

Article

Multivariate Analysis Based on Geochemical, Isotopic, and Mineralogical Compositions of Uranium-Rich Samples

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Abstract: The chemical and isotopic (U, Pb, Sr) signatures for a suite ($n = 23$) of pristine (>80 wt. % UO_2) and altered uraninite samples (>70 – 80 wt. % UO_2) from various locations worldwide have been determined for the purpose of identifying potential fingerprints for nuclear forensic analysis. The characterization of the uraninite samples included determination of major, minor and trace element contents, Sr, Pb, and U isotopic compositions, and secondary mineral assemblages. Due to the multivariate approach adopted in this study, principal component analysis (PCA) has been employed to allow the direct comparison of multiple variable types. The PCA results indicate that the geological origin (sandstone, metamorphite, intrusive, granite and unconformity) of pristine uraninite can be readily identified utilizing various combinations of major and/or trace element concentrations with isotopic compositions.

Keywords: uraninite; principal component analysis; nuclear forensics; uraninite provenance

1. Introduction

Over the past few decades, there has been increased interest in developing novel methods/approaches for the forensic analysis of nuclear materials. This has been primarily motivated by the need to have fast and efficient methods of identifying the origin of any intercepted illicit nuclear materials (e.g., stolen or illegally mined material, uranium ore concentrates, fuel pellets, etc.). Since 1993, the International Atomic Energy Agency (IAEA) has reported 3497 incidents of nuclear and/or radioactive materials outside of regulatory control with 285 confirmed as “acts of trafficking or malicious use” [1]. The predominant source of nuclear fuel worldwide is uranium ore deposits, and therefore, these have been the subject of a growing number of mineralogical, chemical, and isotopic investigations [2–12]. These previous studies have determined and established various geochemical signatures as fingerprints for uranium ore deposits, which can then be used as a means of deciphering the origins of any unknown intercepted nuclear materials.

Uranium ore deposits are found globally with >800 locations identified [13]. These ore deposits are found in a wide variety of geological settings and span Earth’s history from the Archean/early Proterozoic to recent times [5,12,13]. For the purpose of this study, we have adopted the IAEA [13] classification, which is based on the geological setting of the uranium deposit and takes into account either the host rock (e.g., intrusive type) or geological structure (e.g., Paleozoic unconformity). Fifteen distinct deposit types are described with each deposit type containing numerous subtypes [13].

As uranium ore deposits form under a large range of conditions including temperature and source (elemental availability), variations in their major and trace element compositions and isotopic signatures are expected. This has been the main premise of many of the previous studies listed above. Combinations of these various signatures can indicate the deposit type of the unknown nuclear material, which in turn will limit the possibilities of geographic origin. For example, an elevated temperature of formation ($>350\text{ }^{\circ}\text{C}$) allows for cations other than U^{4+} to substitute more readily into the crystal structure of uraninite, $\text{U}^{4+}_{1-x-y-z-u}\text{U}^{6+}_x\text{REE}^{3+}_y\text{M}^{2+}_z\text{□}^{4-}_u\text{O}_{2+x-0.5y-z-2u}$ the primary mineral of uranium ore (REE = rare earth elements, M^{2+} represents divalent metals and □ indicates a vacant site; [14–16]). Thus, uraninite formed at higher temperatures typically contains higher abundances of Th^{4+} , REEs^{3+} , and Y^{3+} [2,7,11,12,17]. To date, chondrite-normalized REE (CN-REE) patterns have proven to be the most effective forensic tool in terms of distinguishing between uraninite samples formed under different geologic conditions (e.g., [7,11,12]). Moreover, the concentration of Pb is linked to the age of the uraninite deposit as a result of the decay of ^{235}U and ^{238}U over geological time to ^{207}Pb and ^{206}Pb , respectively [15]. In addition to the elements that are preferentially incorporated due to geological setting or time of formation, the geochemical signature of uraninite samples will change due to the degree of alteration experienced by the uraninite. For example, due to auto-oxidation and decay damage, various elements will be lost and substituted with other elements depending on the degree of availability of the elements in the alteration fluid (e.g., Ca, Fe, and Si; [18]).

In this study, the geochemical and isotopic (U, Pb, Sr) compositions of a global suite of uraninite from various deposit types have been characterized in detail. Individual uraninite samples consist of different regions/segments that have undergone various degrees of alteration. Therefore, we assign the term *pristine* to uraninites that contain $>80\text{ wt. \% UO}_2$, and *altered* to samples that contain 70–80 wt. % UO_2 . All regions of the uraninite that contain $<70\text{ wt. \%}$ are considered to represent secondary uranium minerals and are beyond the scope of this study. Initially, the major and trace element contents that have been used previously to identify various deposit types are examined. However, to date, no single geochemical signature has been able to successfully distinguish between all the different ore deposit types. Therefore, an additional multivariate statistical approach, namely principal component analysis (PCA), was used based on the geochemical data to help discern further information, and evaluate its application as a tool for nuclear forensic analysis. PCA is a commonly-used statistical procedure for multivariate datasets that allows for the reduction of dimensionality, but maintains the most information possible, e.g., removal of variables to reveal any systematic hidden trends/patterns within the dataset [19,20]. To our knowledge, PCA has been adopted previously for forensic analysis of nuclear materials [9,10], however, this study is the first to combine chemical signatures with isotope and mineralogical data.

2. Materials and Methods

2.1. Deposit Type and Sample Descriptions

A total of 23 uraninite samples (Table 1) representing five of the IAEA uranium deposit types are investigated here: intrusive non-granite-related (type 1), intrusive granite-related (type 2), metamorphite (type 6), Proterozoic unconformity (type 7), and sandstone (type 9). A brief description of each uranium deposit type is provided below in addition to the geological setting for each uraninite investigated here.

Intrusive non-granite-related deposits are classified according to the host rock of the deposit, which can include a wide range of whole rock compositions, e.g., pegmatite to carbonatite. Intrusive non-granite-related deposit types can be related to either anatectic or plutonic activity [13]. Six intrusive type samples were investigated in this study (Table 1): Yancey (#333), Mitchell 1 (#334), Mitchell 2 (#338), Bancroft (#340), Ruggles (#344), and Foster Lake (#348). Yancey, Mitchell 1, and Mitchell 2 uraninite (North Carolina, USA) are hosted in Paleozoic pegmatite from the Spruce Pine pegmatite, which crosscuts Precambrian interlayered mica and amphibole gneiss and schist [21]. The Bancroft

sample (Bancroft region, Ontario, Canada) is hosted within the Central Metasedimentary Belt of the Proterozoic Grenville Province, which was intruded by a number of plutonic suites. The uranium mineralization is associated with fenite-carbonatite and granite pegmatite post-tectonic plutonic events [22]. The Ruggles sample (New Hampshire, USA) is a pegmatite hosted uraninite found in the Devonian-aged Littleton Formation, which is related to the closing of the Proto-Atlantic Ocean [2,23]. The Foster Lake uraninite (Saskatchewan, Canada) is hosted within narrow belts of biotite schist along the margins of granitic pegmatites at fault intersections of the Wollaston Group and Archean basement contacts [24–26].

Table 1. Location and deposit classification of uraninite samples (n = 23).

