520319

TECHNICAL REPORT FOR THE SILVER STREAK PROPERTY

British Columbia, Canada

Completed By:
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Tenajon Resources Corp.

Latitude: 54°11'N
Longitude: 126°45'E
NTS: 93L 02E

Completed For:

Tenajon Resources Corp.
860-625 Howe Street,
Vancouver, B.C.
V6C-2T6

January 14, 2003

Vancouver, Canada **Dave Visagie, P. Geo.**

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3.0 SUMMARY

This report is written as a Technical Report for the Silver Streak Property on behalf of Tenajon Resources Corp. The report is written so as to comply with the standards set out in National Instrument 43-101 for the Canadian Securities Administrators.

The Silver Streak Property is located 25 km south-southwest of Houston, British Columbia. Access is by all-season gravel roads from Houston: along the Morice Main logging road for 26.5 kilometres then 6 kilometres east on the Carrier Main Road.

The property consists of eight contiguous mineral claims. It is approximately 150 hectares in size. Tenajon Resources Corp can earn a 100% interest in the property, subject to a 1% Net Smelter Royalty, by making cash payments, issuing stock and completing work programs over a two year period.

Hazelton Group rocks, locally consisting of volcanic tuffs, argillite, conglomerate and sandstone, underlie the property. Previous work, by Equity Silver Mines Ltd., located widespread anomalous silver and base metal values within quartz veins/stockwork and breccia. At the time the anomalous values were interpreted as being confined to a favourable horizon.

Prior to being optioned a review of the available date relevant to the Silver Streak Property was undertaken by Tenajon personnel. The review concluded that rather than being confined to a favourable horizon the mineralization is structurally controlled by a series of north-northeast trending, sub-vertical faults that transect the property with the mineralization occurring in all units. In order to evaluate this interpretation, a series of trenches were excavated along strike from the main showing. The trenches verified the hypothesis, showing extensive faulting and shearing to occur along the projection of the mineralized trend. Within the fault zones minor quartz +/- carbonate veining was observed. Chip samples taken from across the fault zones returned anomalous silver and base metal values.

Based on this new interpretation the Silver Streak Property hosts at least eight, epithermal style, mineralized structures within a 150 metre wide northeast striking trend. The structures host variable amounts of chalcopyrite, galena, sphalerite, tetrahedrite and pyrite occurs in quartz veins/stockwork and breccia zones. Individual structures are up to 13 metres wide and have been traced along strike, through drilling and trenching, for up to 70 metres. The mineralized structures are all open along strike and down-dip. The largest zone, the Carrier Main, has been traced 75 metres down-dip, along strike for 70 metres with its' width variable to in excess of 15 metres metres. Significant results from the Carrier Main Zone include a drill hole intersection of 8.78 metres averaging 253.3 gpt Ag, 0.49% Cu, 0.15% Pb, 0.16% Zn and 0.25% Sb and a trench averaging 268.4 gpt Ag, 0.31% Cu, 0.37% Pb, 0.16% Zn and 0.06% Sb across 13.3 metres.

Based on the results it is concluded that the Silver Streak property has the potential to host significant silver and base metal values. Additional work consisting of geophysical surveying, mapping and drilling is required to further investigate the potential of the property.

It is recommended that an initial program consisting of mapping, geophysical surveying and diamond drilling be undertaken at the Silver Streak Property. The geophysical survey would be completed in an attempt to follow along strike the fault structures that have been shown to host the mineralized zones. Drilling would be undertaken along strike from the known intersections, in particular to the south. The cost of the program is calculated to be $92,180.

4.0 INTRODUCTION AND TERMS OF REFERENCE

This report is written as a Technical Report fof the Silver Streak Property on behalf of Tenajon Resources Corp. It is written so as to comply with the standards set out in National Instrument 43-101 for the Canadian Securities Administrators.

The author, a qualified person and employee of Tenajon Resources Corp., completed a preliminary evaluation of the Silver Streak property on September 12. Additional follow-up was completed between November 25 and 28, 2002.

In preparing this report, the author uses as sources of information the results of the 2002 program and those reports listed in the bibliography. In particular, the author relied on the descriptions and assay results as entered in the Equity Silver Mines drill logs a series of drill programs completed on the property between 1990-92.

5.0 DISCLAIMER

The author in writing this report uses as sources of information those reports listed in the bibliography and his own familiarity with the property. The government reports were prepared by person(s) holding post secondary geology or related degree(s), prior to the implementation of the standards relating to National Instrument 43-101. The information, based on the author's experience, is assumed to be accurate. The accuracy of the drill logs and the mode employed in the assaying of drill core has not been determined. However, again based on the author's experience, the work completed by Equity Silver Mines (Placer Dome Gold Mines) appears to have been completed to an industry standard.

6.0 PROPERTY DESCRIPTION AND LOCATION

The Silver Sleeper Property is centred at latitude 54°11"N, longitude 126°45'E. It occurs on NTS Sheet 93L 02E. The property is located approximately 25 km south-southwest of Houston, British Columbia.

The property is composed of eight, contiguous, unpatented mineral claims. It is approximately 200 hectares in size. All of the claims occur within the Omineca Mining District. Table 1 is a listing of the claims and their status composing the Silver Streak Property.

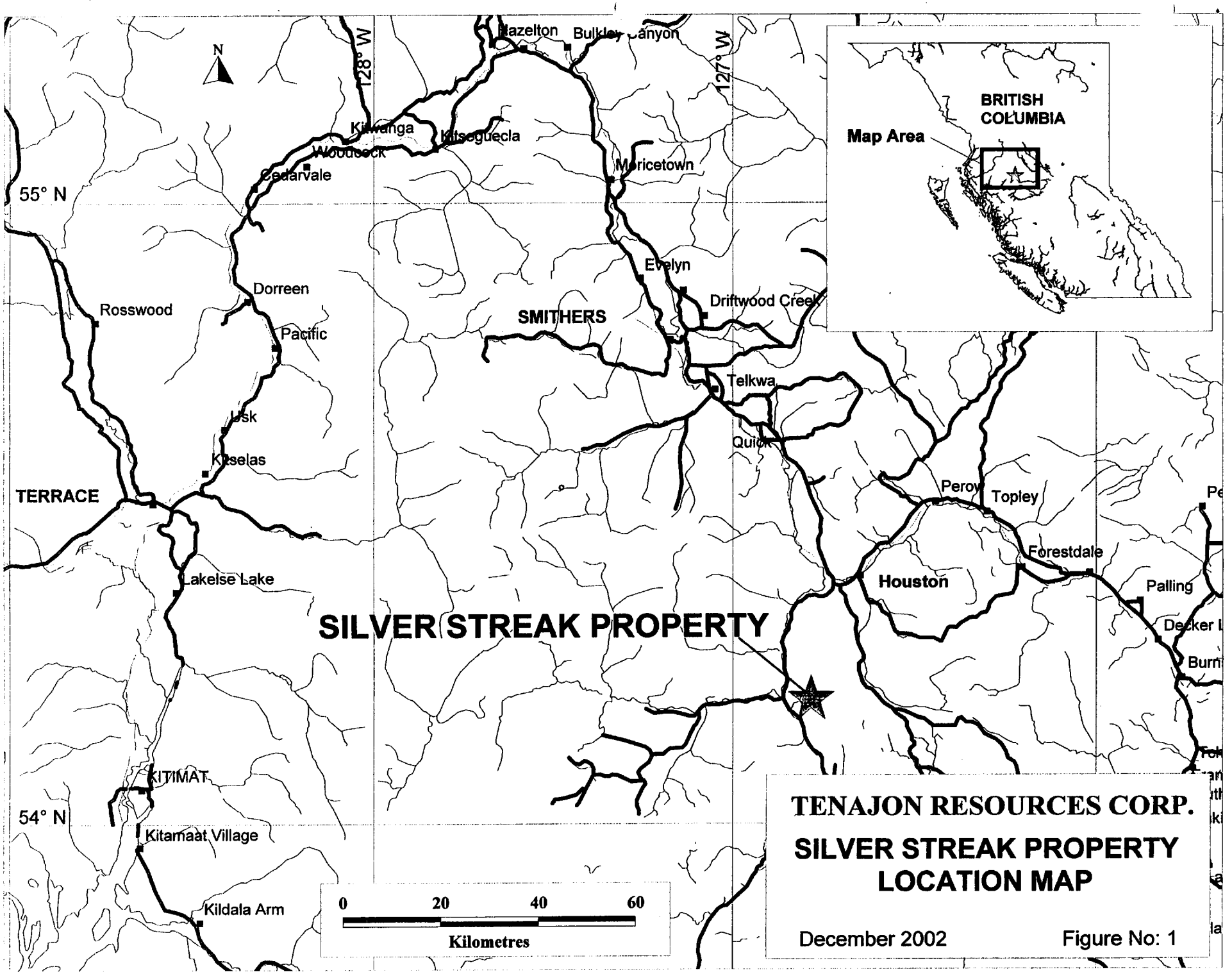

N
128° W
127° W
55° N
54° N
Hazelton
Bulkley Canyon
Kitwanga
Kitseguecla
Woodcock
Cedarvale
Moricetown
Evelyn
Driftwood Creek
SMITHERS
Dorreen
Rosswood
Pacific
Telkwa
Usk
Kitselas
Quick
TERRACE
Perow
Topley
Forestdale
Houston
Palling
Lakelse Lake
SILVER STREAK PROPERTY
KITIMAT
Kitamaat Village
Kildala Arm
BRITISH COLUMBIA
Map Area
0
20
40
60
Kilometres
TENAJON RESOURCES CORP.
SILVER STREAK PROPERTY
LOCATION MAP
December 2002
Figure No: 1

Table 1: Claim Status-Silver Sleeper Property

Claim	Owner	Record #	Units	Hectares	NTS	Date Recorded	Present Expiry Date
Silver Sleeper #1	E. Westgarde	385665	1	25	93L 02W	April 12/01	April 12/03
Silver Sleeper #2	E. Westgarde	385666	1	25	"	April 12/01	April 12/03
Cr #1	B. Hosfink	393123	1	25	"	May 3/02	May 3/03
Cr #2	B. Hosfink	193124	1	25	"	May 3/02	May 3/03
SS #1	G. Westgarde	392965	1	25	"	May 3/02	May 3/03
SS #2	G. Westgarde	392966	1	25	"	May 3/02	May 3/03
SS #25	E. Westgarde	385758	1	25	"	April 20/01	April 20/03
SS #26	E. Westgarde	385759	1	25	"	April 20/01	April 20/03

Upon acceptance of an assessment report filed for the property the expiry year for the Silver Streak property will be 2005.

All of the claims occur on provincially owned (Crown) ground. The claims give the titleholder only subsurface rights.

The claims require annual work commitments of $100/unit for the first three years and $200/unit for each year thereafter. The work required to hold the claims must be filed with the mining recorder prior to the expiry date.

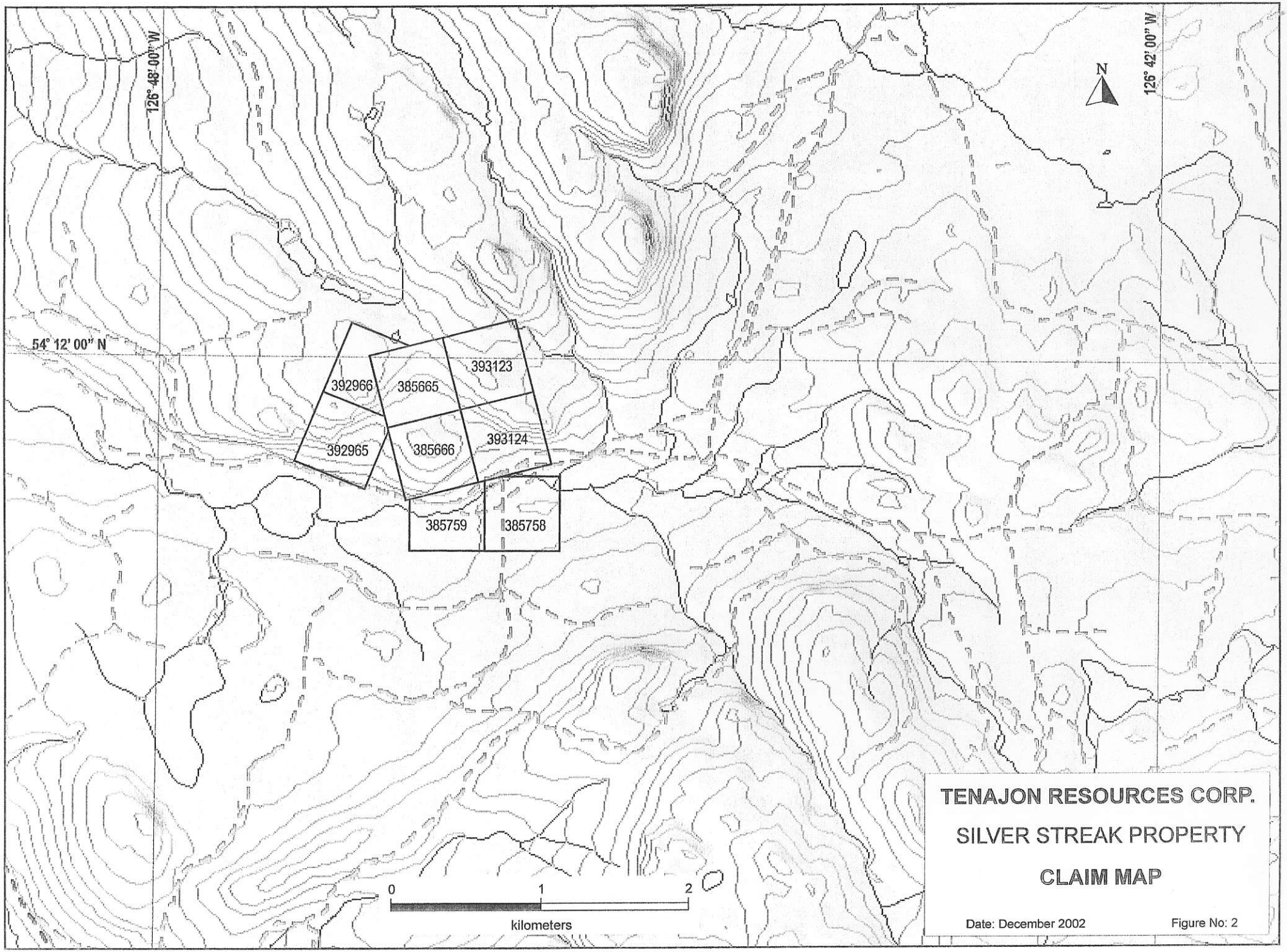
Under the terms of the option agreement with Westgarde et al., Tenajon Resources Corp can earn a 100% interest in the property by completing expenditures, issuing stock and cash as outlined in Table 2.

