0.43	2	0.25
0.43		0.33	7	1.13
0.33	-	0.23	5	1.76
0.23		0.13	19	4.14
0.13	—	0.03	41	9.28
0.03	-	-0.07	65	17.44
-0.07	-	-0.17	189	41.15
-Ø.17		-0.27	140	58.72
-0.27		-0.37	129	74.91
-Ø.37		-0.47	118	89.71
-0.47	-	-0.57	55	96.61
-Ø.57	-	-0.67	Ø	96.61
-0.67	-	-0.77	18	98.87
-0.77		-0.87	Ø	98.87
-0.87		-0.97	Ø	98.87
-0.97	-	-1.07	9	100.00

DELAWARE Project Mean and standard deviation for LDG SILVER

Mean value= -0.229 Standard deviation= 0.208 Number of samples= 798 Histogram for log gold - DELAWARE PROJECT

CELL	. 1	FREQ	
			⋾⋷⋳∊∊⋳∊∊⋵⋼∊∊∊⋼∊⋵⋍∊⋵⋼⋼⋳∊∊∊∊∊∊∊∊∊∊∊∊∊∊∊∊∊∊∊∊
3.02-	2.81	1	
2.81-	2.60	1	
2.60-	2.39	1	
2.39-	2.18	1	
2.18-	1.97	4	
1.97-	1.76	11	**
1.76-	1.55	17	***
1.55-	1.34	34	****
1.34-	1.13	61	****
1.13-	. 92	106	*****
.92-	.71	119	******
.71-	.50	179	*****
.50-	.29	218	******
.29-	. 28	Ø	
.08-	13	44	****
==========		=====	
SUM		797	
Histogram	for	log	silver - DELAWARE PROJECT
CELL	E	- DE O	
	۲ ۲۵۱۰ ۲۰۰۰ ۲۰۰۰ ۲۰۰۰	- ~ ~ ~ ~ ~	
.53-	.43	2	
.43-	. 33	7	*
.33-	.23	5	*
.23-	.13	19	***
. 1.3-	03	41	****

CEL	L F	FREQ	
		====	
.53-	.43	2	
.43-	.33	7	×
.33-	.23	5	*
.23-	.13	19	***
.13-	.03	41	******
.03-	07	65	*********
07-	17	189	**************************************
17-	27	140	******
27-	37	129	***********************
37-	47	118	****************
47-	57	55	*****
57-	67	Ø	
67-	77	18	***
77-	87	Ø	
87-	97	Ø	
97-	-1.07	9	
SUM	:	==== 797	



Please find enclosed, copy of analytical results from samples collected from my visit to the Delaware property. The locations for the samples will be general.

- #89026 Grab samples from the main dump area comprised of pyritic argillite referred to as syngenetic pyrite banding. The sample displays slight Au and Ag enrichment.
- #89027 Grab samples from the main dump. Graphitic argillite breccia mixture of syngenetic and replacement pyrite up to 5 to 10%. This sample contains anomalous Au levels and also illustrates that the As levels in the vein systems are anomalous.
- #89028 Grab sample main portal vein material, pyritic.
- #89029 Trench 2, east of main workings, grab sample from dump. Very graphitic - pyrite-quartz vein.
- #89030 East of main workings. Grab sample from dump with small ore car. Pyritic quartz vein material.
- #89031 Rhyolite float from surface near main workings. Represents sample from rhyolific sill which is in close proximity to the main workings. Contains 2 - 3% finely disseminated Pyrite.
- #89032 Vein quartz minor pyrite (5%) with 1 2% Galena. Grab sample from footwall of vein from crosscut.

DELAWARE RESOURCES CORPORATION

R. K. Netolitzky

RKN:sc Enclosure

P. Santos

# TERRAMIN RESEARCH LABS LTD.





## KAMLOOPS RESEARCH & ASSAY LABORATORY LTD.

912 - 1 LAVAL CRESCENT - KAMLOOPS, B.C. V2C 5P5 PHONE: (604) 372-2784 - TELEX: 048-8320 **CERTIFICATE OF ASSAY** 

B.C. LICENSED ASSAYERS **GEOCHEMICAL ANALYSTS** METALLURGISTS

TO <u>Mr. P. J. Santos</u>	Mr. R. Netolitzky		
	Delaware Resources Corp.	Certificate No	<u>K 5649</u>
626-9th Ave.	<u>150. 1300-8th St. S.W.</u>	Data	Julv 26, 1983
	Calgary, Alta.		
<u>_Castlegar, B.C.</u>	<u>I2R 1B1</u> Project: <u>Delaware</u>		
••• · · · · · · · · · · · · · · · · · ·			