Sample	Mine Location	State	Country	IAEA Classification
333	Yancey County	North Carolina	USA	Intrusive non-granite-related
334	Mitchell County	North Carolina	USA	Intrusive non-granite-related
338	Mitchell County	North Carolina	USA	Intrusive non-granite-related
340	Bancroft	Ontario	Canada	Intrusive non-granite-related
344	Ruggles Mine	New Hampshire	USA	Intrusive non-granite-related
348	Foster Lake	Saskatchewan	Canada	Intrusive non-granite-related
353	Happy Jack, San Juan County	Utah	USA	Sandstone tabular
423	Great Bear Lake	Northern Territories	Canada	Metamorphite
437	Shinkolobwe	Katanga	D.R. Congo	Metamorphite
511	Rabbit Lake	Saskatchewan	Canada	Proterozoic unconformity
516	Australia	Northern Territories	Australia	Metamorphite
522	Billiken Lode, Jefferson County	Colorado	USA	Metamorphite
531	Marshall Pass, Gunnison County	Colorado	USA	Metamorphite hydrothermal vein
564	Jachymov	Karlovy Vary	Czech Republic	Intrusive granite-related
604	Australia	Northern Territories	Australia	Proterozoic unconformity
623	Marshall Pass	Colorado	USA	Metamorphite hydrothermal vein
626	Echo Mine, Great Bear Lake	Northern Territories	Canada	Metamorphite
662	Shinkolobwe	Katanga	D.R. Congo	Metamorphite
809	Marshall Pass	Colorado	USA	Metamorphite hydrothermal vein
1232	Big Indian Wash	Utah	USA	Sandstone
1237	Adair Mine, Cane Spring Canyon	Utah	USA	Sandstone
1262	Adair Mine, Cane Spring Canyon	Utah	USA	Sandstone
1303	Big Indian Wash	Utah	USA	Sandstone

Granite-related uranium deposits are associated with granite intrusions and may consist of uranium ore veins, either endogranitic or perigranitic, or disseminated mineralization within granitic bodies. The Jachymov (#564) uraninite sample is an intrusive granite-related uranium ore from the Czech Republic. It is associated with the Erzgebirge granites that formed during the collisional stage of the Variscan Orogeny along the NW trending Gera-Jachymov Fault Zone present within the central Bohemian Massif [13]. Metamorphite uranium deposits form deposit types similar to granite-related deposits, e.g., veins and disseminated mineralization; however, metamorphite deposits are associated with regional metamorphism of uraniferous sediments rather than a granite intrusion. Six of the samples investigated here have been grouped as metamorphite uranium deposits: Great Bear 1 (#423), Great Bear 2 (#626), Shinkolobwe 1 (#437), Shinkolobwe 2 (#662), Australia 1 (#516), and Billiken (#522). The Great Bear samples (Great Bear Lake region, Northwest Territories, Canada) are found within quartz and carbonate gangue in pumice-dominated pyroclastic flows of late Aphebian age. The remobilization of uranium has been linked to diabase intrusions in the Echo Bay region [27–29]. The Shinkolobwe uraninite (Democratic Republic of Congo) is located within fractures of siliceous dolomite and dolomitic shale of the lower Roan Group of the Neoproterozoic Katanga Supergroup that unconformably overlies basement [7]. The Billiken uraninite (Colorado, USA) consists of uranium ore veins hosted in Neoproterozoic metasediments [2,30]. Three samples from Marshall Pass (#531, 623, 809), Colorado, USA, have been further classified as metamorphite hydrothermal veins [23]. The uranium deposits formed due to hydrothermal activity and are constrained within fault-controlled veins and brecciated zones of Pennsylvanian-aged limestone [31].

Proterozoic unconformity uranium deposits are defined by geological structure rather than host rock type; e.g., uranium deposits occur either above or below an unconformity that separates the crystalline basement from overlying Proterozoic sediments. Two uraninite samples from Proterozoic

unconformities are investigated in this study: Rabbit Lake (#511) and Australia 2 (#604). The Rabbit Lake sample is from northern Saskatchewan, Canada. The orebody is located within the Wollaston domain, 10–100 m below the Helikian sub-Athabasca Formation unconformity [32]. The Australian sample is from Northern Territories, Australia. A mine location is not known for this sample, however, its CN-REE signature indicates that it is an unconformity-related deposit (Figure S1; [11,12]).

The last type of uranium deposit discussed in this study is sandstone. Sandstone uranium deposit types form due to the reduction of U^{6+} in fluids within sandstone resulting in U^{4+} precipitation. Sandstone uranium deposits are hosted in fluvial and marginal marine sedimentary environments [13]. Four samples are classified as sandstone type: Big Indian Wash 1 (#1232), Big Indian Wash 2 (#1303), Adair Mine 1 (#1237), and Adair Mine 2 (#1262). An additional sample, Happy Jack (#353) is identified as sandstone tabular deposit [2,12]. These five samples are from San Juan County, Utah; Adair Mine, Big Indian Wash, and Happy Jack. Adair Mine is uraninite from Cane Spring Canyon. It is hosted within the Moss Back member of the Triassic Chinle formation. It consists of calcite-cemented sandstone that grades into calcareous siltstone interbedded with limestone, shale, and minor sandy layers. The uraninite is primarily found as a replacement of woody material or as cement [33]. Big Indian Wash is hosted in the Cutler formation located below the Permian-Triassic unconformity (below the Moss Back member). The Cutler formation consists of alternating lenses of mudstone, calcareous siltstone, and arkosic sandstone [34]. The Happy Jack sample from the White Canyon area is classified as a tabular sandstone due to the geometry of the deposit. The Happy Jack U ore deposit is hosted in the same Permian to Triassic formations as Adair Mine and Big Indian Wash [35].

2.2. Analytical Methods

For each uraninite investigated here (Table 1), an aliquot was prepared for both in-situ and solution mode analyses. For in-situ analysis, a $\sim 1\text{ cm}^2$ cut portion of each uraninite was mounted in epoxy, cured, and then polished. The portion of sample selected, where possible, included the transition from pristine to altered uraninite, as determined by color and hardness of the sample (Figure 1). The spatial distribution and qualitative assessment of the major element abundances, U, Pb, Si, Ca, Th, and Fe (Figure S2), were determined by micro-X-ray Fluorescence Spectroscopy (μ XRF). An EDAX Orbis PC μ XRF was used to collect the elemental maps with an X-ray aperture of 30 μm , 40 kV voltage, 300 μA current, and a 100 μs dwell time over a 512×400 matrix. Appropriate transects through individual uraninite samples that encompass transition zones between pristine and altered sections were conducted for further investigation (e.g., Figure 1). A suite of major element oxide abundances (wt. %; UO_2 , PbO, ThO_2 , SiO_2 , FeO, Al_2O_3 , MnO, TiO_2 , As_2O_3 , V_2O_3 , CaO, ZrO_2 , K_2O , P_2O_5 , SO_3 , and Y_2O_3) were determined along the transects identified by μ XRF utilizing a Cameca SX-50 electron microprobe (EMP; Table S1). Standard EMP operating conditions included an accelerating voltage of 15 kV and beam size of ca. 3 μm with a beam current of 100 nA. A suite of well characterized standards were used to calibrate the EMP prior to each analytical session and included; synthetic uranium dioxide (UO_2), pyrite (PbS), synthetic thorium dioxide (ThO_2), synthetic zircon (ZrSiO_4), anorthite glass ($\text{CaAl}_2\text{Si}_2\text{O}_8$), titanium dioxide (TiO_2), Mn metal, synthetic Y-Al garnet ($\text{Y}_3\text{Al}_5\text{O}_{12}$), synthetic calcium phosphide (Ca_2P), synthetic vanadium oxide (V_2O_3), K-feldspar, Mn-bearing olivine, and asbestos. The trace element concentrations were determined using a New Wave Research UP213 Nd:YAG laser ablation (LA) system coupled to a Thermo Finnegan Element2 sector field high resolution (HR) inductively coupled plasma mass spectrometer (ICP-MS; Table S2). Analyses were conducted using a repetition rate of 5 Hz on a 30 μm diameter spot corresponding to a fluence of 12 Jcm^{-2} . Acquisition parameters consisted of 1 pass with 160 cycles of 8 ms dwell time for each mass of interest [36–39]. Background signals were collected for 60 s with the shutter closed prior to sample ablation of 60 s duration. The typical pit depth for a 60 s ablation was $<75\text{ }\mu\text{m}$ with clean edges and flat base [23,38]. A standard-bracketing technique was employed with NIST SRM 610 glass wafer as the external standard [40]. Where possible, PbO wt. % as determined by EMP was used as the internal standard [23]; if the abundance of PbO was considered too low ($<0.5\text{ wt. \%}$), then the CaO

wt. % content was used as the internal standard for data reduction purposes within the GLITTER[®] software [23,41].