Table 2: Summary of Silver Streak Agreement

Date	Expenditure	Accumulated Expenditures	Shares Issued	Cash Payment
Nov 18, 2002			100,000	
May 18, 2003	$10,000	$10,000	100,000	$10,000
Nov 18, 2003			100,000	$10,000
Nov 18, 2004			200,000	$30,000
Total		$10,000	500,000	$50,000

The vendors retain a 1% Net Smelter Return royalty on precious and base metals that can be brought out by Tenajon, at its' sole discretion, for $1,000,000.

There are no known environmental liabilities associated with the project.


126° 48' 00" W
126° 42' 00" W
N
54° 12' 00" N
392966
385665
393123
392965
385666
393124
385759
385758
0
1
2
kilometers
TENAJON RESOURCES CORP.
SILVER STREAK PROPERTY
CLAIM MAP
Date: December 2002
Figure No: 2

In order to conduct work on the property, the operator must forward a completed Notice of Work Form to the British Columbia Ministry of Energy and Mines for its' approval. Depending on the work a reclamation bond may be required, with the amount being determined by the government based on the data supplied by the company relating to work disturbance. To get the bond back reclamation must be completed and the work inspected and passed by a district inspector.

7.0 ACCESSIBILITY, CLIMATE, LOCAL RESOURCES, INFRASTRUCTURE AND PHYSIOGRAPHY

Access to the property is via all season logging roads from Houston, B.C. To get to the property requires traveling 26.5 km southwards along the Morice Lake Main/Huckelberry Mine Roads to the intersection with the Carrier Main Road then proceeding 6 km eastwards. The Morice Lake/Huckelberry Main Roads are maintained year round while the Carrier Main is maintained through the winter if logging is taking place.

Houston is located approximately 60 km south of Smithers. B.C. Smithers is serviced daily by jet and turbo-prop service from Vancouver, B.C.

The climate is typical of central British Columbia being cool to cold in the winter while the summers are mild. Snowfall is variably moderate.

Houston has a population of approximately 4,000 people. Most services are available in Houston including groceries, hotel, expediting services and camp supplies. Anything that is not available in Houston can be shipped directly from major centers. Power-lines pass to within 6 km of the property's boundaries

The property is situated along a northeast trending valley that is mostly covered by thick overburden and swampy areas. Local elevation ranges from approximately 853 metres in the valley floor to 1127 metres along nearby ridges.

Bedrock exposure is generally poor within the valley but is excellent along the ridge tops.

In 1983, a large forest fire passed over the property. Subsequently large portions of the property were clear-cut logged. In areas that weren't clear-cut the trees have fallen down and extensive areas of deadfall formed. Recently, slide alder has grown in significant sized patches throughout the property.

There is sufficient crown land available for mining operations to proceed if the project was found to host an economic deposit.

8.0 PROPERTY HISTORY

Work on the Silver Streak property dates back to 1989 when local prospectors located a showing, now referred to as the Carrier Main Showing, containing anomalous copper and silver values within brecciated and veined lapilli tuff and argillite along the Carrier Main logging road. The property was subsequently optioned to Equity Silver Mines Ltd. who

completed several exploration programs that included trenching, mapping, geochemical and geophysical surveying along with reverse circulation and diamond drilling. The drill records were obtained by the vendors from the government however there is no report that documents all of the work completed on the property. The only work filed for assessment was a 1990 geophysical report that documents the results of an Induced Polarization Survey completed by Peter E. Walcott and Associates on behalf of Equity Silver Mines.

Table 3: Silver Streak Property History

Year/ Company	Program	Description/Results
1989 F. Haden, L. Bourgh	Prospecting	Prospecting along the Carrier Main Road located an outcrop of brecciated and veined lapilli tuff and argillite hosting anomalous copper and silver values. Staked the Eric claims to cover the property. Property optioned to Equity Silver Mines.
1990 Equity Silver Mines	Mapping, trenching, soil sampling, drilling	A grid, consisting of thirteen-1700 m long lines at 200-metre spacing with a 2400 m baseline located over the showing. Stations established every 25 metres along the lines. Mapping and sampling completed-results not known. Trenching of the Carrier Main Showing undertaken. Results included a 32 metre exposure averaging 338 gpt Ag with 1.99% Cu. Induced Polarization survey completed over grid. The 50 metre pole-dipole survey showed the "...property to have low chargeability with two zones of moderate chargeability response. Reproducing the data showed a series of sub-parallel higher chargeability zones that may be related to mineralization or argillite." Seven reverse circulation and diamond drill holes, totaling 1076.9 metres, completed. Drilling tested a variety of targets. One hole drilled close to Carrier Showing intersected 7.5 metres averaging 39.6 gpt Ag.
1991 Equity Silver Mines	Drilling	Thirteen holes, totaling 1,081.99 m in length, completed. Drilling was based on the idea of a shallow favourable horizon hosting anomaleus values. Holes drilled under and along strike from the Carried Road Showing intersected the zone over a 50 metre strike length at depths of up to 70 metres. Best result is an 8.78 metre section averaging 253.3 gpt Ag, 0.49% Cu, 0.15% Pb, 0.16% Zn and 0.25% Sb.
1992 Equity Silver Mines	Drilling	Six holes, totaling 439.80 metres, drilled. The holes were located so as to test the projection of an interpreted flat to shallow east dipping projection of the mineralization located at the Carrier Road Showing. Zones intersected to the east of the main showing. Results include a 9.78m @ 49.88 gpt Ag with 0.07% Cu, 0.15% Pb, 0.15% Zn and 0.02% Sb and a 0.86m section assaying >200g Ag, 0.22% Cu, 0.23% Pb

		and 0.12% Zn. Option allowed to lapse.
2001 Westgarde et al.	Staking	Staked six claims over the Carrier Road Showing.
2002 Westgarde et al.	Staking, trenching	Two claims added to property. Trenching, completed approximately 10 metres north of the Carrier Road Showing, located an extensive zone of malachite, azurite, chalcopyrite and tetrahedrite within veins and host argillite and lapilli tuff.
2002 Tenajon Resources Corp.	Trenching, mapping, sampling and soil sampling	Trench over the Carrier Road Showing expanded. Three other trenches completed. Backhoe trenching totaled 146.3 metres. Seventy-four rock chip samples collected from trenches. Soil sampling resulted in the taking of 11 samples. In addition one soil profile, totaling two samples, completed. At the Carrier Road Showing three mineralized structures identified. Results include a 13.3 metre section averaging 268 gpt Ag, 0.31% Cu, 0.40% Pb and 0.17% Zn. Forty metres along strike to the north a 4 metre section assaying 20 gpt Ag, 0.10% Pb and 0.07% Zn was located.

9.0 GEOLOGICAL SETTING

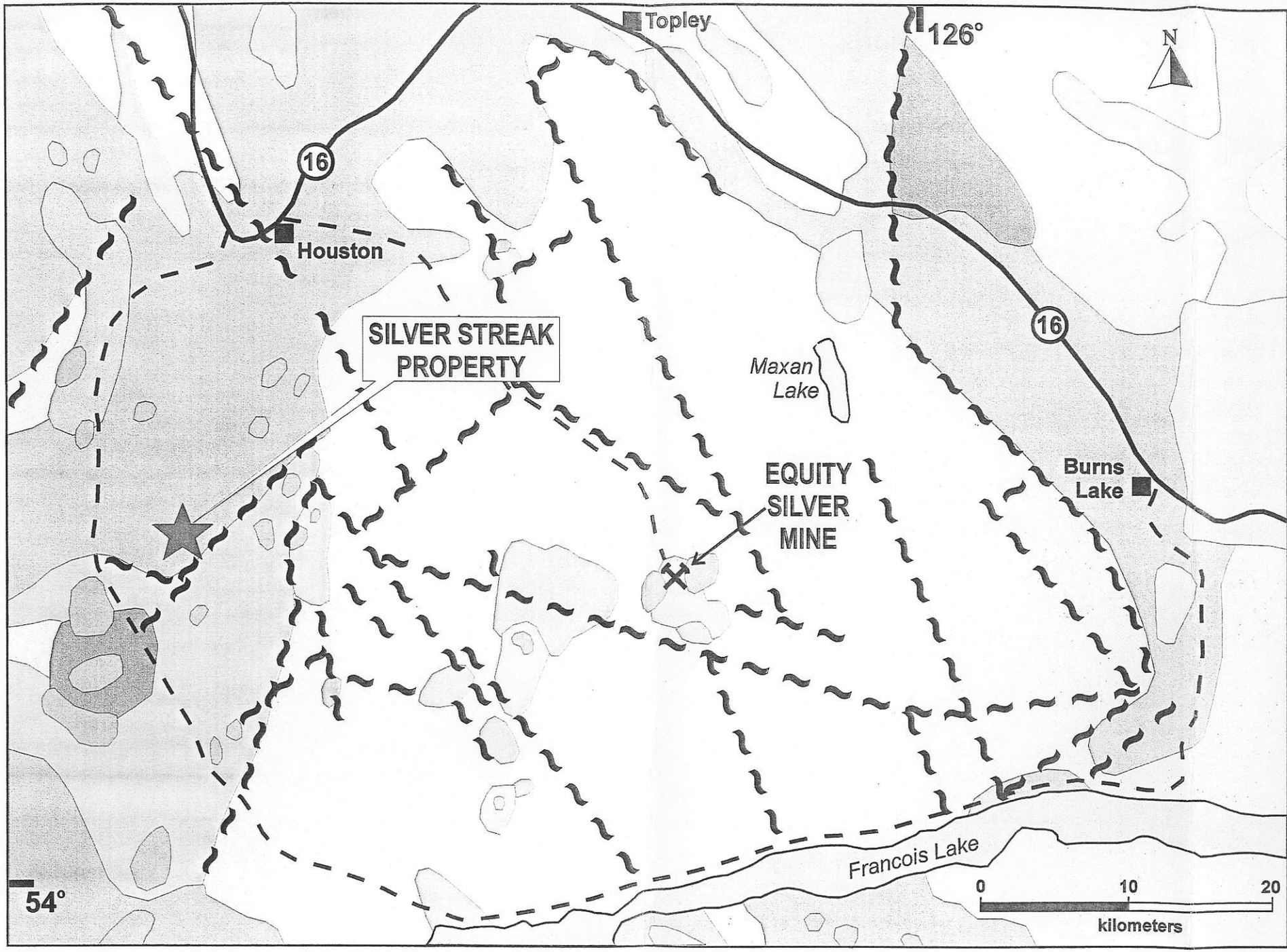
Various mapping programs have been completed on a regional scale by different government agencies. Systematic mapping of the Silver Streak (Erik) property was probably completed by Equity Silver Mines Ltd. however the author does not have access to the results.

9.1 REGIONAL GEOLOGY

The regional geological setting of the Silver Sleeper Property is based largely on the maps published by Tipper and Richards (GSC Open File 351) and by the map compiled by Church accompanying BCDM Paper 1990-2.

Regionally the property occurs within the central portion of the Intermontaine Belt. The Intermontaine Belt is principally composed of Mesozoic volcanic and sedimentary rocks. Within the belt are a series of successor basins. The Silver Streak Property lies south of the Skeena Arch, a prominent, early Mesozoic, northeasterly trending transverse structure that marks the approximate boundary between the Bowser Successor Basin to the north and a broad area to the southeast that is covered by a veneer of early to late Tertiary volcanic rocks. The Skeena Arch provided one of the controls for emplacement of Upper Triassic and Lower Jurassic granitic plutons. Granitic intrusions of Late Cretaceous and Early Tertiary age that intrude Mesozoic volcanic and sedimentary rocks throughout the Intermontaine Belt show no apparent relationship to the Skeena Arch as now defined.

The Silver Streak Property occurs within an area that is underlain by an incomplete succession of Lower Jurassic to Miocene volcanic and sedimentary rocks. Although Lower to Middle Jurassic age formations are regionally the most extensive units in the area, there are substantial amounts of the area covered by younger (Eocene to Miocene) volcanic



LEGEND

SEDIMENTARY & VOLCANIC ROCKS

Tertiary

- Basalt & andesite flows & breccias

Cretaceous and Tertiary

- Basalt, andesite, dacite
- Sandstone, conglomerate, shale

Cretaceous

- Siltstone, sandstone, shale

Jurassic and Cretaceous

- Siltstone, greywacke, sandstone, conglomerate, argilite, rhyolite

Jurassic

- Andesite, basalt, dacite tuffs & breccias

INTRUSIVE ROCKS

Tertiary

- Gabbro, syenomonzonite, quartz monzonite porphyry, feldspar porphyry

Cretaceous

- Porphyritic quartz monzonite, granodiorite

Jurassic

- Porphyritic quartz monzonite, granodiorite, quartz diorite
- Fault
- Road - highway
- Road - secondary
- Town
- Lake

Map from: Carter 1981

TENAJON RESOURCES CORP.

SILVER STREAK PROPERTY REGIONAL GEOLOGY

Date: December 2002 Figure No: 3

rocks largely consisting of plateau basalts and andesitic flows. Upper Jurassic to Middle Miocene intrusive rocks intrude these strata, with only Cretaceous and Tertiary intrusions appearing to host significant mineralization. The stratigraphy is summarized in Table 4.