J ljcrcby ccrtify that the following are the results of assays made by us upon the herein described \_\_\_\_\_ \_\_\_\_\_ samples

)	Marked	Au	Ag	Pb	Zn	· · · · · ·				 _
		ozs/ton	ozs/ton	per cent	per cen					 -
1	10101	.007	.15	.03	.48					
2	10102	.013	.12	.02	.02					
3	10103	.015	.52	.38	.54					
4	10104	.006	.12	.06	.09				1	
5	10105	.013	.15	.37	.24					8
6	10106	.005	.15	.17	18					00
7	10107	.023	.15	.04	.03		1	1		
8	10108	.008	.12	.04	.09		1			
9	10109	.023	.15	.03	.03					
0	10110	.004	.09	.01	.06					
1	10111	0.03	00	01	07					
2	10111	.007	.09	.01	.05					
2	10172	.042	.06	1 • UZ	•06					
	10112	.010	•12	.04	.08		1	1	1	
5	10115	•UZ I	•20	.21	- 14					
	10112	.062	.35	•24	.09		l			
0	10116	.007	.09	.05	.55		1	1		
	10117	.073	.29	.19	.38					
σ	10138	.005	.26	.24	.23					
9	10119	.014	.29	.25	.31					
u (	10120	.156	.20	1.11	.63		ļ			

NOTE:

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Rejects retained three weeks. Pulps retained three months

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Registered Assayer, Province of British Columbia

CTA,
$\mathbf{U}$
Member
Canadian Testing
Association

#### KAMLOOPS RESEARCH & ASSAY LABORATORY LTD.

912 - 1 LAVAL CRESCENT — KAMLOOPS, B.C. V2C 5P5 PHONE: (604) 372-2784 — TELEX: 048-8320 CERTIFICATE OF ASSAY B.C. LICENSED ASSAYERS GEOCHEMICAL ANALYSTS METALLURGISTS

2

ТО	Mr.	Ρ.	J.	Santos

Mr. R. Netolitzky

Project: <u>Delaware</u>

Certificate No. <u>K 5649</u>

Date \_\_\_\_\_ July 26, 1983

Jurchy criticy that the following are the results of assays made by us upon the herein described \_\_\_\_\_\_ samples

Kral No	Marked	Au	Ag	Ph	Zn						
		ozs/ton	ozs/ton	per cent	per cen	t					
21	10121	.49	.32	.14	•28						
22	10151	.006	.29	.09	.33						
23	10152	.007	.17	.08	.14						
24	10153	.003	.09	.03	.36						8
25	10154	.017	•15	.13	.51		1				Ψ
26	10155	.014	.44	.37	.33						
27	10156	.040	.12	.15	.29						
28	10157	.62	.12	.05	.17						
29	10158	.018	.23	.10	.24						
30	<u> </u>	.055	.38	.30	.29						
31	10160	.62	.17	.02	.09				1		
32	10161	.015	.09	.03	.36						
33	10162	.39	.70	.56	1.18						
34	10163	.038	.50	.40	.99				1		
35	10164	.027	.23	.14	.58	i	ļ				
36	10165	.004	.20	.13	.72	ļ					
37	10166	.009	.32	.20	1.10						
38	10167	.108	1.34	1.15	.96	ļ					
39	10168	.175	.44	.28	.89	1					
40	10169	.011	.29	.21	.74						
										]	
A DET	ľ	1			1	1	1	i	1	1	

NOTE: Rejects retained three weeks.

Pulps retained three months unless otherwise arranged.

Registered Assayer, Province of British Columbia



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### KAMLOOPS RESEARCH & ASSAY LABORATORY LTD.

912 - 1 LAVAL CRESCENT --- KAMLOOPS, B.C. V2C 5P5 PHONE: (604) 372-2784 --- TELEX: 048-8320 **CERTIFICATE OF ASSAY** 

Mr. P. J. Santos Mr. R. Netolitzky

Certificate No. <u>K 5649</u>

**METALLURGISTS** 

3 B.C. LICENSED ASSAYERS

**GEOCHEMICAL ANALYSTS** 

July 26, 1983 Date

#### Project: Delaware

Jurrby certify that the following are the results of assays made by us upon the herein described \_\_\_\_\_ \_\_\_\_\_ samples

Kral No	Marked	Au	Ag	Pb	Zn				
		ozs/ton	ozs/ton	per cent	per cer	t			_
41 42 43 44 45 46	10170 10171 10172 10173 10174 10175	.41 .204 .006 .010 .007 .006	.41 .55 .12 .17 .12 .15	.25 .39 .02 .07 .03 .04	.32 .60 .32 .21 .53 .38				8

NOTE: Rejects retained three weeks Pulps retained three months unless otherwise arranged.