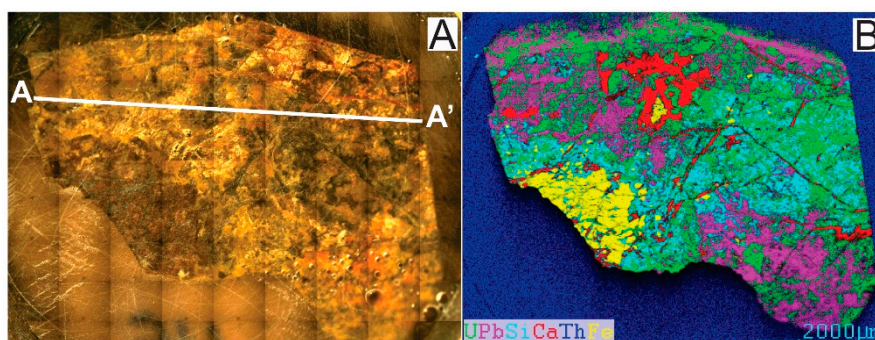


Figure 1. Sample image (A) and micro-XRF major element map (B) for Rabbit Lake uraninite (#511) illustrating typical compositional changes observed in samples investigated here (Tables S1 and S2). Line A to A' demonstrates a typical transect that was chosen to investigate both pristine and altered sections of each sample.

For solution and powder X-ray diffraction (PXRD) analyses, the uraninite samples were separated into pristine and altered aliquots (Figure 1). If a clear distinction could not be made between the pristine and altered portions, then a third aliquot identified as bulk was prepared. Approximately, 0.5–1.0 g of each sample aliquot (pristine, altered, $\pm$ bulk) was powdered using an agate pestle and mortar with ca. 0.05 g of sample being digested using a two-acid method (HNO_3 and HF; [42]). An additional 0.1–0.2 g of the powdered sample was used for PXRD analysis for the purpose of phase identification within the altered and bulk samples only (Table S3). PXRD data were collected using a Bruker D8 Advance Davinci powder diffractometer in Bragg–Brentano configuration. Cu-K α radiation was produced with an accelerating voltage of 40 kV and 40 mA current. An incident-beam slit of 1.0 mm was reduced by a 0.6 mm slit in combination with 0.02 mm absorber and diffraction (0.6 mm) slits. Data were collected using a step scan with a step velocity of $0.8^\circ \text{ min}^{-1}$ in the range of $5\text{--}55^\circ 2\theta$ using a LynxEye solid-state detector.

Trace element concentrations of all solution aliquots were collected on a Nu Instruments AttoM High Resolution (HR) inductively coupled plasma mass spectrometer (ICP-MS). Analyses were conducted using medium mass resolution ($M/M\Delta \approx 2500$) employing a spike addition method in wet plasma mode modified after Jenner et al. [43]. Acquisition consisted of 3 cycles of 300 sweeps with equal dwell times of 1 ms for each isotope of interest. In addition to trace element concentrations, Pb, U, and Sr isotopic compositions were also investigated in this study (Table S4). All isotopic measurements were conducted on a Nu Plasma II Multi-Collector ICP-MS with a desolvating nebulizing system (DSN-100 from Nu Instruments). With regard to Pb isotope measurements, lead separation was accomplished by a two-step anion-exchange chromatographic procedure modified after Manhès et al. [44]. Following lead separation, the samples were evaporated to dryness and re-dissolved in 2% HNO_3 solution spiked with NIST SRM 997 Thallium standard (5 ppb). Simultaneous measurements were acquired for all Pb and Tl isotopes as the Tl isotope ratio, $^{205}\text{Tl}/^{203}\text{Tl}$, was used to correct for instrumental mass bias [45]. External reproducibility was monitored by repeated measurements of a 25 ppb NIST SRM 981 (Pb) spiked with NIST SRM 997 (Tl) and yields average values (2σ) of $^{206}\text{Pb}/^{204}\text{Pb} = 16.937 \pm 0.006$, $^{207}\text{Pb}/^{204}\text{Pb} = 15.490 \pm 0.005$, $^{208}\text{Pb}/^{204}\text{Pb} = 36.697 \pm 0.015$, and $^{207}\text{Pb}/^{206}\text{Pb} = 0.91462 \pm 0.00004$ ($n = 6$). For U isotope analyses, UTEVA ion exchange columns were utilized to successfully separate uranium from any potential interferences [46]. A standard-bracketing technique was employed with New Brunswick Laboratory CRM 112-A as the external standard, which was used to correct for instrumental mass bias. The external reproducibility (2σ) associated with the $^{238}\text{U}/^{235}\text{U}$, $^{234}\text{U}/^{238}\text{U}$, and $^{235}\text{U}/^{238}\text{U}$ ratios are 0.38‰, 4.32‰, 0.38‰, respectively. Strontium isotope analyses involved separation of Sr from the sample matrix by the cation-exchange chromatographic procedure modified after Crock

et al. [47]. Due to the high abundances of REEs within uraninites (e.g., [2,11,12]), all samples were processed through the cation-exchange chromatography twice in order to achieve sufficient purity for Sr isotopic analysis on the MC-ICP-MS, which aided in the removal of plasma-based isobaric interferences (i.e., doubly-charged REEs) on Sr isotopes of interest (Tables S2 and S3). However, for six uraninite samples with a Sr/total REEs ratio of <0.03 , adequate separation of Sr from REE was not achieved despite the double pass on the ion exchange columns. Five of the 6 samples excluded for Sr isotope analysis are intrusive non-granite-related deposit type and one is a Proterozoic unconformity deposit type. The external reproducibility for the Sr isotope analyses was monitored by repeated measurement of a 100 ppb NIST SRM 987 strontium standard that yielded an average value and associated 2σ standard deviation of $^{87}\text{Sr}/^{86}\text{Sr}$ of 0.71025 ± 0.00004 ($n = 9$).

The principal component analysis was performed using R statistical software [48]. All data were log-transformed to normalize the data prior to the multivariate analyses using FactoMineR and factoextra (Factor analysis and data Mining with R; [49]) with ggplots2, GGally, and ggsci to aid in graphing [50–52]. For the purposes of this study, solely principal components 1 and 2 will be discussed.