Table 4: Regional Stratigraphy: Silver Streak Property

VOLCANIC AND SEDIMENTARY ROCKS				
ERA	**PERIOD**	**EPOCH**	**FORMATION**	**LITHOLOGY**
Cenozoic	Tertiary	Eocene and Miocene	Endako Group, Goosly Lake and Buck Creek Volcanic Rocks	Basalt and andesite flows and breccias, some rhyolite and dacite
			Uncoformity	
Mesozoic and Cenozoic	Cretaceous and Tertiary	Upper Cretaceous	Ootsa Lake Group, Tip Top Hill Volcanic Rocks	Basalt, andesite, dacite and related tuffs and breccias; some rhyolite flows and breccias
			Sustut Group (in part)	Sandstone, conglomerate and shale
			Unconformity	
Mesozoic	Cretaceous	Lower Cretaceous	Skeena Group, Brian Boru and Red Rose Formations	Siltstone, sandstone, shale, porphyrytic andesite flows; breccias and tuffs
	Jurassic and Cretaceous	Middle Jurassic, Upper Cretaceous	Hazelton Group (in part)	Siltstone, greywacke, sandstone, conglomerate, argillite
			Kasalka Group	Pebble conglomerate, rhyolite and andesitic pyroclastic and flow rocks
			Unconformity	
	Jurassic	Middle Jurassic	Hazelton Group	Andesite, basalt, dacite tuffs and breccias, volcanic sandstone and conglomerate siltstone and greywacke
			Unconformity	
		Lower Jurassic	Hazelton Group	Green, red and purple andesite and basalt tuffs and breccias, volcanic sandstone and conglomerate; argillite and greywacke
Intrusive Rocks				

Cenozoic	Tertiary	Eocene	Goosly Lake Intrusions Nanika Intrusions	Gabbro, syenomonzonite, quartz monzonite porphyry, feldspar porphyry and felsite
Mesozoic	Cretaceous	Upper Cretaceous	Bulkley Intrusions	Porphyrytic quartz monzonite and granodiorite
	Jurassic	Upper Jurassic	Francois Lake Intrusions	Porphyrytic quartz monzonite, granodiorite and quartz diorite

The volcanic and sedimentary units are generally shallow dipping with the strike being highly variable. Throughout the area, faulting is common with both north-northwest and north-northeast being predominant.

9.2 PROPERTY GEOLOGY

Thick overburden and swamp cover limits bedrock exposure south of the Carrier Main Showing. North of the showing outcrop ridges and isolated outcrops are common.

Detailed mapping of the property was probably completed by Equity Silver Mines, however the author has not been able to obtain the results of this work. The following is based on the limited information listed in the government files, government reports, the interpretation of the drill logs and observations made by the author.

9.2.1 Lithology

Government maps show the Silver Streak property to be underlain by Hazelton Group rocks locally consisting of argillite, andesitic tuffs/ash tuffs, andesitic lapilli tuffs and volcanic sandstone, conglomerate and siltstone. Based on the drill hole data the units are flat to gently east dipping.

Drilling indicates that andesitic tuff is dominant at depth, with the tuff being overlain by lapilli tuff then by argillite. Within the volcanic cycle argillite and volcanic sediments occur throughout. Andesite dykes intrude the units. Airborne geophysics indicates the property is south proximal to a magnetic high that may be indicative of a buried intrusive. Approximately 2 kilometres south of the property a small exposure of felsic porphyry intrusive has been located.

9.2.2 Structure

Government mapping indicates the property is close to the junction of north-northeast and northwest trending faults. The faults are common throughout the region. At the Carrier Main Showing, mapping has located a series of north-northeast trending, steeply dipping

faults, that appear to be related to mineralization. In drill core, extensive zones of faulting, shearing and gouge occur that appear related to mineralization emplacement. Fracture zones are observed on surface and within drill core that reflect the overall regional faulting patterns. Breccia zones occur intermittently throughout.

9.2.3 Alteration

At the Carrier Main Showing the rocks are variable propylitically (chlorite, calcite, pyrite, epidote, zeolite) altered. The alteration is best developed within and immediately proximal to the main mineralized trend. Sericite and silicification are localized to the zone itself.

10.0 DEPOSIT TYPE

Based on the style of mineralization, mineral assemblage and geochemistry the Silver Streak Property can be best described as hosting low sulphidation epithermal mineralization. Epithermal deposits are considered to be formed by "hot ascending waters of uncertain origin, but charged with igneous emanations." The main characteristics of epithermal deposits are listed below.

- The deposits form near the surface. Mineralization takes place from surface to a maximum depth of about 1,000 metres. Ore can be developed to over a considerable strike length, but is restricted In vertical extent to intervals varying from 100 to 1000 metres. Average range of ore is 350 metres, it rarely exceeds 600 metres. Ore zones bottom in either barren rock or pass downward into sub-economic zones containing base metal sulphides.

- Veins are the most common ore host; they tend to branch or flare upwards into complicated wedge or cone-like features. Brecoia zones, stockworks and fine-grained bedding replacement zones also occur; larger zones of these types may extend to tens of millions of tones in size.

- Deposits form in extensional tectonic settings, in areas with well-developed tension fracture systems and normal faults. The fracture systems are commonly, but not necessarily, associated with large-scale volcanic collapse structures.

- Mineralization commonly occurs in volcanic terranes with well-differentiated, subaerial pyroclastic rocks and numerous small-sub-volcanic intrusions. Hot spring deposits and fumarolic volcanic phenomena are sometimes evident where centres of hydrothermal dischargs have not been deeply eroded.

- Ore and associated minerals are deposited dominantly as open space filling with banded crustiform, vuggy drusy, colloform and cockscomb textures. Repeated cycles of mineral deposition are evident. Ore minerals are generally fine grained but commonly have coarse grained, well-crystallized overgrowths of gangue minerals. Some replacement textures are evident. Multiple periods of brecciation common.

- Gold and silver are the main economic minerals along with enhanced amounts of Hg, As, Sb and rarely Tl. Gold to silver ratios range widely. Main ore minerals are native gold and silver, electrum, acanthite (argentite) and silver bearing arsenic-antimony sulphosalts. Tellurides are locally important. In addition, galena and sphalerite are common; copper occurs generally as chalcopyrite but in some deposits forms as enargerite. Cinnibar, stibnite, tetrahedrite and selenides are important in some deposits.

- Gangue minerals are mainly quartz and calcite with lesser fluorite, barite and pyrite. Chlorite, hematite, dolomite, rhodonite and rhodochrosite are less common. Silica occurs in many varieties, most commonly as quartz or amethystine quartz, but also as opal, chalcedony and crystobilite.

- Hydrothermal alteration is pronounced. Precious metal mineralization is frequently associated with silicification. Zones of silicification can be flanked by zones of illite-sericite and clay alteration, all occurring within larger zones of propyllitic alteration. At depth, vein structures contain adularia; near surface broad argillic zones can predominate. Some deposits have aluminous, advanced argillic alteration assemblages containing kaolinite/dickite, sericite, pyrophyllite and andalusite with accessory diaspore, corundum, topaz, lazulite or scorzalite, dumortierite and rutile or anatase.

Epithermal ores occur in all rock types, particularly those that sustain large open-spaced fractures over extended periods of time. The heat source for the fluids does not appear to be the latent heat of volcanism but is more likely derived from structurally controlled sub-volcanic intrusions or deeper plutons. Any rock type that maintains primary or structurally induced permeability and permits focused hydrothermal flow can provide sites for ore deposition.

Epithermal districts are common throughout the world. Type deposits include the Hishikari gold deposit in Japan, the Comstock gold-silver District of Nevada and the silver deposits of the Guanajuato District in Mexico. The nearby Equity Silver Mine (past production of 33.8 million tonnes averaging 0.40% Cu, 64.9 gpt Ag and 0.46 gpt Au is considered by some to be transitional deposit linking mineralization of the porphyry copper and epithermal types while others consider it to be a remobilized volcanogenic massive sulphide.

Rock samples are variably Au, Ag, Zn, Pb, Cu, As, Sb, Ba, F, Mn and locally Te, Se and Hg enriched.

With regards to geophysical surveying VLF-EM has been used to trace structures while radiometric surveys may outline strong potassic alteration of wallrocks. Detailed gravity surveys may delineate boundaries of structural blocks with large density contrasts contrasts.

11.0 MINERALIZATION

The Silver Streak Property hosts sulphides consisting of variable amounts of pyrite, galena, sphalerite, chalcopyrite and tetrahedrite as:

- fine disseminations and blebs within quartz veins/stockwork and breccia zones. The style of mineralization occurs within all rock types and appears to be related to fracture and faults zones,
- as fracture coatings and occasionally as
- disseminated sulphides within lapilli tuffs and to a much lesser degree in argillite.

At surface malachite and azurite are commonly developed. Drilling indicates the bulk of mineralization is hosted within quartz±carbonate veins/stockwork and breccia zones that appear to be related to extensive fault and fracture zones. Eight mineralized structures have been identified over a 150 m width. These structures vary in width from less than a metre to in excess of 10 metres. The largest is up to 15 metres wide and has been traced through drilling and trenching for in excess of 70 metres with the along strike extensions being open. Overall the structures strike north-northeast with the dips being vertical. The mineralization has been traced for up to 75 metres down dip and is open at depth. Grade is highly variable with the best intercept, located 35 metres down dip from the Carrier Main Showing, averaging 253.3 Ag, 0.49% Cu, 0.15% Pb and 0.16% Zn over 8.78 metres. The intersection was adjacent to a 12.36 metre section assaying 38.29 gpt Ag.

12.0 2002 EXPLORATION PROGRAM

The program undertaken by Tenajon Resources Corp was supervised by Dave Visagie a Qualified Person employed by Tenajon Resources Corp. The program was completed in two visits as documented below.

Table 5: 2002 Silver Streak Property-Exploration Personnel

Name	Address	Position	Field Days	Mobe	Total
Dave Visagie	860-625 Howe Street, Vancouver, B.C. V6C-2T6	Senior Geologist, Tenajon Resources Corp.	Sept 12, Nov 26-28	½ Sept 11, ½ Sept 13, Nov 25, 29	9
Topley Contractors (Bob Hogstead)	Box 976, Houston, B.C. V0J 1Z0	Excavator Operator	Nov 25-29		
Total			**8**	**3**	**11**

The purpose of the program was to complete excavator trenching, mapping and sampling at and within the immediate vicinity of Carrier Main Road Showing. To complete the

trenching program an excavator was employed from Topley Contractors. The program resulted in the following being completed:

Table 6: 2002 Trenching Program Summary-Silver Streak Property

Showing	Trench	Total Continuous Chip Length (m)	# of Samples
Carrier Main	2002-1	50m	27
	2002-2	68m	33
	2002-3	17m	7
	2002-4	16m	8
Totals	**4**	**151m**	**75**

Sample locations were tied to a grown over grid previously emplaced in 1990 by Equity Silver Mines Ltd.

The program located two drill collars from the 1990-1992 drill programs. Two other drill sites were located but the collars not located. In addition, four other collars were looked for but could not be positively located. In general, the drill collars and sites appear to be close to where they were identified in the drill logs.

The trench program was designed to evaluate the extent of the mineralization at the Carrier Main Road Showing. The trench locations are plotted on Figure 4. Appendix 1 is a listing of the sample descriptions while Appendix 2 lists the assay results. The results of the trench-sampling program are summarized below.

Table 7: 2002 Trenching Program Summary of Results Silver Streak Property

Trench #	From (m)	To (m)	Int (m)	Ag (gpt)	Cu (%)	Pb (%)	Zn (%)	Sb (%)	As (%)	Comments
2002-1	6.0E	10.0E	4.0	225.2	0.29	0.41	0.28	0.09	0.04	Lapilli Tuff, Cp, Tet, Ga, Sp in veins/stockwork and diss
	10.0E	12.0E	2.0	12.5	0.01	0.03	0.22	<0.01	<0.01	Lapilli Tuff-barren
	12.0E	19.3E	7.3	362.2	0.40	0.44	0.08	0.08	0.07	Argillite-well mineralized
	19.3E	21.0E	1.7	6.7	<0.01	<0.01	0.05	<0.01	<0.01	Lapilli Tuff-barren
	21.0E	24.0E	3.0	89.7	0.12	0.05	0.14	0.02	0.04	Argillite-weak gossan
	36.0E	37.0E	1.0	24.2	0.01	0.67	0.24	<0.01	<0.01	Argillite minor veins with ga
or	6.0E	19.3E	13.3	268.4	0.31	0.37	0.16	0.07	0.05	
or	6.0E	24.0E	18.0	213.9	0.25	0.28	0.15	0.06	0.04	
2002-2	27.0E	29.0E	2.0	11.3	0.03	0.04	0.03	<0.01	<0.01	Highly gossanous andesite? clay altered.
	45.0E	49.0E	4.0	5.2	0.01	0.12	0.08	<0.01	<0.01	
	60.0E	62.0E	2.0	28.4	0.02	0.10	0.07	<0.01	0.02	
or	58.0E	68.0E	10.0	12.5	0.01	0.07	0.06	<0.01	0.01	
2002-3										Highly sheared argillite with elevated Pb, Zn and As values t/o
2002-4										Sporadic elevated Zn values within andesitic tuffs

Trench 2001-1 is located at the Carrier Main Showing. Previous work by Equity Silver Mines at the Carrier Main Showing located a 32 metre trench in the area that reportedly assayed 388 gpt Ag, 1.99% Cu. The author does not know the exact location and orientation of this trench however based on talks with the vendors it appears to be 5-10 metres north of Trench 2001-1.

Trench 2002-2 is lacated approximatoly 35 metres north of 2002-1 while 2002-3 is approximately 20 metres north of the east end of 2002-2. Trench 2002-4 is centred approximately 30 metres northwest of the west end of Trench 2002-2.

The results show that the fracture/fault system that hosts the mineralization at the Carrier Main Showing to continue northeastward. Elevated base metal and silver values occur in Trench 2002-2 at the projection of what is interpreted to be the mineralization controlling fault system. Drilling has intersected the zone 20 metres to the south of trench 2002-1.

Soil profiling was completed in one cut located in Trench 2001-1 over an area of heavy sulphide mineralization. The results are summarized below.

Depth	Ag (gpt)	Cu (ppm)	Pb (ppm)	Zn (ppm)	Sb (ppm)	As (ppm)
30 cm	<0.3	23	8	62	<3	6
120 cm	1.5	38	21	90	7	9

The results show samples collected at a shallow depth would not detect the zone. Deep sampling is required to detect the mineralization.

A line of B-horizon soil sampling was completed on an east-west line located at 150N on the existing grid. The limited survey was hampered by frozen ground. As a result all of the samples were collected at depths of less than 30 centimetres. The survey results did not define any zones of interest.

Table 8: Soil Sample Locations and Results Silver Streak Property

Sample	Easting	Ag (ppm)	Cu (ppm)	Pb (ppm)	Zn (ppm)	As (ppm)
1892974	-50	<0.3	18	7	103	5
1892975	-25	<0.3	8	5	62	<2
1892976	0	0.3	33	11	111	2
1892977	25	<0.3	8	7	69	<2
1892978	50	<0.3	6	9	64	2
1892979	75	<0.3	8	4	52	3
1892980	100	<0.3	9	6	51	4
1892981	125	0.4	9	5	57	<2
1892982	150	<0.3	7	4	49	2
1892983	175	<0.3	6	3	49	2

13.0 DRILLING

Tenajon Resources Corp. has completed no drilling on the property. Equity Silver Mines completed three drill programs between 1990 and 1992. The programs included both diamond and reverse circulation drill holes. The author has copies of the drill holes but cannot determine, outside of a couple holes, which were drilled employing reverse circulation methods. The following summarizes the drill hole locations for all three years.

Table 9: 1990-1992 Drill Hole Locations: Silver Streak Property

Year	Hole	Northing	Easting	Dip	Bearing	Elevation (metres)	Length (metres)
1990	90-1	-6.00	2.00	-45	340	838.0	115.2
	90-2	-6.00	2.00	-90	0	838.2	188.1
	90-3	64.50	186.00	-45	340	Not listed	182.9
	90-4	104.00	-37.30	-45	160	Not listed	113.9
	90-5	-49.00	231.00	-49	340	Not listed	194.2
	90-6	-117.00	43.00	-45	340	Not listed	170.1
	90-7	-13.50	-102.00	-45	340	Not listed	112.5
						Total	**1076.9**
1991	91-1	12.50	-30.50	-45	090	497.0	102.11
	91-2	16.50	44.50	-45	270	500.0	100.58
	91-3	17.50	80.50	-45	270	498.0	98.92
	91-4	67.00	47.00	-45	270	510.0	97.54
	91-5	18.00	83.00	-90	0	498.0	53.64
	91-6	-60.00	82.00	-90	0	495.0	79.25
	91-7	-111.00	66.00	-90	0	492.0	106.68
	91-8	-81.00	-67.00	-90	0	490.0	70.10
	91-9	-80.50	-17.00	-90	0	490.0	54.86
	91-10	68.50	99.50	-45	270	505.0	27.43
	91-11	66.00	9.50	-45	270	510.0	76.20
	91-12	111.00	7.50	-45	270	520.0	105.18
	91-13	109.50	-23.00	-45	270	520.0	109.50
						Total	**1081.99**
1992	92-1	-16.00	62.00	-50	272	Not listed	70.08
	92-2	-15.50	92.00	-47	272	Not listed	73.15
	92-3	-45.00	103.00	-48	272	Not listed	66.14
	92-4	-43.50	133.00	-45	271	Not listed	85.34
	92-5	-75.00	172.50	-45	270	Not listed	63.40
	92-6	-74.00	142.50	-45	270	Not listed	81.69
						Total	**439.80**
						Total All	**2598.69**

The drill collar elevations for holes 90-1 & 2 are inconsistent with the rest of the holes.

The results of the drill programs are summarized below.

Table 10: Summary of 1990-1992 Drill Results: Silver Streak Property

Hole	From	To	Int (m)	Ag (gpt)	Cu (%)	Pb (%)	Zn (%)	Sb (%)	Description
90-1	16.9	24.4	7.5	39.6	0.07	0.02	0.16	0.01	Lapilli Tuff, minor qv/bx minor py.
90-2	13.3	28.8	15.5						0.13% As in carb alt Andesite Tuff / Lapilli tuff.
90-3									No Sig. Results.
90-4									Nothing Sampled.
90-5									Nothing Sampled.
90-6	55.6	60.2	4.6	35.5	0.03	0.16	0.17	0.03	Veined Lapilli Tuff, 2-3% f.g. py.
90-7									Nothing Sampled.
91-1	12.09	14.53	2.44	32.9	0.06	0.02	0.08	0.02	Lapilli Tuff-fractured, minor tet.
	34.38	39.62	5.24	117.4	0.17	0.07	0.12	0.07	Andesitic Tuff-Faulted, car veined, minor tet as dis. and in veins.
	49.39	50.62	1.23	192.7	0.39	0.11	0.11	0.18	Argillite, int chaotic, sheared, fractured, tr tet, tr cp.
	64.30	65.00	0.70	194.8	0.29	0.83	0.49	0.11	And. Tuff, faulted, carb/q vein with abundant py and tr tet t/o.
	73.92	76.45	2.53	34.5	0.04	0.07	0.05	0.02	Arg. With wispy carb veinlets, minor tet + cp.
	91.63	102.11	10.48	100.6	0.11	0.13	0.08	0.06	Lapilli Tuff and argillite, brx and q vn to variable diss tet, cp in host and veinlets. Last 2.94 m of hole assayed 154.4 gpt Ag with 0.20% Cu, 0.02% Pb, 0.06% Zn and 0.10% Sb.
91-2	24.16	24.38	0.22	58.9	0.07	0.36	0.20	0.03	Hanging sample, Argillite-veinlets minor tet, ga within veins.
	30.89	35.05	4.16	36.1	0.05	0.14	0.15	0.02	Argillite and And Tuff. Frac t/o tr tet t/o in vn and host.
	42.67	51.45	8.78	253.3	0.49	0.15	0.16	0.25	Argillite, Andesite Tuff, sheared brecciated q-c vn stkwk, tr-minor tet, cp and ga in veins and host.
	51.45	63.81	12.36	38.3	0.11	0.30	0.10	0.03	Lapilli and Andesite Tuff, minor argillite faulted occ q-c vn/ br tr

									tet in open space filling, minor cp.
91-3	27.49	28.51	1.02	35.4	0.04	0.30	0.10	0.02	Hanging sample, Argillite-minor galena in quartz veins.
	57.90	65.94	8.04	44.2	0.08	0.10	0.11	0.04	Lapilli Tuff-quartz and trace tet filled fractures, tr. Galena, tet.
	71.02	82.30	11.28	43.0	0.07	0.13	0.15	0.03	Lapilli and Andesite Tuff-quartz veinlets t/o-minor diss tet to vns and host.
	87.11	93.95	6.84	57.2	0.12	0.02	0.06	0.01	Lapilli and Andesite Tuf, minor argillie, quartz veined, fractured t/o, sheared in last 1 m.
91-4									Not sampled although several areas of contained q-c veinlets with py and on occ tet noted.
91-5	24.86	26.62	1.76	26.4	0.05	0.03	0.15	0.01	Andesite Tuff, Bx tr tet, cp in carb.
91-6									Tr tet observed in q-c vein 19.24-22.11: not sampled.
91-7									17.45-27.99: argillite with q-c veinlets tr tet in veinlets and rare py-not sampled.
91-8									Clay altered t/o-nothing sampled.
91-9									16.46-27.85: argillite, carb filled fractured.
91-10									6.10-22.81: andesite with abundant quart veining-no sulphide, nothing sampled
91-11	51.15	54.86	3.71	52.3	0.06	0.07	0.08	0.02	Argillite with ank vns, sulphide bearing veins t/o
91-12									Nothing sampled. From 41.17-105.18: argillite with Fe-carb +/- qtz filled fracs, tr py, rare tet.
91-13									59.46-74.32: argillite-rare tet seen in q-c veinlets-nothing sampled
92-1									14.58-30.48: Lapilli Tuff with minor argillite faulted intermittently, q-c filled fractures t/o

									elevated Pb/Zn values t/o
92-2	14.64	24.38	9.78	49.88	0.07	0.15	0.15	0.02	Lapilli tuff, galena and tet diss and within veins, fault at 25.1. Last 3.01 m assays 105.2 gpt Ag, 0.16% Cu 0.12% Pb and 0.14% Zn.
92-3	26.83	33.53	6.70	38.36	0.02	0.27	0.32	0.02	Volcanic sandstone wk c-q vn with tr tet galena. From 32.78-33.53 section assays 173.5 gpt Ag. Prior to 26.83 no sampling.
92-4	29.62	30.48	0.86	>200	0.22	>1.0	0.81	0.10	Hanging Sample-Lapilli Tuff
	30.48	57.91	26.71	7.87	0.01	0.23	0.12	>0.01	Lapilli and Andesite Tuff-minor tet and ga in veinlets.
92-5									35.68-43.65: argillite with wk vn, rare tet-gal in veins-not sampled.
92-6									36.58-49.57: argillite with q-c villed fractures, occ tr tet and ga.

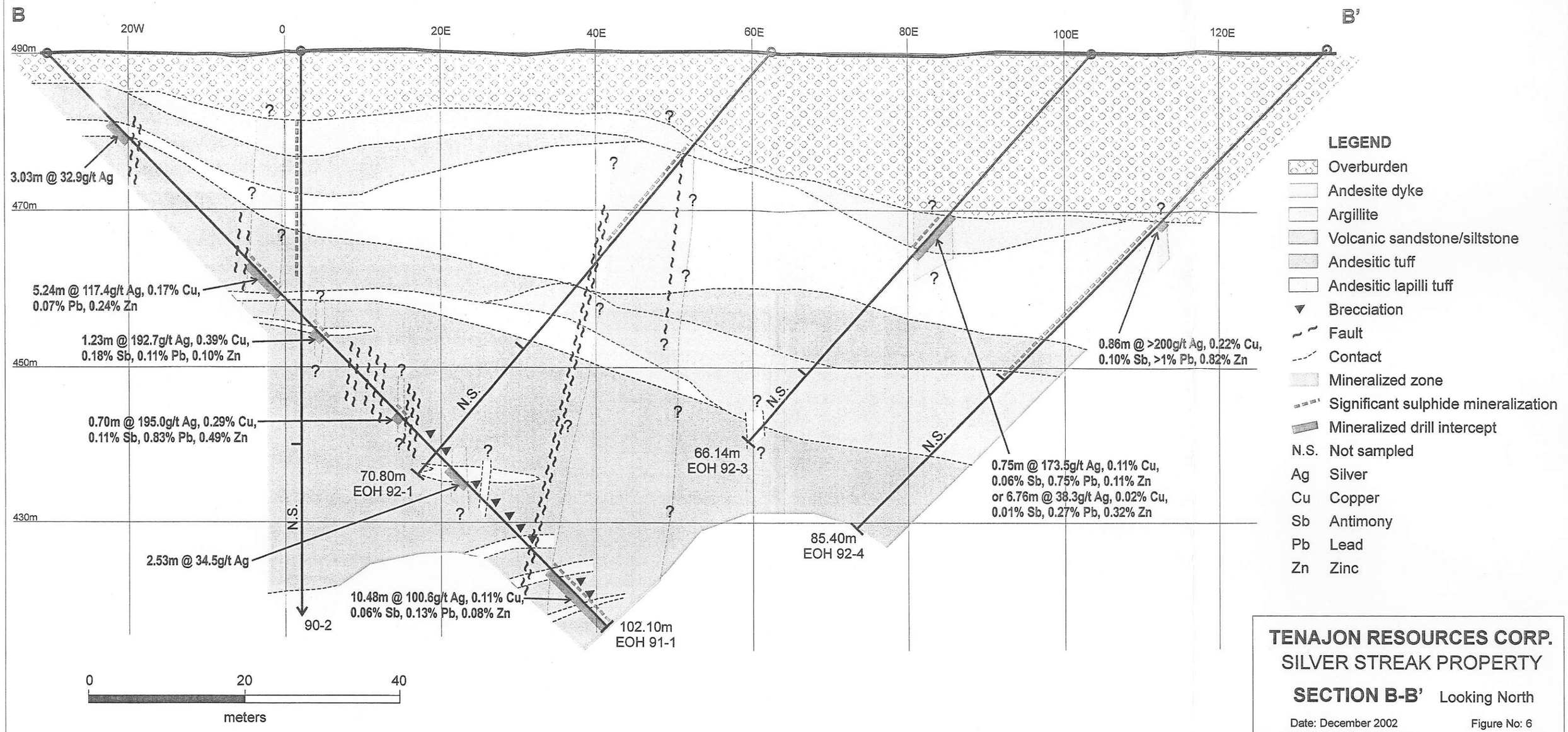
The relationship between the sample length and the true thickness of the mineralization is not known. The Equity Silver drill programs were completed on the basis that the zone shallow dipped to the southeast. Tenajon's interpretation is based on the veins having an overall steep easterly dip. If this is the case the width of the intersections on the holes drilled at –45° would be between 70 and 80% of the zone intercept.

14.0 SAMPLING METHOD AND APPROACH

Continuous chip sampling was completed in the trenches listed in Table 6 by Dave Visagie. All of the samples were taken over measured widths using a nylon tape with the origin point being tied to the existing grid by compass and tape measure. The samples were taken to give a representative at surface value of a selected area. Where possible, the sample lengths employed were dictated by geological contacts and mineralization. The samples were taken using a hammer and chisel. In general, individual samples weighed between 3 and 5 kg.

In the field all of the samples were described, identified and stored in plastic bags. Sample sites were identified using flagging and felt pen. At the end of the program all of the trenches were reclaimed.

The sample descriptions are located in Appendix 1 while the sample results are summarized on Figure 5. The width of the intervals listed on Figure 5 are considered close to be true as the veins are interpreted to be vertical.