3. Results and Discussion

3.1. Major and Trace Element Compositions and PXRD Results

The in-situ major and trace element data for the uraninite samples investigated here are presented in Tables S1 and S2. Of the 900 individual in-situ analyses conducted on the 23 samples (average 40 spots per sample), 354 analyses represent pristine uraninite ($\text{UO}_2 > 80$ wt. %), 213 correspond to altered sections (UO_2 70–80 wt. %), and the remaining analyses are considered secondary minerals as they are characterized by UO_2 contents of <70 wt. %. The totals of included analyses for each type are as follows: intrusive non granite-related = 120 (84 pristine and 36 altered), intrusive granite-related = 39 (17 pristine and 22 altered), metamorphite = 206 (103 pristine and 103 altered), metamorphite hydrothermal vein = 85 (49 pristine and 36 altered), Proterozoic unconformity = 31 (30 pristine and 1 altered), sandstone = 68 (57 pristine and 11 altered), and sandstone tabular = 18 (15 pristine and 3 altered).

As previous studies have shown, major elements can be used to distinguish between different deposit types [7,14,18]. Th^{4+} has a similar ionic radius to U^{4+} and can therefore substitute into the uraninite structure depending on element availability, especially at elevated temperatures [2,7,11,17]. The Th (ppm) and Pb (ppm) contents of the uraninite samples examined in this study are shown in Figure 2A for both pristine and altered areas. The Th and Pb concentrations are distinct for a number of deposit types in the pristine areas, namely for intrusive non-granite-related, metamorphite, and sandstone (Figure 2A). Intrusive non-granite-related samples have the highest concentrations of Th (>5000 ppm), which is attributed to the higher crystallization temperature that favors Th substitution for U (>450 °C, [7]). The Proterozoic unconformity uraninite samples are also characterized by high Th concentrations. Conversely, sandstone type uraninite samples record the lowest abundances of Th (<1 ppm) and relatively low Pb contents (<500 ppm). Pristine metamorphite type samples are identified by low Pb concentrations (typically <100 ppm) and variable Th contents that range from 10 to 1000 ppm (Figure 2A). Of note, for pristine samples, Th and Pb contents may be used to clearly distinguish between metamorphite and metamorphite hydrothermal vein type samples as the latter have higher Pb concentrations (>5000 ppm). The groupings observed for the pristine areas are not apparent in the altered sections of the samples since the latter record larger variations in the concentrations of Th and Pb for both the intrusive non-granite-related and metamorphite. As with the pristine sections, the Th concentrations for the altered regions remain the highest in the intrusive non-granite type (Figure 2A). Moreover, the Pb concentrations for the altered metamorphite samples span the entire range of measured Pb abundances. With progressive alteration of uraninite, there will be preferential loss of Pb along with various other cations that substitute within the uraninite structure, such as Si, Fe, and Ca [8,18]. Figure 2B demonstrates the scatter and lack of correlation between Pb contents and

the total major cation ($\text{SiO}_2 + \text{FeO} + \text{CaO}$) abundances for altered uraninites. In summary, within the pristine areas of the uraninite the signatures shown in Figure 2B indicate that all deposit types do form clusters except for the metamorphite and metamorphite hydrothermal vein deposit types. However, the deposit types are not adequately distinguished from each other, i.e., the various types of uraninite overlap with the metamorphite deposit type spanning the entire range for both the total cation abundance and the Pb concentration. In addition, no discrete groups are identified in the altered areas when comparing the cation and PbO abundances (Figure 2B), which suggests either the presence of other cations not examined during EMP analysis, or that a combination of minor elements contribute to replacing U during alteration.

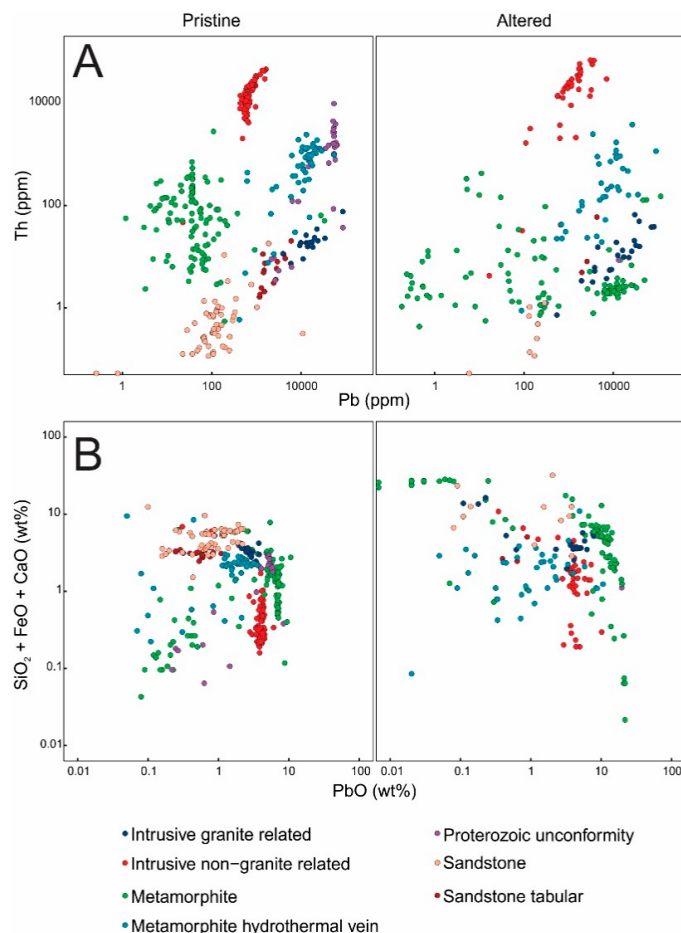


Figure 2. (A) Log–log Th (ppm) versus Pb (ppm) for both pristine ($\text{UO}_2 > 80$ wt. %) and altered (UO_2 70–80 wt. %) uraninite. (B) Log–log cation wt. % ($\text{SiO}_2 + \text{FeO} + \text{CaO}$) and PbO (wt. %) for pristine and altered areas of the uraninite samples.

To date and as stated earlier, CN-REE patterns have been demonstrated to be the most reliable forensic signature in identifying the provenance of uranium ore from different deposit types. Similar to Th^{4+} , the REE signatures of uranium deposits are primarily attributed to the source (elemental availability) and, to a lesser extent, temperature of formation [7,11]. The general consensus is that if the REEs are present at constant concentrations, then their incorporation within uraninite is positively correlated with temperature; i.e., higher temperature of formation will be associated with higher REE abundances. Conversely, if the temperature is held constant, then the availability of the REEs will dictate the degree to which they are sequestered into the uraninite structure [2,7,11,12]. Therefore, the relationship between Th^{4+} and REEs can be used to gain insight into the temperature of formation for uranium deposits (Figure 3A). As previously stated, the Th concentration is highest for the intrusive non-granite-related samples and the Proterozoic unconformity samples followed by the metamorphite

hydrothermal vein samples; therefore, these samples are characterized by the lowest U/Th ratio (U/Th < 1000, Figure 3A; [7]). The REE compositions of the pristine areas for the uraninite samples indicate a large range between all deposit types (10–100,000 ppm) and also within individual deposit types. For example, uraninite from intrusive non-granite-related samples define two distinct groups, which both plot in the magmatic temperature field ($T > 450\text{ }^{\circ}\text{C}$, U/Th < 1000); both groups show variation in the REE concentrations, which is likely due to source availability rather than temperature variability (Figure 3A). Lower temperature deposits such as sandstone, sandstone tabular, and (most of) the metamorphite samples formed below $350\text{ }^{\circ}\text{C}$, as indicated by the total REE concentrations and $450\text{ }^{\circ}\text{C}$ temperature cutoff as defined by a U/Th ratio > 1000. The altered samples exhibit the same large range in the total REE concentrations, which suggests that there is no loss of REEs as a result of alteration (Figure 3A,B and Figure S1). However, altered regions of uraninite from the different deposit types do plot in the expected temperature field, e.g., intrusive non-granite-related plot within the $450\text{ }^{\circ}\text{C}$ or above region (Figure 3A).

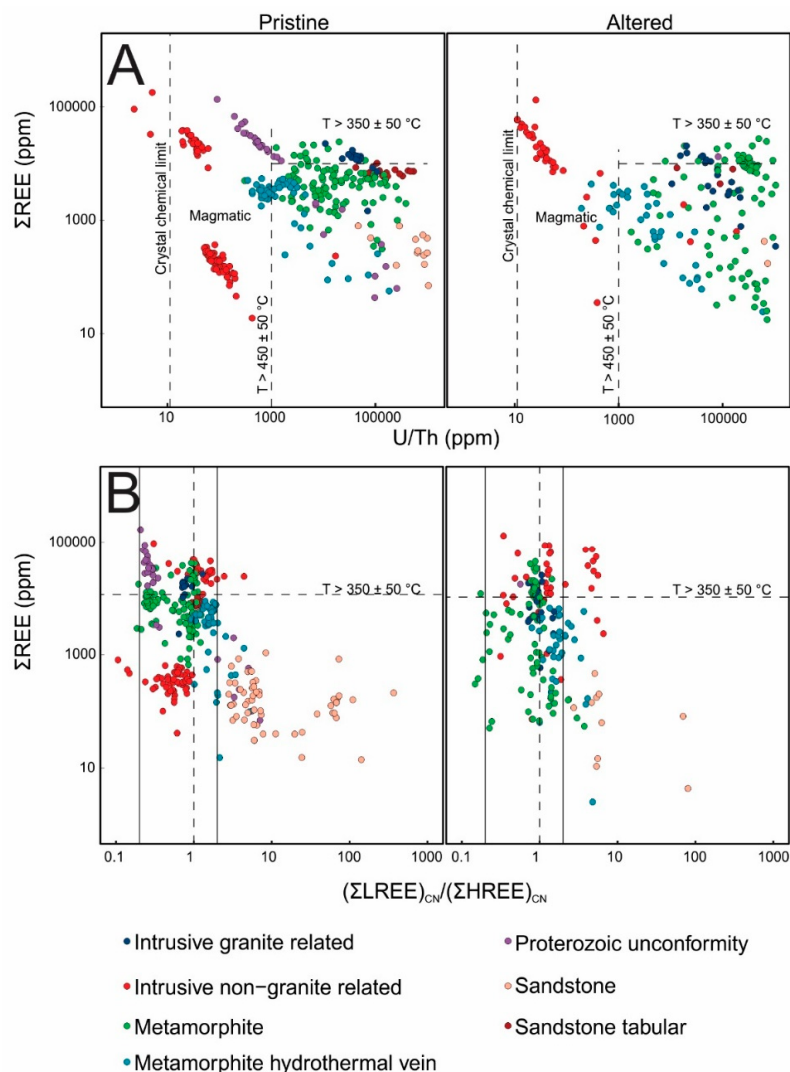


Figure 3. (A) Log–log total rare earth element concentration (ppm) versus U/Th ratio for both pristine and altered portions of the uraninite samples. The temperature threshold limits are modified from Frimmel et al. [7] and Mercadier et al. [11]. (B) Log–log of the sum of REE (ppm) and the chondrite-normalized light over heavy rare earth element fractionation (LREE = La, Ce, Pr, Nd; HREE = Er, Tm, Yb, Lu) for both pristine and altered samples. The temperature threshold limit is modified from Mercadier et al. [11].

Deposit type grouping is also evident based on the ΣREEs and Σcation compositions of pristine uraninite, e.g., sandstone, sandstone tabular, intrusive granite-related, and Proterozoic unconformity (Figure 4A). However, the metamorphite uraninite samples overlap all other deposit types except for sandstone due to the large range in cation abundances (0.1–10 wt. %). Also, no trend is observed in the altered uraninites on the basis of their ΣREE and Σcation compositions (Figure 4A). The CN-REE signature for each deposit type has a typical shape/pattern, e.g., intrusive non-granite-related deposit type is characterized by both positive cerium and negative europium anomalies, where a positive anomaly indicates an enrichment of Ce relative to its neighboring REEs and, conversely, a negative anomaly signifies a lower abundance of Eu relative to its neighboring REEs (Figure S1). As the overall CN-REE pattern remains unaffected by the degree of alteration (Figure S1), comparison of the magnitudes of both the Ce and Eu anomalies illustrates the same relative groupings for both pristine and altered regions (Figure 4B). For example, the sandstone tabular type forms a separate group relative to the sandstone type due to a positive Ce anomaly associated with the sandstone tabular, whereas a negative Ce anomaly is related to the sandstone type. CN-REE patterns for uraninite from intrusive non-granite-related type deposits define a large range in Eu anomaly values and are the most negative of all sample types investigated in this study (Figure 4B).

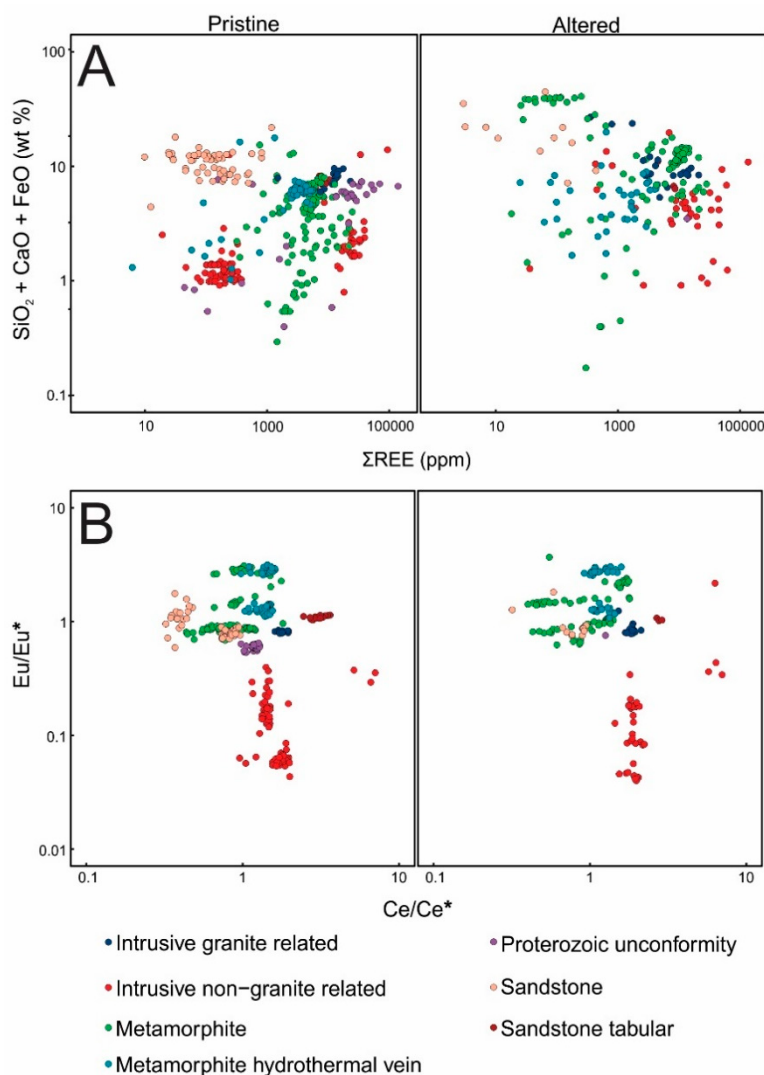


Figure 4. (A) Log–log plot of total cation (wt. %; $\text{SiO}_2 + \text{FeO} + \text{CaO}$) versus total REE (ppm) for pristine and altered uraninite samples. (B) Europium versus cerium anomaly. $\text{Eu}/\text{Eu}^* = \text{Eu}_{\text{CN}}/(\text{Sm}_{\text{CN}} + \text{Gd}_{\text{CN}})^{0.5}$ and $\text{Ce}/\text{Ce}^* = 3*\text{Ce}_{\text{CN}}/(2*\text{La}_{\text{CN}} + \text{Nd}_{\text{CN}})$ for pristine and altered uraninite samples.