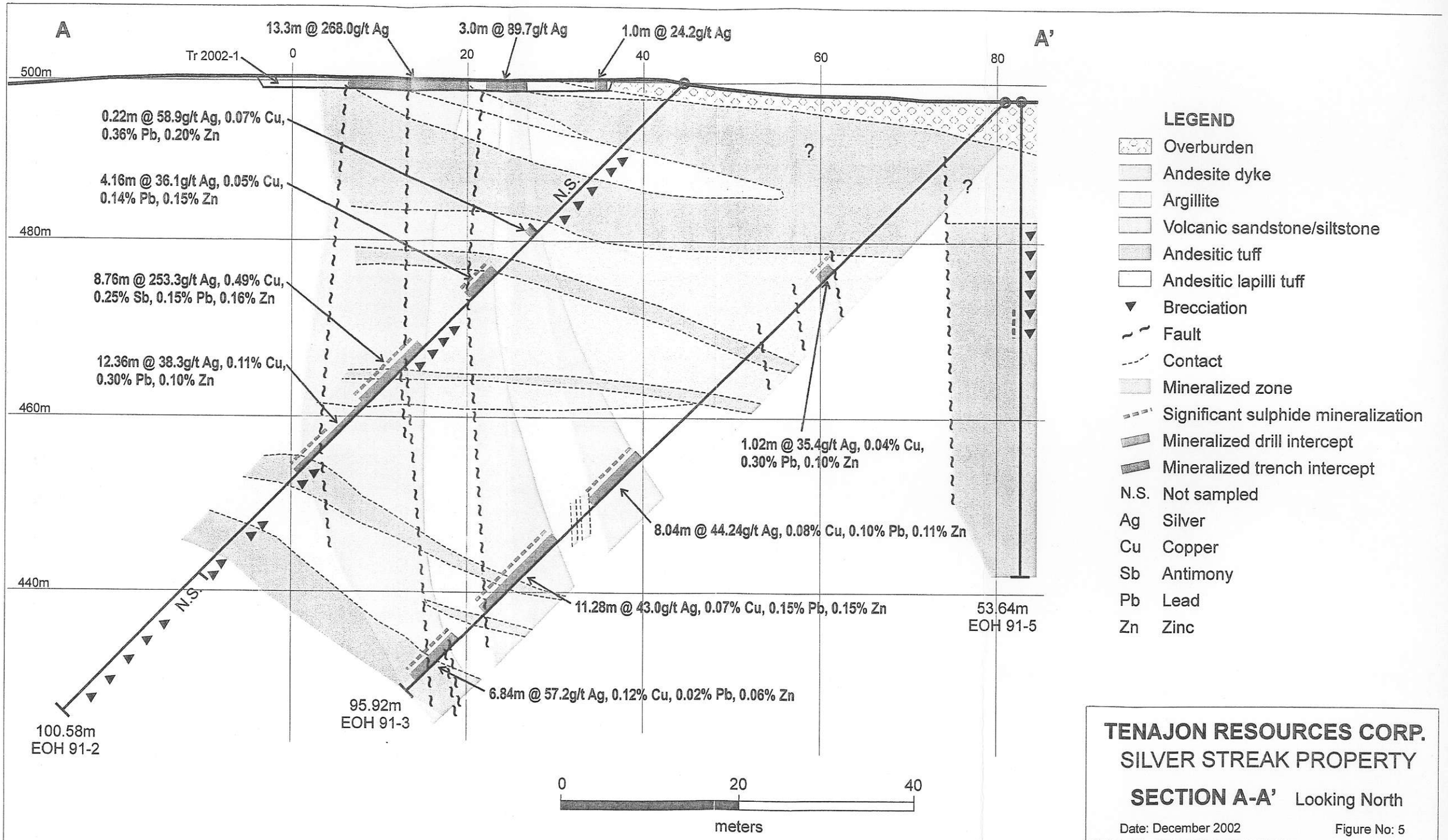


TENAJON RESOURCES CORP.

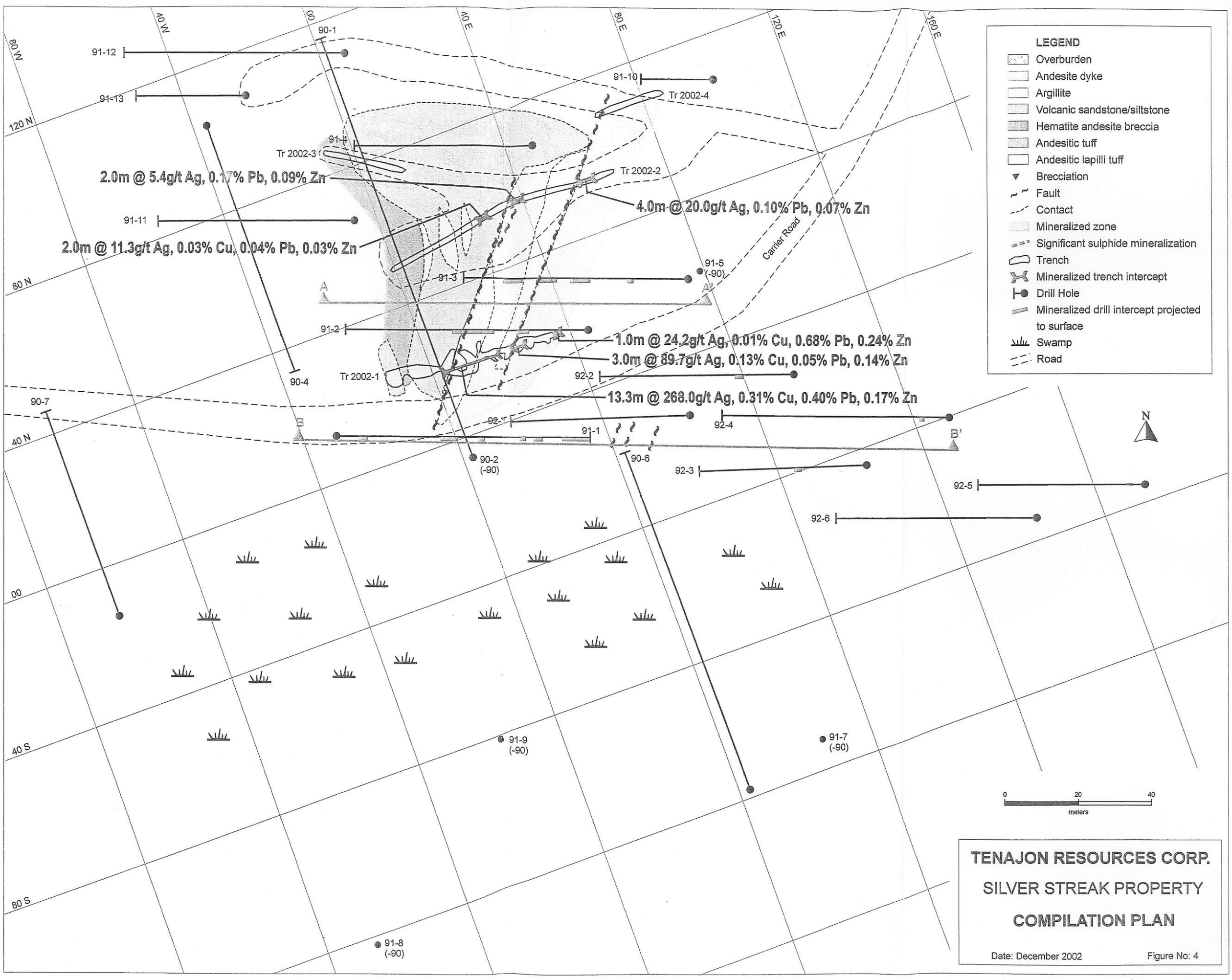
SILVER STREAK PROPERTY

SECTION B-B' Looking North

Date: December 2002 Figure No: 6



TENAJON RESOURCES CORP.
SILVER STREAK PROPERTY
SECTION A-A' Looking North
Date: December 2002 Figure No: 5



TENAJON RESOURCES CORP.

SILVER STREAK PROPERTY

COMPILATION PLAN

Date: December 2002

Figure No: 4

Limited soil spacing was completed along an east-west flagged, compass and hip chained line located at 150N on the old Equity grid. Sample stations were established at 25 metre intervals along the line. B-horizon soil samples were collected at each site using a mattock with the samples being stored in Kraft sample bags. In general, a well-developed B-horizon occurs between 20-30 cm below the surface.

One soil profile was collected from an exposure located above the Carrier Main Showing. Soil profile samples are similar to B-horizon samples with the exception they are collected at varying depths at one site. The objective was to determine whether the soil values increase with depth.

15.0 SAMPLE PREPARATION, ANALYSIS AND SECURITY

None of the samples collected during the 2002 exploration program were prepared by employees, officers, directors or associated of Tenajon Resources Corp.

The assay certificates for the 1990 and 1991 drill programs do not accompany the drill logs however for the 1992 drill program it is known that Min-En Labs, 705 West 15th Street, North Vancouver, B.C. was the lab employed.

All of the samples from the 2002 program were shipped by Bandstra Transportation, Houston B.C. to Acme Analytical Laboratories, 852 East Hastings Street, Vancouver, B.C., V6A 1R6. The author personally sealed and dropped off the samples to Bandstra. He did not however have control over the samples once they left his possession and therefore cannot personally verify if anything happened to the samples from the time they left Houston until the time they were received by Acme. However the author has no reason to believe that sample security was ever compromised.

At Acme, all of the rock and soil samples were analyzed by ICP (Inductively Coupled Plasma). In this method, rock samples are jaw crushed to 70% passing 10 mesh (2mm), a 250 aliquot is then riffle split and pulverized to 95% passing 150 mesh (100um) in a mild-steel ring and puck mill while soils are dried and sieved to –80 mesh. Aliquots of 0.5 gram are weighed into test tubes. QA/QC protocol includes inserting a duplicate of pulp to measure analytical precision, a coarse (10 mesh) rejects duplicate to measure method precision (drill core only), two analytical blanks to measure background and an aliquot of in-house reference material STD DS3 to measure accuracy in each analytical batch of 34 samples.

Aqua Regia, a 2:2:2 mixture of ACS grade concentrated HCL, concentrated HNO_3 and demineralized H_2O, is added to each sample. Samples are digested for one hour in a hot water bath (>95°C). QA/QC protocol requires simultaneous digestion of two regent blanks randomly inserted in each batch.

The sample solutions are aspirated into a Jarrel Ash AtomComp 800 or 975 ICP emission spectrograph to determine the following 30 elements: Ag, Al, As, Au, B, Ba, Bi, Ca, Cd, Co, Cr, Cu, Fe, K, La, Mg, Mn, Mo, Na, Ni, P, Pb, Sb, Sr, Th, Ti, U, V, W, Zn. The leach is partial for Al, B, Ba, Ca, Cr, Fe, K, Mg, Mn, Na, P, Sb, Ti, U and W.

For samples that assayed >100 gpt Ag the samples were jaw crushed and pulverized to –100 mesh (-150 microns). Splits of 30 g are weighed from the pulverized fraction. Each analytical batch (34 samples) has a duplicate pair from a randomly selected material to monitor precision. Reference material (in-house control standards) Au-S, Au-R or FA-100S are added to each batch to monitor accuracy.

The splits are fused at 1000°C for 1 hour. Fire-assay fluxes, containing PbO litharge and Ag inquart, liberate all Au, Pt and Pd. After cooling, lead buttons are recovered and cupelled at 950°C to render Ag +/- Au +/- Pt +/- Pd dore beads. The beads are weighed then leached in 1 mL oc concentrated HNO_3 at >95°C to dissolve Ag leaving Au sponges. Every Ag fire assay is accompanied by a wet assay. Ag concentrations <334 gpt Ag are reported from the wet assay while samples assaying >334 gpt Ag are from the fire assay.

Data is inspected by the Fire Assay Supervisor then undergoes final verification by a British Columbia Certified Assayed who signs the Analytical Report before releasing it to the client. At no stage was preparation or the analysis completed by NDT employees.

In the authors' opinion the sampling, sample preparation, security and analytical precedure was completed to industry standards and as such the data obtained reliable.

16.0 DATA VERIFICATION

Due to the limited number of samples and the preliminary nature of the program there was no effort to enter standards or blanks into the stream of samples. A rigorous quality assurance and quality control program was not undertaken.

17.0 ADJACENT PROPERTIES

Not applicable.

18.0 MINERAL PROCESSING AND METALLURGICAL TESTING

To the author's knowledge none has been completed.

19.0 MINERAL RESOURCE AND MINERAL RESERVE ESTIMATES

Not enough work completed on the property to outline any reserves or resources.

20.0 OTHER RELEVANT DATA AND INFORMATION

The author knows of no other data relevant to the property.

21.0 INTERPRETATIONS AND CONCLUSIONS

The work completed by Tenajon Resources Ltd. and Equity Silver Mines Ltd. has identified eight epithermal style mineralized structures within a 150 m wide trend on the Silver Streak Property. The structures, consisting of quartz veins/stockwork and breccia zones, host varying silver and base metal values ralated to chalcopyrite, galena, sphalerite, tetrahedrite and pyrite mineralization. The control of the vein/stockwork/breccia emplacement appears to be a series of north-northeast, steeply east dipping, faults.

Individual structures are up to 13 metres wide and have been traced through drilling and trenching for up to 70 metres and down-dip for 75 metres. They are all open along strike and down-dip.

Based on the results it is concluded that the work on completed on the Silver Streak Property has demonstrated that it hosts potentially significant silver and base metal values within an epithermal setting. Additional work consisting of drilling along strike is required to further investigate the potential of the property to host significant reserves of silver and base metals.

22.0 RECOMMENDATIONS

It is recommended that an initial drill program totaling 500 m in length be undertaken at the Silver Streak property. Prior to drilling a small grid would be emplaced over the showings and a VLF-EM survey completed in an effort to delineate the hosting structures. The purpose of the drill program would be to test the along strike extensions of the zone. Three holes would be drilled in the south extension of the property, one in the north.

Item	Description	Cost/	Cost
Gridding	400 x 500 metre grid, lines every 50 metres stations at 25 metre intervals	$150/km	$ 1,000
Geophysics	IP, VLF, Mag	$3000/km	$15,000
Drilling	500 metres	$100/m	$50,000
Labour	Geologist -10mandays Labourer-10 mandays	$320/day $200/day	$ 5,200
Room & Board	10 man-days	$120/day	$ 1,200
Truck Rental	10 days	$100/day	$ 1,000
Airfare	Vancouver-Smithers Return		$ 800
Consumables			$ 1,000
Assaying	200 samples	$25/sample	$ 5,000
Preparation			$ 600
Report			$ 3,000
		Sub Total	**$83,800**
Contingency		10%	$ 7,880
		Total	**$92,180**

Contingent upon results additional drilling would be undertaken.

23.0 BIBLIOGRAPHY

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24.0 DATE

This report was completed on January 14, 2003 in Vancouver, British Columbia.

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Dave Visagie, P. Geo.
Senior Geologist
The Northair Group

25.0 STATEMENT OF QUALIFICATIONS

David A Visagie, B.Sc
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V6C 2T6
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E-Mail: visagie@northair.com

I, David A Visagie, do hereby certify that:

- I am currently employed as a Senior Geologist by:

International Northair Mines Ltd.
860-625 Howe Street,
Vancouver, B.C.
V6C 2T6

- I graduated from the University of British Columbia in 1976 with a Bachelor of Science Degree Majoring in Geology.
- I am a member of the Association of Professional Engineers and Geoscientist of B.C. (#19520).
- I have been continuously employed within the mining industry since that time.
- I have read the definition of "qualified person" as set out in National Instrument 43-101 (NI 43-101) and certify that by reason of my education, affiliation with a professional association (as defined in NI 43-101) and past relevant work experience, I fulfill the requirements to be a "qualified person" for the purposes of NI 43-101.
- I am responsible for the preparation of the technical report titled "Qualifying Report for the Silver Streak Property" dated January 14, 2003.
- Prior to writing of this report I was directly involved with the property having completed approximately 4 days on the property completing the evaluation of the Silver Streak Property for Tenajon Resources Corp.
- I am not aware of any material fact of material change with respect to the subject matter of the Technical Report this is not reflected in the Technical Report, the omission to disclose which make the Technical Report misleading.
- I am not independent of the issues applying all of the tests in section 1.5 of National Instrument 43-101 as I am an employee of Tenajon Resources Corp..