In addition to the major and minor elements discussed above, previous studies have suggested that some trace elements have the potential of differentiating between various deposit types; these include Mo, Hf, Zr, and Nb [2,7,10,18,53]. Figure 5 illustrates the variation in both Hf and Nb abundances relative to Mo concentrations. The concentrations of the latter vary from <1 ppm to >10,000 ppm for all deposit types investigated here. The intrusive non-granite-related and sandstone uraninites have the lowest Mo, Hf, and Nb abundances, whereas the metamorphic hydrothermal and Proterozoic unconformity have the highest Mo, Hf, and Nb contents in pristine regions of the samples. The same holds true for the altered regions of the samples with the exception of the Proterozoic unconformity uraninite samples since none of the Proterozoic unconformity samples are considered altered (i.e., UO_2 abundances were all >80 wt. %; Figure 5).

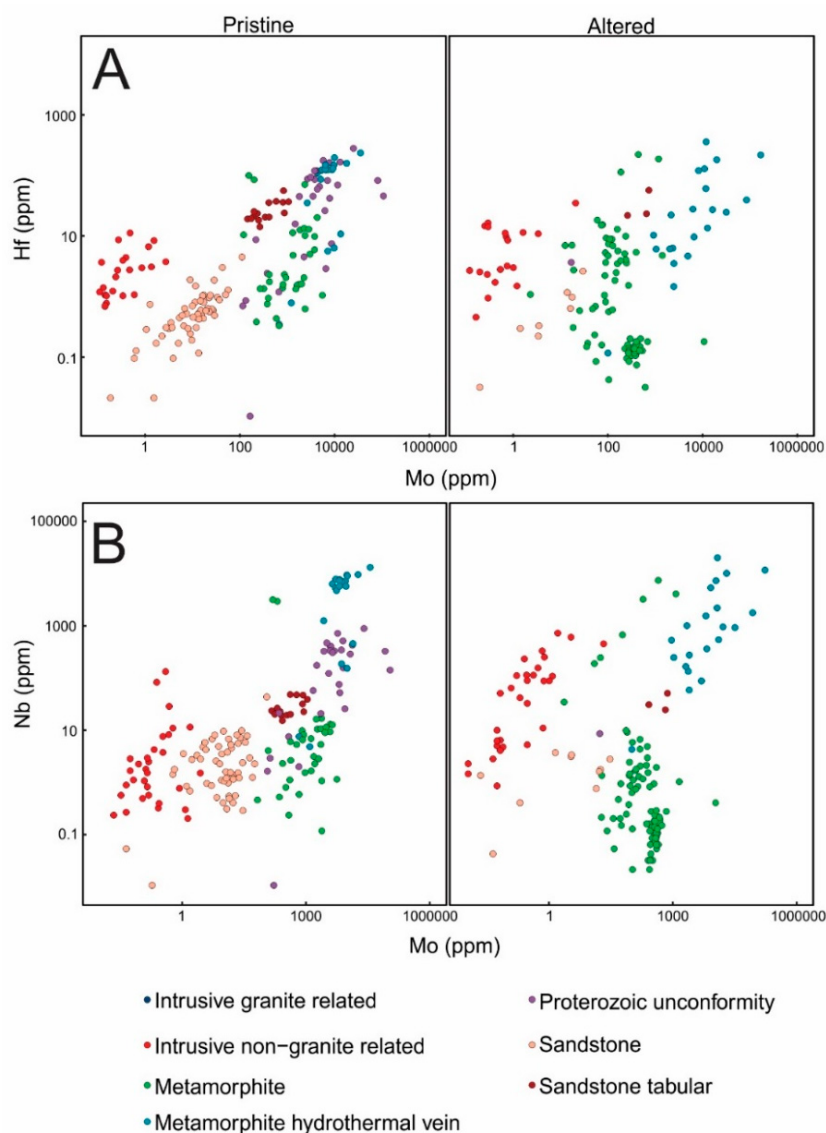


Figure 5. (A) Log–log plots of Hf (ppm) versus Mo (ppm) for pristine and altered uraninite samples and (B) Nb (ppm) versus Mo (ppm) for pristine and altered uraninite samples.

The secondary mineral phases identified by PXRD are listed in Table S3. Plasil et al. [16] summarized the sequence of secondary mineral formation expected with progressive uraninite alteration. The secondary minerals identified here encompass the range of alteration, and include mixed valence, e.g., oxidized uraninite to phosphuranylite ($\text{KCa}(\text{H}_3\text{O})_3(\text{UO}_2)_7(\text{PO}_4)_4\text{O}_4 \cdot 8\text{H}_2\text{O}$), e.g., uranocircite ($\text{Ba}(\text{UO}_2)_2(\text{PO}_4)_2 \cdot 10\text{H}_2\text{O}$), and natrouranospinite ($(\text{Na}_2,\text{Ca})(\text{UO}_2)_2(\text{AsO}_4)_2 \cdot 5\text{H}_2\text{O}$).

The secondary mineral assemblages documented here do not show any systematic variation/correlation with their associated deposit types (Table S3). Thus, this feature suggests that the geological setting of the uraninite deposit does not exert an important control on the type of secondary minerals that form; rather, it is possible that secondary mineral formation is related to the chemical composition of the alteration fluids, which reflects the regional geology rather than just the localized uranium-rich deposit. However, a larger dataset may allow for a more rigorous comparison and investigation of any significant correlation between the secondary mineral assemblage, degree of alteration, and deposit types.

3.2. Radiogenic Isotopes

The Sr, Pb, and U isotope compositions recorded for the uraninite samples investigated in this study are listed in Table S4 and shown in Figure 6. There is no clear, systematic variation between the geological setting and any of the three radiogenic isotope systems examined here. Unlike the previously discussed major and trace element concentrations, the U, Pb, and Sr isotopic data reported here were collected using bulk solution aliquots rather than in-situ methods (EMP or LA-ICP-MS), and thus this may have had the effect of averaging or eliminating discrete isotopic variations within individual samples. The total range for the $^{87}\text{Sr}/^{86}\text{Sr}_i$ values reported here is 0.70913 to 0.77885 (Table S4). The sandstone uraninite samples typically have a lower $^{87}\text{Sr}/^{86}\text{Sr}_i$ ratio (~ 0.71858), whereas the metamorphite hydrothermal vein samples all cluster around 0.72450. Both the intrusive non-granite-related and metamorphite samples define the largest range in $^{87}\text{Sr}/^{86}\text{Sr}_i$ ratios (0.71382 to 0.77885). These two uraninite types also have a large range in deposit ages (Table S4), however, there is no clear relationship between initial $^{87}\text{Sr}/^{86}\text{Sr}_i$ ratios and age of deposit (Figure S3). There also appears to be little change in the overall range in the initial $^{87}\text{Sr}/^{86}\text{Sr}_i$ ratios for both altered or pristine uraninite (Figure 6 and Figure S3).

The $^{207}\text{Pb}/^{206}\text{Pb}$ values for the samples studied here range between 0.05243 and 0.51904. The lower $^{207}\text{Pb}/^{206}\text{Pb}$ values (<0.100) are associated with the intrusive non-granite-related samples, sandstone, and sandstone tabular sample (with the exception of sample 1237; Table S4). As with their initial Sr compositions, the metamorphite uraninite samples record the largest range in $^{207}\text{Pb}/^{206}\text{Pb}$ compositions such that these cover the entire observed range. The $^{206}\text{Pb}/^{204}\text{Pb}$ and $^{207}\text{Pb}/^{204}\text{Pb}$ ratios also define a large range, 32–364450 and 16.7–27065, respectively. The $^{238}\text{U}/^{235}\text{U}$ compositions range from 137.502 to 137.934 (Figure 6). The lowest $^{238}\text{U}/^{235}\text{U}$ value is reported for a metamorphite sample, whereas the highest is for a sandstone tabular uraninite. However, based on the U isotope compositions reported here (Figure 6; Table S4), it is clear that these cannot be used as an effective tool for provenance determination within a nuclear forensic analysis.

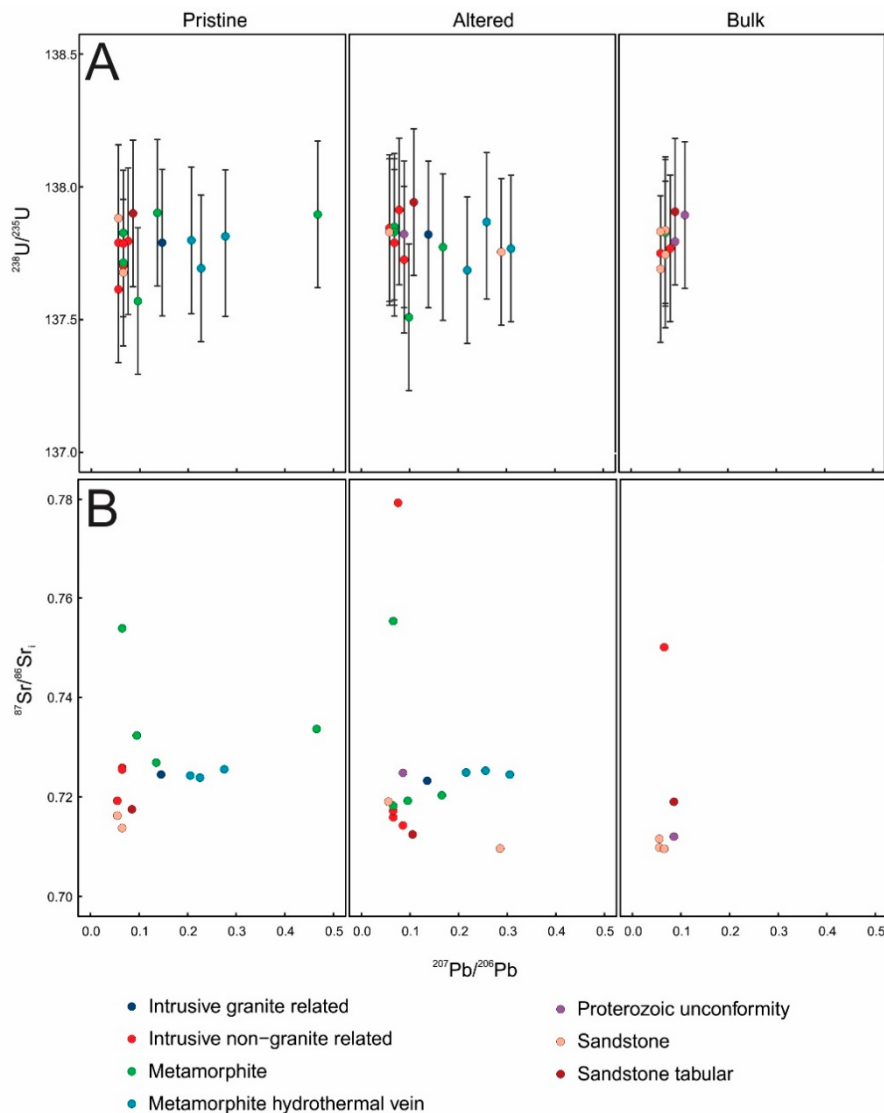


Figure 6. (A) $^{238}\text{U}/^{235}\text{U}$ ratio versus $^{207}\text{Pb}/^{206}\text{Pb}$ ratio for pristine, altered, and bulk solution. (B) $^{87}\text{Sr}/^{86}\text{Sr}$ versus $^{207}\text{Pb}/^{206}\text{Pb}$ ratio for pristine, altered, and bulk solution. The error bars represent 2σ (relative standard deviation) for the given isotope ratio. The error bars for $^{207}\text{Pb}/^{206}\text{Pb}$ and $^{87}\text{Sr}/^{86}\text{Sr}$ are within the symbol size.

3.3. Principal Component Analysis (PCA) Results

Principal component analysis (PCA) was utilized to compare the data reported here as a whole rather than looking at any one particular trace element or isotopic signature. All variables (major, minor, trace, Sr, Pb, and U isotope ratios for both pristine and altered, and secondary minerals for the altered uraninite only) were used to produce the two PCA shown in Figure 7. Using all variables, the sum of PC 1 and PC 2 explains 50.1% of the variance of the pristine areas and 38.5% of the variance within the altered samples (Figure 8 and Table S5). The PCA results shown in Figure 7 indicate that the pristine uraninite samples are readily grouped according to deposit types. Of particular note, the secondary mineral assemblage observed within the altered uraninite areas does not contribute significantly to the PCA since all secondary minerals account for <7% of the variance recorded in principal component one and <20% for principal component two (Table S5). The variables that influence each PCA in the altered samples are similar to the variables in the pristine samples, e.g., Nb, Hf, and Cs, as well as the REEs; however, the Pb isotopic composition has little to no influence (Figure 8). The metamorphite uraninite samples plots within the negative PC 2 field, which is controlled primarily by the uranium

isotopic compositions, whereas the metamorphite hydrothermal vein samples all plots within the positive PC 2 field (Nb, Hf, Cs, Ta, and Zr). Both metamorphite and metamorphite hydrothermal vein samples span the range of PC 1, which is similar to the pristine samples and is controlled primarily by the REE signatures.