- I have read National Instrument 43-101 and Form 43-101F and the Technical Report has been prepared in compliance with that instrument and form.

- I consent to the filing of the Technical Report with any stock exchange and other regulatory authority and any publication by them, including electronic publication in the public company files on their websites accessible by the public, of the Technical Report.

Dated the 14th day of January, 2003.

_____

Dave Visagie, P. Geo.

From ACME ANALYTICAL LABORATORIES LTD. 852 E. HASTINGS ST. VANCOUVER BC V6A 1R6 PHONE(604)253-3158 FAX(604)253-1716 @

To Tenajon Resources Corp.

Acme file # A205337 Page 1 Received: DEC 5 2002 * 65 samples in this disk file.

Analysis: GROUP 1D - 0.50 GM

ELEMENT	From	To	Int		Cu	Pb	Zn	Ag	Fe	As	Sb	Ca	Mg	Ba	Al	K	Au
SAMPLES	(M)	(M)	(M)		ppm	ppm	ppm	ppm	%	ppm	ppm	%	%	ppm	%	%	ppb
SI					1	< 3	2	< .3	0.01	5	< 3	0.08	< .01	1	< .01	< .01	
Tr 2002-1																	
E 182915	12.0W	13.0W	1	andesite tuff, fg, green/grey clay alt'd barren	8	9	295	< .3	3.89	5	< 3	3.98	0.81	627	0.68	0.29	
E 182914	10.0W	12.0W	2	andesite tuff, fg, green/grey clay alt'd barren	10	7	252	< .3	3.04	5	3	4.38	0.96	1443	0.63	0.26	
E 182913	8.0W	10.0W	2	andesite tuff, fg, green/grey clay alt'd barren	5	15	459	< .3	4.07	3	< 3	6.05	0.96	778	0.57	0.27	
E 182912	6.0W	8.0W	2	hematite rich andesite tuff, carb vn t/o, barren	131	43	317	9.6	4.26	24	42	4.19	1.28	235	0.75	0.23	
E 182911	4.0W	6.0W	2	andesite tuff, fg, green/grey clay alt'd barren	10	35	167	< .3	3.96	12	5	2.14	0.39	410	0.58	0.22	
E 182910	2.0W	4.0W	2	andesite tuff, fg, green/grey clay alt'd barren	20	17	254	0.4	3.45	7	4	0.45	0.2	2722	0.63	0.14	
E 182909	0	2.0W	2	andesite tuff, fg, green/grey clay alt'd barren	15	27	402	< .3	2.92	8	3	1.84	0.44	391	0.80	0.18	
E 182201	0	2.0E	2	andesite tuff, fg, green/grey clay alt'd barren	49	175	596	0.3	3.91	7	<3	3.51	0.28	956	0.51	0.23	10
E 182202	2.0E	4.0E	2	andesite tuff, fg, green/grey clay alt'd barren	4	47	493	0.3	2.9	4	<3	3.82	0.53	303	0.43	0.2	3
E 182203	4.0E	6.0E	2	andesite tuff, fg, green/grey clay alt'd barren	90	50	461	4.6	2.74	16	10	3.78	0.43	259	0.37	0.18	5
E 182204	6.0E	8.0E	2	andesite lapilli tuff, pale grey/beige with white lapilli frags to 1/2 cm, erratic q vn with tet, ga, sp, mal stain	3497	1916	1713	245.3	2.77	502	1161	4.69	1.59	393	0.21	0.09	2
E 182205	8.0E	10.0E	2	as above	2274	6361	3885	205	5.15	316	696	8.08	2.71	344	0.20	0.08	3
E 182206	10.0E	12.0E	2	barren lapilli tuff	152	297	2197	12.5	5.37	38	20	10.4	3.66	483	0.26	0.11	<2

E 182207	12.0E	14.0E	2	argillite, wk stockwork, extensive gossan along fractures, tet, ga, sp, cp in veins and along fracture faces, extensive mal, az stain.	4294	2545	1069	392.6	1.54	687	976	2.15	0.45	573	0.3	0.12	<2
E 182208	14.0E	16.0E	2	as above	4530	8581	1054	363.5	2.23	783	917	2.72	0.97	555	0.29	0.11	<2
E 182209	16.0E	18.0E	2	as above	3106	3989	501	348.2	1.56	634	368	0.39	0.08	379	0.26	0.11	2
E 182210	18.0E	19.3E	1	as above	3909	1399	689	335.1	1.75	624	1062	1.4	0.37	629	0.20	0.08	7
E 182211	19.3E	21.0E	2	andesite lapilli tuff, barren	84	58	484	6.7	3.11	21	8	4.38	0.41	229	0.41	0.2	<2
E 182212	21.0E	22.7E	2	argillite with wk gossan	1704	324	1547	127.5	5.65	442	172	3.76	0.34	370	0.42	0.17	3
E 182901	22.7E	24.0E	1	argillite with wk gossan along fractures	575	811	1267	40.2	4.21	205	156	0.49	0.09	235	0.47	0.25	
E 182902	24.0E	26.0E	2	argillite with wk gossan along fractures	111	1567	1635	10.6	3.93	93	36	0.56	0.12	188	0.40	0.21	
E 182903	26.0E	28.0E	2	argillite with wk gossan along fractures	90	1155	1780	7.7	4.03	104	47	0.8	0.1	284	0.48	0.24	
E 182904	28.0E	30.0E	2	argillite with wk gossan along fractures	105	920	1646	9.7	3.84	100	40	0.41	0.1	373	0.41	0.23	
E 182905	30.0E	32.0E	2	argillite with wk gossan along fractures	76	915	1244	5.2	4.83	108	27	0.31	0.1	156	0.50	0.24	
E 182906	32.0E	34.0E	2	argillite with wk gossan along fractures	106	1403	1182	8.8	5.08	111	32	0.42	0.29	189	0.85	0.2	
E 182907	34.0E	36.0E	2	argillite with wk gossan along fractures	59	975	1390	6.6	4.11	78	21	0.35	0.08	388	0.45	0.23	
E 182908	36.0E	37.0E	1	argillite with wk gossan, minor quartz veining with minor galena.	132	6726	2358	24.2	4.68	67	58	0.75	0.12	575	0.47	0.22	
Tr 2002-2																		
E 182916	0	2.0E	2	hematite rich andesite tuff, wk gossan along fractures	6	18	132	< .3	3.4	5	< 3	4.44	1.12	137	0.50	0.21	
E 182917	2.0E	4.0E	2	hematite rich andedsite tuff, extensive carbonate viening-barren	3	14	145	< .3	2.81	5	< 3	4.4	1.37	170	0.43	0.21	

E 182918	4.0E	5.5E	2	argillite, fg, black, wk limonite stain	6	11	288	< .3	3.5	4	< 3	4.96	1.65	945	0.46	0.22	
E 182919	5.5E	7.8E	2	argillite, fg, black, wk limonite stain	166	60	251	5	3.95	68	54	2.33	1.42	435	0.51	0.27	
E 182920	7.8E	10.0E	2	hematite rich andesite tuff, wk gossan along fracture faces	2.5	10.5	39.5	0	1.765	3.5	4	0.39	0.18	209	0.31	0.14	
					5	21	79	< .3	3.53	7	8	0.78	0.36	418	0.61	0.28	
E 182921	10	11.3E	1	hematite rich andesite tuff, wk gossan along fracture faces	5	23	85	< .3	3.85	9	6	0.67	0.6	592	0.79	0.31	
E 182922	11.3	13.4E	2	argillite, fg, black, wk limonite stain	113	16	112	0.8	2.96	39	32	1.15	0.9	226	0.89	0.21	
E 182923	13.4	15.1E	2	andesite lapilli tuff, pale grey green, barren	249	55	154	2.6	3.21	70	71	0.87	0.74	133	0.77	0.28	
E 182924	15.1	17.0E	2	argillite with minor fg black sulphides	83	601	259	2.6	3.44	87	24	1.28	0.52	773	0.71	0.25	
E 182925	18.8	21.0E	2	argillite with minor fg black sulphides	69	55	131	0.7	3.62	64	18	2.52	0.28	677	0.60	0.23	
E 182926	21.0E	23.0E	2	andesite lapilli tuff, pale grey green, barren	12	12	148	< .3	4.08	14	5	2.16	0.9	390	0.68	0.26	
E 182927	23.0E	25.0E	2	argillite with minor fg black sulphides	142	72	118	1.8	3.48	59	30	0.27	0.11	125	0.51	0.23	
E 182928	25.0E	27.0E	2	andesite lapilli tuff, pale grey green, gossanous t/o	56	74	256	1.4	6.38	57	16	0.48	0.22	160	0.74	0.24	
E 182929	27.0E	29.0E	2	andesite lapillit tuff, pale grey/green, gossanous t/o	276	430	290	11.3	4.79	82	73	0.42	0.2	195	0.78	0.2	
E 182930	29.0E	31.0E	2	andesite lapillit tuff, pale grey/green, gossanous t/o	146	18	208	1.4	5.98	50	36	0.62	0.16	171	0.79	0.23	
E 182931	31.0E	33.0E	2	argillite, fg, black, wk limonite stain	67	20	129	0.4	4.71	26	15	0.53	0.24	209	0.88	0.2	
E 182932	33.0E	35.0E	2	argillite, fg, black, wk limonite stain, gouge on occ	144	122	331	5.5	5.88	42	26	0.55	0.21	189	0.72	0.21	

E 182933	35.0E	37.0E	2	argillite, fg, black, wk limonite stain, gouge on occ	59	152	211	2.2	4.69	21	10	0.46	0.19	175	0.82	0.21	
E 182934	37.0E	39.0E	2	argillite, fg, black, wk limonite stain, gouge on occ	6	13	205	< .3	5.9	12	4	0.5	0.14	202	0.83	0.27	
E 182935	39.0E	41.0E	2	argillite, fg, black, wk limonite stain, gouge on occ	30	8	142	0.4	3.72	5	< 3	0.29	0.11	120	0.59	0.23	
E 182936	41.0E	43.0E	2	argillite, as above	39	48	288	0.7	3.7	12	5	0.27	0.09	111	0.54	0.22	
E 182937	43.0E	45.0E	2	argillite, as above	26	448	419	1.6	3.6	42	5	0.48	0.21	217	0.64	0.24	
E 182938	45.0E	47.0E	2	argillite, as above minor ga on fracture faces	101	1649	896	5.4	6.68	14	27	0.63	0.15	441	0.6	0.21	
E 182939	47.0E	49.0E	2	argillite, fg, black, wk limionite stain	93	808	707	4.9	5.26	24	24	0.58	0.2	245	1.06	0.28	
E 182951	49.0E	52.0E	3	andesite lapilli tuff, as previous	21	12	241	0.3	3.97	5	3	5.3	0.65	609	0.86	0.28	
E 182952	52.0E	54.0E	2	andesite lapilli tuff	37	32	357	1.6	4.16	25	9	3.55	0.57	379	0.7	0.27	
E 182953	54.0E	56.0E	2	andesite lapilli tuff	37	22	165	0.3	4.22	6	< 3	4.02	1.02	484	0.58	0.3	
E 182954	56.0E	58.0E	2	andesite lapillit tuff, pale grey/green, gossanous t/o	59	612	486	4.3	3.79	13	13	1.44	0.21	133	0.58	0.27	
E 182955	58.0E	60.0E	2	argillite with wk gossan, minor fg black sulphides	95	948	866	12.8	4.96	63	14	0.92	0.34	502	0.83	0.21	
E 182956	60.0E	62.0E	2	as above	204	994	670	28.4	4.51	188	71	0.33	0.4	268	0.72	0.22	
E 182957	62.0E	64.0E	2	as above	49	498	592	7.1	3.83	102	16	0.24	0.11	162	0.49	0.21	
E 182958	64.0E	66.0E	2	as above	54	622	506	7.9	4.13	93	15	0.26	0.13	176	0.55	0.22	
E 182969	66.0E	68.0E	2	as above	51	209	396	6.5	3.63	98	13	0.26	0.13	114	0.59	0.21	
Tr 2002-3																	
E 182960	0	3.0E	3	argillite, highly sheared	84	65	365	0.7	4.12	187	10	0.3	0.1	91	0.61	0.27	
RE E 1829	0	3.0E	3	argillite, highly sheared	83	64	360	0.7	4.06	184	9	0.29	0.1	87	0.59	0.26	
E 182961	3.0E	6.0E	3	argillite, highly sheared	49	62	176	1.2	4.4	112	8	0.28	0.1	169	0.48	0.2	
E 182962	6.0E	9.0E	3	turbiditic argillite	31	225	284	0.8	3.93	59	6	1.13	0.34	150	0.73	0.21	
E 182963	9.0E	12.0E	3	turbiditic argillite	35	140	356	2	3.98	91	7	0.58	0.28	179	0.43	0.23	
E 182964	12.0E	15.0E	3	turbiditic argillite	64	82	311	2.6	4.95	121	8	0.26	0.1	415	0.42	0.22	

E 182965	15.0E	17.0E	2	turbiditic argillite	40	137	191	2	4.29	95	6	0.35	0.18	159	0.65	0.23	
Tr 2002-4																	
E 182966	0	2.0E	2	hematite andesite breccia tuff, ext carb vn	15	19	94	< .3	4.28	8	< 3	0.52	0.18	209	1.01	0.2	
E 182967	2.0E	4.0E	2	as above	12	14	90	< .3	4.17	7	< 3	1.62	0.47	172	0.62	0.24	
E 182968	4.0E	6.0E	2	as above	11	20	197	< .3	4.91	2	3	4	1.48	565	0.48	0.17	
E 182969	6.0E	8.0E	2	as above	65	14	77	< .3	4.4	11	< 3	0.75	0.19	179	0.65	0.23	
E 182970	8.0E	10.0E	2	as above	22	17	73	< .3	4.43	6	< 3	0.52	0.2	179	0.7	0.16	
E 182971	10.0E	12.0E	2	barren andesite lapilli tuff	7	27	112	< .3	3.91	4	< 3	4.96	2.38	110	0.45	0.15	
E 182972	12.0E	14.0E	2	barren andesite lapilli tuff	12	16	76	< .3	5.12	8	< 3	0.46	0.19	148	0.82	0.15	
E 182973	14.0E	16.0E	2	barren andesite lapilli tuff	5	9	71	< .3	3.84	7	< 3	2.48	1.06	183	0.52	0.18	
STANDARD DS4					122	31	160	< .3	3.21	22	5	0.52	0.6	144	1.69	0.16	

From ACME ANALYTICAL LABORATORIES LTD. 852 E. HASTINGS ST. VANCOUVER BC V6A 1R6 PHONE(604)253-3158 FAX(604)253-1716 @ CSV TEXT FORMAT

To Tenajon Resources Corp.

Acme file # A205338 Received: DEC 5 2002 * 12 samples in this disk file.

Analysis: GROUP 1D - 0.50 GM

ELEMENT	Location	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	W
SAMPLES		ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	%	ppm	ppm	%	ppm	%	ppm	%	%	%	ppm
G-1		1	3	3	40	< .3	5	4	491	1.9	< 2	< 8	< 2	5	81	< .5	< 3	< 3	39	0.5	0.1	6	15	0.5	235	0.1	< 3	1	0.1	0.5	< 2
E 182974	150N/050	1	18	7	103	< .3	28	11	961	2.8	5	< 8	< 2	< 2	23	0.6	< 3	< 3	60	0.4	0.1	7	25	0.7	253	0.1	< 3	1.7	< .01	0.1	< 2
E 182975	150N/025	< 1	8	5	62	< .3	13	6	686	1.8	< 2	< 8	< 2	< 2	24	< .5	< 3	< 3	40	0.3	0	7	16	0.3	169	0	< 3	1.1	0	0	< 2
E 182976	150N/0	1	33	11	111	0.3	37	.11	1622	3.1	2	< 8	< 2	2	57	< .5	< 3	< 3	57	0.7	0.1	32	32	0.6	415	0	< 3	2.5	0	0.1	< 2
E 182977	150N/025	< 1	8	7	69	< .3	17	6	501	2	< 2	< 8	< 2	< 2	20	< .5	< 3	< 3	47	0.2	0	6	17	0.3	162	0.1	< 3	1.1	0	0	< 2
E 182978	150N/050	1	6	9	64	< .3	12	7	308	2.3	2	< 8	< 2	2	14	< .5	< 3	< 3	48	0.2	0.2	4	17	0.2	158	0	< 3	1.1	0	0	< 2
E 182979	150N/075	< 1	8	4	52	< .3	12	6	342	1.9	3	< 8	< 2	< 2	20	< .5	< 3	< 3	45	0.2	0	7	17	0.3	119	0.1	< 3	1	0	0	< 2
E 182980	150N/100	1	9	6	51	< .3	11	7	237	2.2	4	< 8	< 2	< 2	18	< .5	< 3	< 3	51	0.3	0.1	4	17	0.3	87	0	< 3	1.1	0	0	< 2
E 182981	150N/125	< 1	9	5	57	0.4	17	7	542	2.1	< 2	< 8	< 2	< 2	24	< .5	< 3	< 3	44	0.4	0.1	3	18	0.3	110	0	< 3	1.1	0	0.1	< 2
E 182982	150N/150	< 1	7	4	49	< .3	10	6	640	2	2	< 8	< 2	< 2	24	< .5	< 3	< 3	46	0.3	0.1	4	17	0.2	135	0	< 3	0.9	0	0.1	< 2
E 182983	150N/175	< 1	6	3	49	< .3	13	6	531	2	2	< 8	< 2	< 2	23	< .5	< 3	< 3	44	0.3	0.1	5	18	0.3	115	0	< 3	0.9	0	0.1	< 2
STANDARD DS4		7	120	31	149	< .3	34	12	756	3.1	23	< 8	< 2	4	26	5.4	5	5	72	0.5	0.1	16	158	0.6	138	0.1	< 3	1.6	0	0.1	4

ACME ANALYTICAL LABORATORIES LTD. ... (ISO 9002 Accredited Co.)

GEOCHEMICAL ANALYSIS CERTIFICATE

Tenajon Resources Corp. File # A205337 Page 1

860 - 625 Howe St., Vancouver BC V6C 2T6

SAMPLE#	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Ni ppm	Co ppm	Mn ppm	Fe %	As ppm	U ppm	Au ppm	Th ppm	Sr ppm	Cd ppm	Sb ppm	Bi ppm	V ppm	Ca %	P %	La ppm	Cr ppm	Mg %	Ba ppm	Ti %	B ppm	Al %	Na %	K %	W ppm
SI	<1	1	<3	2	<.3	<1	<1	3	.01	5	<8	<2	<2	2	<.5	<3	<3	<1	.08	<.001	<1	1	<.01	1	<.01	<3	<.01	.41	<.01	<2
E 182901	6	575	811	1267	40.2	64	26	903	4.21	205	<8	<2	2	39	12.8	156	<3	27	.49	.072	5	11	.09	235	<.01	6	.47	.01	.25	<2
E 182902	3	111	1567	1635	10.6	56	21	582	3.93	93	<8	<2	2	44	11.6	36	<3	27	.56	.068	4	11	.12	188	<.01	5	.40	.01	.21	<2
E 182903	2	90	1155	1780	7.7	64	23	591	4.03	104	<8	<2	<2	61	16.3	47	<3	25	.80	.071	3	11	.10	284	<.01	6	.48	.01	.24	2
E 182904	3	105	920	1646	9.7	58	20	482	3.84	100	<8	<2	2	33	12.5	40	<3	23	.41	.067	3	10	.10	373	<.01	4	.41	.01	.23	<2
E 182905	3	76	915	1244	5.2	63	19	732	4.83	108	<8	<2	2	28	8.0	27	<3	30	.31	.073	4	12	.10	156	<.01	6	.50	.01	.24	<2
E 182906	3	106	1403	1182	8.8	58	22	629	5.08	111	<8	<2	2	35	2.5	32	<3	51	.42	.078	6	19	.29	189	.01	5	.85	.02	.20	<2
E 182907	5	59	975	1390	6.6	56	18	752	4.11	78	<8	<2	2	26	8.2	21	<3	28	.35	.066	3	16	.08	388	<.01	4	.43	.01	.23	<2
E 182908	5	132	6726	2358	24.2	42	17	1177	4.68	67	<8	<2	2	39	19.3	58	<3	35	.75	.097	4	15	.12	575	<.01	4	.47	.01	.22	5
E 182909	1	15	27	402	<.3	39	14	821	2.92	8	<8	<2	<2	48	2.1	3	<3	60	1.84	.091	6	17	.44	391	<.01	4	.80	.01	.18	<2
E 182910	1	20	17	254	.4	36	18	1107	3.45	7	<8	<2	<2	64	1.1	4	<3	60	.45	.066	6	16	.20	2722	.01	4	.63	.02	.14	<2
E 182911	<1	10	35	167	<.3	39	21	961	3.96	12	<8	<2	2	49	1.0	5	<3	73	2.14	.121	10	34	.39	410	.01	6	.58	.01	.22	<2
E 182912	1	131	43	317	9.6	60	26	1219	4.26	24	<8	<2	<2	117	3.2	42	<3	89	4.19	.113	9	45	1.28	235	.01	7	.75	.05	.23	<2
E 182913	<1	5	15	459	<.3	40	19	1521	4.07	3	<8	<2	<2	91	2.4	<3	<3	87	6.05	.084	9	16	.96	778	.01	8	.57	.01	.27	<2
E 182914	<1	10	7	252	<.3	40	19	1145	3.04	5	<8	<2	2	92	1.4	3	<3	50	4.38	.129	7	19	.96	1443	.01	7	.63	.01	.26	<2
E 182915	1	8	9	295	<.3	44	22	1273	3.89	5	<8	<2	2	71	1.9	<3	<3	55	3.98	.123	9	35	.81	627	.01	9	.68	.01	.29	<2
E 182916	<1	6	18	132	<.3	36	17	849	3.40	5	<8	<2	2	83	.7	<3	<3	60	4.44	.114	7	35	1.12	137	.03	8	.50	.01	.21	<2
E 182917	1	3	14	145	<.3	35	16	854	2.81	5	<8	<2	<2	90	.6	<3	<3	68	4.40	.105	9	28	1.37	170	.01	6	.43	.01	.21	<2
E 182918	<1	6	11	288	<.3	46	21	1186	3.50	4	<8	<2	<2	110	1.5	<3	<3	57	4.96	.107	8	18	1.65	945	<.01	6	.46	.01	.22	<2
E 182919	1	166	60	251	5.0	57	24	939	3.95	68	<8	<2	3	74	1.4	54	<3	23	2.33	.069	5	10	1.42	435	<.01	5	.51	.01	.27	<2
E 182920	1	5	20	81	<.3	22	9	351	3.60	10	<8	<2	4	43	<.5	9	<3	42	.82	.071	10	28	.37	431	.01	6	.61	.02	.30	<2
RE E 182920	1	5	21	79	<.3	25	8	336	3.53	7	<8	<2	4	41	<.5	8	<3	42	.78	.069	9	27	.36	418	<.01	5	.61	.01	.28	<2
E 182921	1	5	23	85	<.3	27	10	350	3.85	9	<8	<2	4	52	<.5	6	<3	50	.67	.079	8	27	.60	592	.01	6	.79	.02	.31	<2
E 182922	1	113	16	112	.8	32	14	640	2.96	39	<8	<2	3	61	<.5	32	<3	50	1.15	.091	9	15	.90	226	.01	4	.89	.03	.21	<2
E 182923	1	249	55	154	2.6	34	13	532	3.21	70	<8	<2	4	52	.9	71	<3	37	.87	.083	12	13	.74	133	<.01	5	.77	.02	.28	<2
E 182924	2	83	601	259	2.6	60	29	743	3.44	87	<8	<2	2	67	2.1	24	<3	34	1.28	.077	6	12	.52	773	.01	5	.71	.02	.25	<2
E 182925	2	69	55	131	.7	48	21	943	3.62	64	<8	<2	3	65	.6	18	<3	24	2.52	.065	8	10	.28	677	<.01	5	.60	.01	.23	<2
E 182926	1	12	12	148	<.3	58	22	833	4.08	14	<8	<2	3	66	.5	5	<3	55	2.16	.114	11	20	.90	390	<.01	7	.68	.02	.26	<2
E 182927	3	142	72	118	1.8	43	16	487	3.48	59	<8	<2	3	29	.5	30	<3	24	.27	.066	9	10	.11	125	<.01	5	.51	<.01	.23	<2
E 182928	3	56	74	256	1.4	73	25	1132	6.38	57	<8	<2	3	43	1.2	16	<3	61	.48	.118	10	25	.22	160	<.01	6	.74	.02	.24	<2
E 182929	5	276	430	290	11.3	63	23	1191	4.79	82	<8	<2	2	41	1.4	73	<3	59	.42	.108	10	21	.20	195	.01	5	.78	.02	.20	<2
E 182930	4	146	18	208	1.4	77	27	1030	5.98	50	<8	<2	2	48	.9	36	<3	66	.62	.176	10	28	.16	171	.01	7	.79	.01	.23	<2
E 182931	4	67	20	129	.4	57	22	993	4.71	26	<8	<2	2	48	.5	15	<3	62	.53	.137	10	22	.24	209	.01	6	.88	.02	.20	<2
E 182932	3	144	122	331	5.5	60	20	852	5.88	42	<8	<2	3	39	1.5	26	<3	66	.55	.140	9	26	.21	189	.01	7	.72	.01	.21	<2
E 182933	2	59	152	211	2.2	58	21	521	4.69	21	<8	<2	2	40	.9	10	<3	62	.46	.119	8	21	.19	175	<.01	6	.82	.01	.21	<2
STANDARD DS4	6	120	29	152	<.3	33	11	752	3.04	24	<8	<2	3	26	5.2	6	5	72	.50	.088	15	154	.57	143	.07	<3	1.63	.03	.13	6

GROUP 1D - 0.50 GM SAMPLE LEACHED WITH 3 ML 2-2-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR, DILUTED TO 10 ML, ANALYSED BY ICP-ES.
UPPER LIMITS - AG, AU, HG, W = 100 PPM; MO, CO, CD, SB, BI, TH, U & B = 2,000 PPM; CU, PB, ZN, NI, MN, AS, V, LA, CR = 10,000 PPM.
ASSAY RECOMMENDED FOR ROCK AND CORE SAMPLES IF CU PB ZN AS > 1%, AG > 30 PPM & AU > 1000 PPB
- SAMPLE TYPE: ROCK R150 60C Samples beginning 'RE' are Reruns and 'RRE' are Reject Reruns.

DATE RECEIVED: DEC 5 2002 DATE REPORT MAILED: Dec 16/02 SIGNED BY: D. TOYE, C.LEONG, J. WANG; CERTIFIED B.C. ASSAYERS

Data FA

ACME ANALYTICAL LABORATORIES LTD. 852 E. HASTINGS ST. VANCOUVER BC V6A 1R6 PHONE(604)253-3158 FAX(604)

(ISO 02 Accredited Co.)

GEOCHEMICAL ANALYSIS CERTIFICATE

Tenajon Resources Corp. File # A205338

860 - 625 Howe St., Vancouver BC V6C 2T6

SAMPLE#	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Ni ppm	Co ppm	Mn ppm	Fe %	As ppm	U ppm	Au ppm	Th ppm	Sr ppm	Cd ppm	Sb ppm	Bi ppm	V ppm	Ca %	P %	La ppm	Cr ppm	Mg %	Ba ppm	Ti %	B ppm	Al %	Na %	K %	W ppm
G-1	1	3	3	40	<.3	5	4	491	1.89	<2	<8	<2	5	81	<.5	<3	<3	39	.51	.084	6	15	.53	235	.11	<3	.96	.09	.47	<2
E 182974	1	18	7	103	<.3	28	11	961	2.75	5	<8	<2	<2	23	.6	<3	<3	60	.44	.110	7	25	.69	253	.06	<3	1.67	<.01	.13	<2
E 182975	<1	8	5	62	<.3	13	6	686	1.79	<2	<8	<2	<2	24	<.5	<3	<3	40	.34	.039	7	16	.28	169	.04	<3	1.06	.01	.03	<2
E 182976	1	33	11	111	.3	37	11	1622	3.09	2	<8	<2	2	57	<.5	<3	<3	57	.72	.098	32	32	.55	415	.01	<3	2.45	.02	.09	<2
E 182977	<1	8	7	69	<.3	17	6	501	2.03	<2	<8	<2	<2	20	<.5	<3	<3	47	.21	.034	6	17	.31	162	.06	<3	1.08	.01	.03	<2
E 182978	1	6	9	64	<.3	12	7	308	2.26	2	<8	<2	2	14	<.5	<3	<3	48	.16	.156	4	17	.24	158	.04	<3	1.12	.01	.04	<2
E 182979	<1	8	4	52	<.3	12	6	342	1.93	3	<8	<2	<2	20	<.5	<3	<3	45	.23	.033	7	17	.31	119	.05	<3	1.00	.01	.03	<2
E 182980	1	9	6	51	<.3	11	7	237	2.21	4	<8	<2	<2	18	<.5	<3	<3	51	.26	.076	4	17	.27	87	.02	<3	1.05	.01	.03	<2
E 182981	<1	9	5	57	.4	17	7	542	2.06	<2	<8	<2	<2	24	<.5	<3	<3	44	.38	.066	3	18	.30	110	.02	<3	1.07	.01	.05	<2
RE E 182981	1	9	3	59	<.3	18	7	564	2.12	3	<8	<2	<2	27	<.5	<3	<3	45	.41	.070	4	19	.31	117	.02	<3	1.11	.01	.05	<2
E 182982	<1	7	4	49	<.3	10	6	640	2.00	2	<8	<2	<2	24	<.5	<3	<3	46	.30	.075	4	17	.24	135	.04	<3	.90	.01	.07	<2
E 182983	<1	6	3	49	<.3	13	6	531	1.97	2	<8	<2	<2	23	<.5	<3	<3	44	.25	.070	5	18	.27	115	.04	<3	.94	.01	.06	<2
STANDARD DS4	7	120	31	149	<.3	34	12	756	3.08	23	<8	<2	4	26	5.4	.5	5	72	.50	.092	16	158	.58	138	.07	<3	1.63	.04	.14	4

GROUP 1D - 0.50 GM SAMPLE LEACHED WITH 3 ML 2-2-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR, DILUTED TO 10 ML, ANALYSED BY ICP-ES.
UPPER LIMITS - AG, AU, HG, W = 100 PPM; MO, CO, CD, SB, BI, TH, U & B = 2,000 PPM; CU, PB, ZN, NI, MN, AS, V, LA, CR = 10,000 PPM.
- SAMPLE TYPE: SOIL SS80 60C Samples beginning 'RE' are Reruns and 'RRE' are Reject Reruns.

DATE RECEIVED: DEC 5 2002 **DATE REPORT MAILED:** Dec 13/02 **SIGNED BY.** D. TOYE, C.LEONG, J. WANG; CERTIFIED B.C. ASSAYERS

Date 12- FA ___

Tenajon Resources Corp. File # A203359R
806 - 625 Howe St., Vancouver, BC V6C 2T6

SAMPLE#	Ag** gm/mt
E 182204	245.3
E 182205	205.0
E 182207	392.6
E 182208	363.5
E 182209	348.2
E 182210	335.7
E 182212	127.1

AG** BY FIRE ASSAY FROM 1 A.T. SAMPLE.
ASSAY RECOMMENDED FOR ROCK AND CORE SAMPLES IF CU PB ZN AS > 1%, AG > 30 PPM & AU > 1000 PPB
- SAMPLE TYPE: ROCK PULP

DATE RECEIVED: OCT 4 2002 DATE REPORT MAILED: Oct 8/02 SIGNED BY. C. L. D. TOYE, C.LEONG, J. WANG; CERTIFIED B.C. ASSAYERS

SAMPLE#	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Ni ppm	Co ppm	Mn ppm	Fe %	As ppm	U ppm	Au ppm	Th ppm	Sr ppm	Cd ppm	Sb ppm	Bi ppm	V ppm	Ca %	P %	La ppm	Cr ppm	Mg %	Ba ppm	Ti %	B ppm	Al %	Na %	K %	W ppm
E 182934	1	6	13	205	<.3	73	28	1259	5.90	12	<8	<2	2	33	1.0	4	<3	81	.50	.155	9	22	.14	202	<.01	7	.83	.01	.27	<2
E 182935	1	30	8	142	.4	54	19	733	3.72	5	<8	<2	2	17	<.5	<3	<3	39	.29	.087	6	14	.11	120	<.01	5	.59	.01	.23	<2
E 182936	1	39	48	288	.7	55	17	672	3.70	12	<8	<2	2	15	1.0	5	<3	37	.27	.091	6	15	.09	111	<.01	5	.54	.01	.22	<2
E 182937	1	26	448	419	1.6	64	24	616	3.60	42	<8	<2	3	20	1.9	5	<3	51	.48	.080	7	17	.21	217	<.01	6	.64	.01	.24	<2
E 182938	4	101	1649	896	5.4	97	29	2430	6.68	14	<8	<2	2	26	6.3	27	<3	75	.63	.136	9	24	.15	441	<.01	6	.60	.01	.21	<2
E 182939	2	93	808	707	4.9	96	34	1450	5.26	24	<8	<2	3	25	3.6	24	<3	82	.58	.155	12	30	.20	245	<.01	8	1.06	.02	.28	8
E 182951	1	21	12	241	.3	52	24	1499	3.97	5	<8	<2	2	48	1.2	3	<3	80	5.30	.130	10	26	.65	609	<.01	7	.86	.01	.28	<2
E 182952	<1	37	32	357	1.6	57	23	1587	4.16	25	<8	<2	<2	35	1.6	9	<3	72	3.55	.125	11	34	.57	379	<.01	8	.70	.01	.27	<2
E 182953	<1	37	22	165	.3	58	27	1164	4.22	6	<8	<2	2	70	.9	<3	<3	56	4.02	.119	9	35	1.02	484	<.01	5	.58	.01	.30	<2
E 182954	1	59	612	486	4.3	58	18	1500	3.79	13	<8	<2	2	20	3.0	13	<3	59	1.44	.112	7	28	.21	133	<.01	6	.58	<.01	.27	<2
E 182955	2	95	948	866	12.8	90	36	1885	4.96	63	<8	<2	2	41	6.2	14	3	75	.92	.114	8	47	.34	502	<.01	5	.83	.01	.21	<2
E 182956	6	204	994	670	28.4	107	44	1337	4.51	188	<8	<2	2	23	6.7	71	<3	47	.33	.063	3	23	.40	268	.01	6	.72	<.01	.22	<2
E 182957	5	49	498	592	7.1	73	25	1057	3.83	102	<8	<2	3	19	4.5	16	<3	37	.24	.052	2	16	.11	162	<.01	4	.49	.01	.21	<2
E 182958	2	54	622	506	7.9	68	21	994	4.13	93	<8	<2	2	21	3.3	15	<3	37	.26	.049	2	15	.13	176	<.01	4	.55	.01	.22	<2
E 182959	2	51	209	396	6.5	48	17	552	3.63	98	<8	<2	<2	20	1.8	13	<3	34	.26	.055	2	12	.13	114	<.01	4	.59	<.01	.21	<2
E 182960	17	84	65	365	.7	77	35	683	4.12	187	<8	<2	2	20	.9	10	<3	44	.30	.095	6	21	.10	91	<.01	4	.61	<.01	.27	<2
RE E 182960	17	83	64	360	.7	76	35	669	4.06	184	<8	<2	<2	20	.8	9	<3	43	.29	.094	6	21	.10	87	<.01	4	.59	<.01	.26	<2
E 182961	2	49	62	176	1.2	74	26	764	4.40	112	<8	<2	<2	25	.6	8	<3	41	.28	.047	1	18	.10	169	<.01	4	.48	.01	.20	<2
E 182962	2	31	225	284	.8	44	18	873	3.93	59	<8	<2	2	35	1.8	6	<3	49	1.13	.059	3	20	.34	150	<.01	5	.73	.02	.21	<2
E 182963	1	35	140	356	2.0	65	22	1011	3.98	91	<8	<2	<2	28	2.2	7	<3	40	.58	.045	1	17	.28	179	<.01	3	.43	.01	.23	<2
E 182964	3	64	82	311	2.6	80	27	852	4.95	121	<8	<2	<2	29	1.5	8	<3	47	.26	.037	1	21	.10	415	<.01	3	.42	.01	.22	<2
E 182965	1	40	137	191	2.0	65	24	758	4.29	95	<8	<2	<2	28	.6	6	<3	51	.35	.037	2	22	.18	159	<.01	4	.65	.02	.23	<2
E 182966	1	15	19	94	<.3	40	26	1559	4.28	8	<8	<2	2	33	.5	<3	<3	98	.52	.152	14	44	.18	209	.02	5	1.01	.03	.20	<2
E 182967	1	12	14	90	<.3	53	24	1493	4.17	7	<8	<2	2	40	<.5	<3	<3	95	1.62	.141	13	47	.47	172	.02	5	.62	.03	.24	<2
E 182968	<1	11	20	197	<.3	53	28	2518	4.91	2	<8	<2	2	119	1.3	3	<3	84	4.00	.133	9	42	1.48	565	.03	8	.48	.03	.17	<2
E 182969	1	65	14	77	<.3	51	25	1117	4.40	11	<8	<2	2	43	<.5	<3	<3	81	.75	.164	10	35	.19	179	.01	5	.65	.02	.23	<2
E 182970	<1	22	17	73	<.3	40	20	1635	4.43	6	<8	<2	<2	36	.5	<3	<3	67	.52	.123	10	53	.20	179	.06	5	.70	.04	.16	<2
E 182971	<1	7	27	112	<.3	45	22	2171	3.91	4	<8	<2	<2	227	.8	<3	<3	61	4.96	.098	9	37	2.38	110	.04	4	.45	.04	.15	<2
E 182972	<1	12	16	76	<.3	43	23	1067	5.12	8	<8	<2	<2	38	<.5	<3	<3	99	.46	.144	10	47	.19	148	.04	4	.82	.04	.15	<2
E 182973	<1	5	9	71	<.3	51	22	1138	3.84	7	<8	<2	2	86	.5	<3	<3	81	2.48	.118	8	43	1.06	183	.02	5	.52	.04	.18	<2
STANDARD.DS4	7	122	31	160	<.3	33	12	779	3.21	22	<8	<2	4	28	5.4	5	5	76	.52	.096	16	166	.60	144	.08	<3	1.69	.04	.16	3

Sample-type: ROCK R150 60C. Samples beginning 'RE' are Reruns and 'RRE' are Reject Reruns.

Data ✓ FA ___

ACME ANALYTICAL LABORATORIES LTD. 852 E. HASTINGS ST. VANCOUVER BC V6A 1R6 PHONE(604)253-3158 FAX(604)253-1716
(ISO 9002 Accredited Co.)

GEOCHEMICAL ANALYSIS CERTIFICATE

Tenaion Resources Corp. File # A205338

660 - 625 Howe St., Vancouver BC V6C 2T6

SAMPLE#	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Ni ppm	Co ppm	Mn ppm	Fe %	As ppm	U ppm	Au ppm	Th ppm	Sr ppm	Cd ppm	Sb ppm	Bi ppm	V ppm	Ca %	P %	La ppm	Cr ppm	Mg %	Ba ppm	Ti %	B ppm	Al %	Na %	K %	W ppm
G-1	1	3	3	40	<.3	5	4	491	1.89	<2	<8	<2	5	81	<.5	<3	<3	39	.51	.084	6	15	.53	235	.11	<3	.96	.09	.47	<2
E 182974	1	18	7	103	<.3	28	11	961	2.75	5	<8	<2	<2	23	.6	<3	<3	60	.44	.110	7	25	.69	253	.06	<3	1.67	<.01	.13	<2
E 182975	<1	8	5	62	<.3	13	6	886	1.79	<2	<8	<2	<2	24	<.5	<3	<3	40	.34	.039	7	16	.28	169	.04	<3	1.06	.01	.03	<2
E 182976	1	33	11	111	.3	37	11	1622	3.09	2	<8	<2	2	57	<.5	<3	<3	57	.72	.098	32	32	.55	415	.01	<3	2.45	.02	.09	<2
E 182977	<1	8	7	69	<.3	17	6	501	2.03	<2	<8	<2	<2	20	<.5	<3	<3	47	.21	.034	6	17	.31	162	.06	<3	1.08	.01	.03	<2
E 182978	1	6	9	64	<.3	12	7	308	2.26	2	<8	<2	2	14	<.5	<3	<3	48	.16	.156	4	17	.24	158	.04	<3	1.12	.01	.04	<2
E 182979	<1	8	4	52	<.3	12	6	342	1.93	3	<8	<2	<2	20	<.5	<3	<3	45	.23	.033	7	17	.31	119	.05	<3	1.00	.01	.03	<2
E 182980	1	9	6	51	<.3	11	7	237	2.21	4	<8	<2	<2	18	<.5	<3	<3	51	.26	.076	4	17	.27	07	.02	<3	1.05	.01	.03	<2
E 182981	<1	9	5	57	.4	17	7	542	2.06	<2	<8	<2	<2	24	<.5	<3	<3	44	.38	.066	3	18	.30	110	.02	<3	1.07	.01	.05	<2
RE E 182981	1	9	3	59	<.3	18	7	564	2.12	3	<8	<2	<2	27	<.5	<3	<3	45	.41	.070	4	19	.31	117	.02	<3	1.11	.01	.05	<2
E 182982	<1	7	4	49	<.3	10	6	640	2.00	2	<8	<2	<2	24	<.5	<3	<3	46	.30	.075	4	17	.24	135	.04	<3	.90	.01	.07	<2
E 182983	<1	6	3	49	<.3	13	6	531	1.97	2	<8	<2	<2	23	<.5	<3	<3	44	.25	.070	5	18	.27	115	.04	<3	.94	.01	.06	<2
STANDARD DS4	7	120	31	149	<.3	34	12	756	3.08	23	<8	<2	4	26	5.4	5	5	72	.50	.092	16	158	.58	138	.07	<3	1.63	.04	.14	4

GROUP 1D - 0.50 GM SAMPLE LEACHED WITH 3 ML 2-2-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR, DILUTED TO 10 ML, ANALYSED BY ICP-ES.
UPPER LIMITS - AG, AU, HG, W = 100 PPM; MO, CO, CD, SB, BI, TH, U & B = 2,000 PPM; CU, PB, ZN, NI, MN, AS, V, LA, CR = 10,000 PPM.
- SAMPLE TYPE: SOIL SS80 60C Samples beginning 'RE' are Reruns and 'RRE' are Reject Reruns.

DATE RECEIVED: DEC 5 2002 DATE REPORT MAILED: Dec 13/02 SIGNED BY. D. Toye ..D. TOYE, C.LEONG, J. WANG; CERTIFIED B.C. ASSAYERS

Data FA