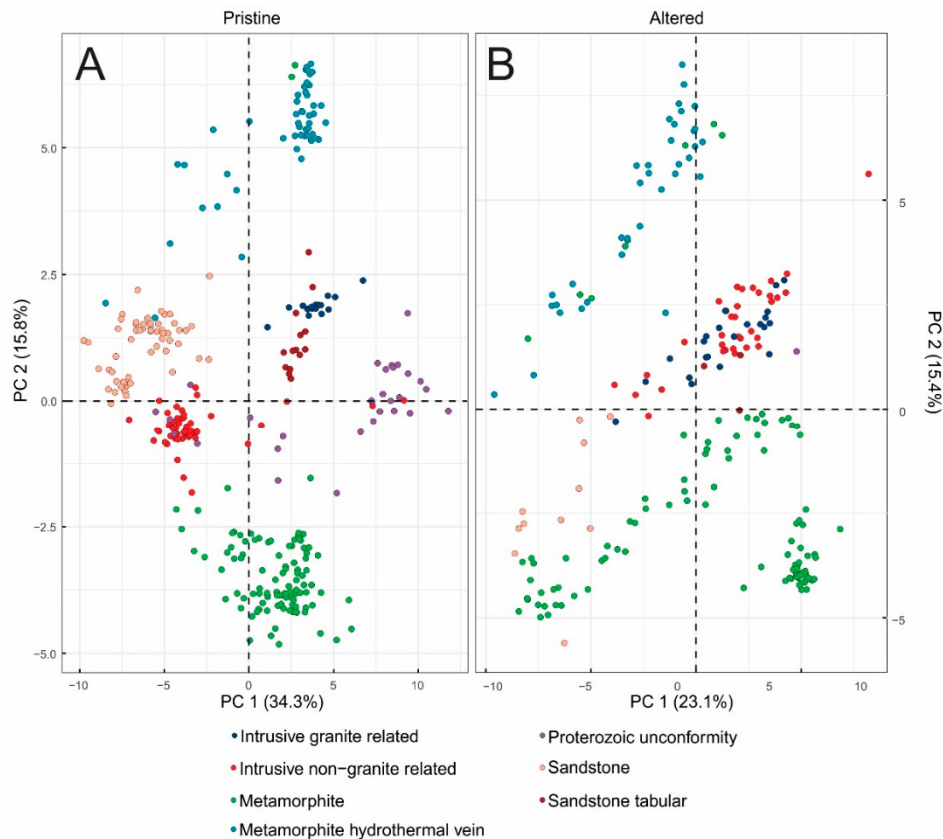


Figure 7. Principal component analysis for (A) pristine and (B) altered uraninite samples using all variables; pristine and altered both use major (wt. %; Table S1), minor and trace (ppm; Table S2) abundances and Sr, Pb, and U isotopic compositions (Table S4). The altered PCA also uses the secondary minerals that were identified with PXRD (Table S3). Two principal component analyses were conducted; therefore, the PC 1 and PC 2 are different vectors for both the pristine and altered PCA.

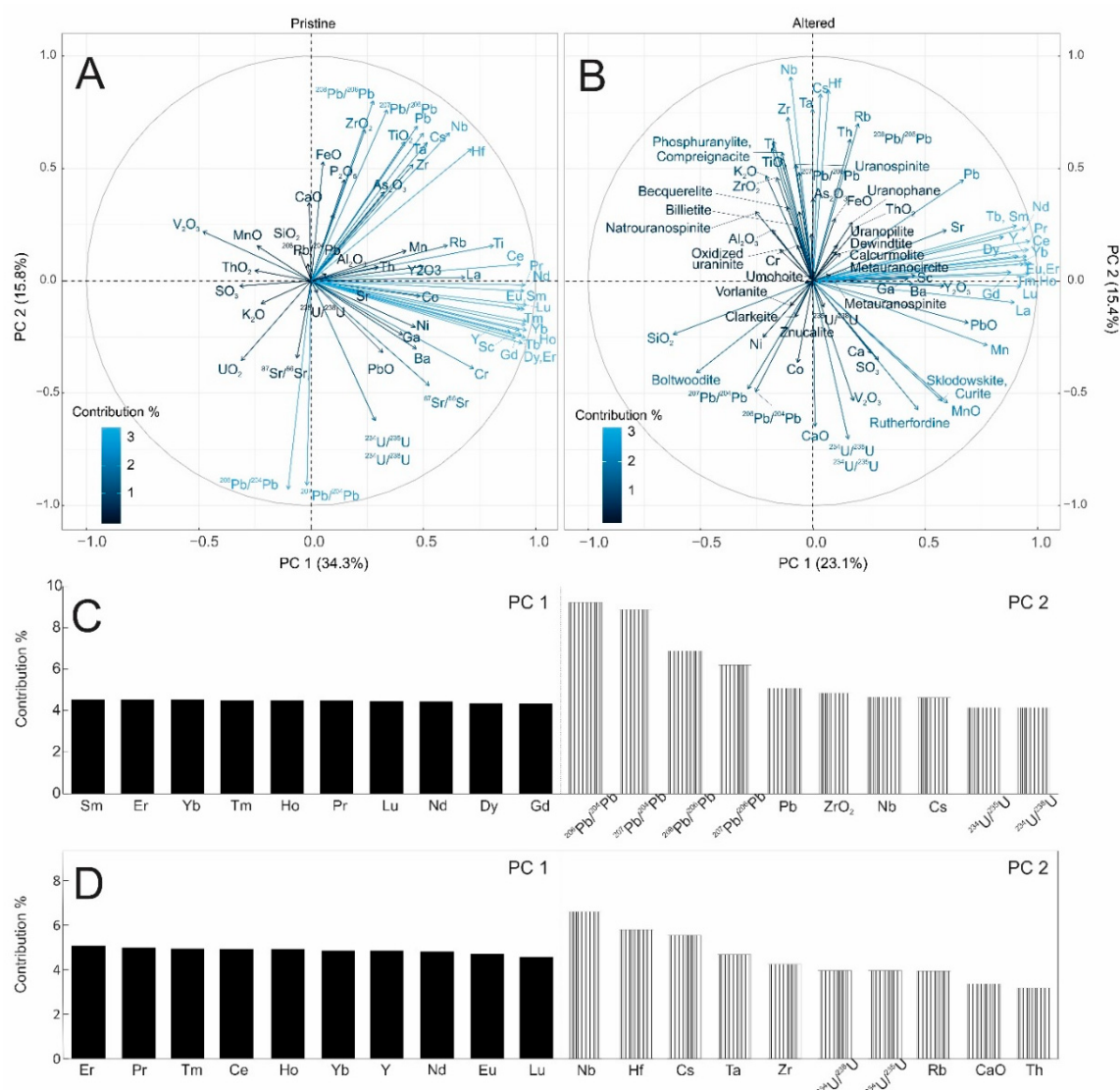


Figure 8. Variable contribution for (A) pristine and (B) altered principal component analysis using all variables (Figure 7). The % each variable contributes is illustrated by the length and color of the individual vector with the longer the vector and lighter the color, the more that particular vector (variable) influences either PC1 or PC2. The top 10 variables that contribute to PC 1 and PC 2 for pristine and altered are given in C and D, respectively (Table S5).

The PCA based solely on the major element abundances (EMP analysis) and the REE data (LA-ICP-MS) is shown in Figure 9. Using solely these variables results in the largest proportion of variance being accounted for (sum of PC 1 and PC 2) within the samples with pristine = 62.3% and altered = 60.2%. However, the grouping for the pristine samples is not as well defined as when based on all the variables (Figure 7). The intrusive non-granite-related samples plot in both negative PC 1 and PC 2, which is controlled by the U, Th, S, and K abundances (Figure S4 and Table S5). In contrast, sandstone samples plot in the negative PC 1 and positive PC 2 space, which is defined by V and Mn concentrations (Figure S4 and Table S5). The other sample types do form groups; however, they are all controlled by their REE compositions, which results in their proximal location on the PCA plots Figure 9 and Figure S4). The altered intrusive non-granite-related samples are the only deposit type that is clearly separated within the altered samples, primarily due to the variation recorded in their ThO₂ concentrations (Figure 9 and Figure S4).

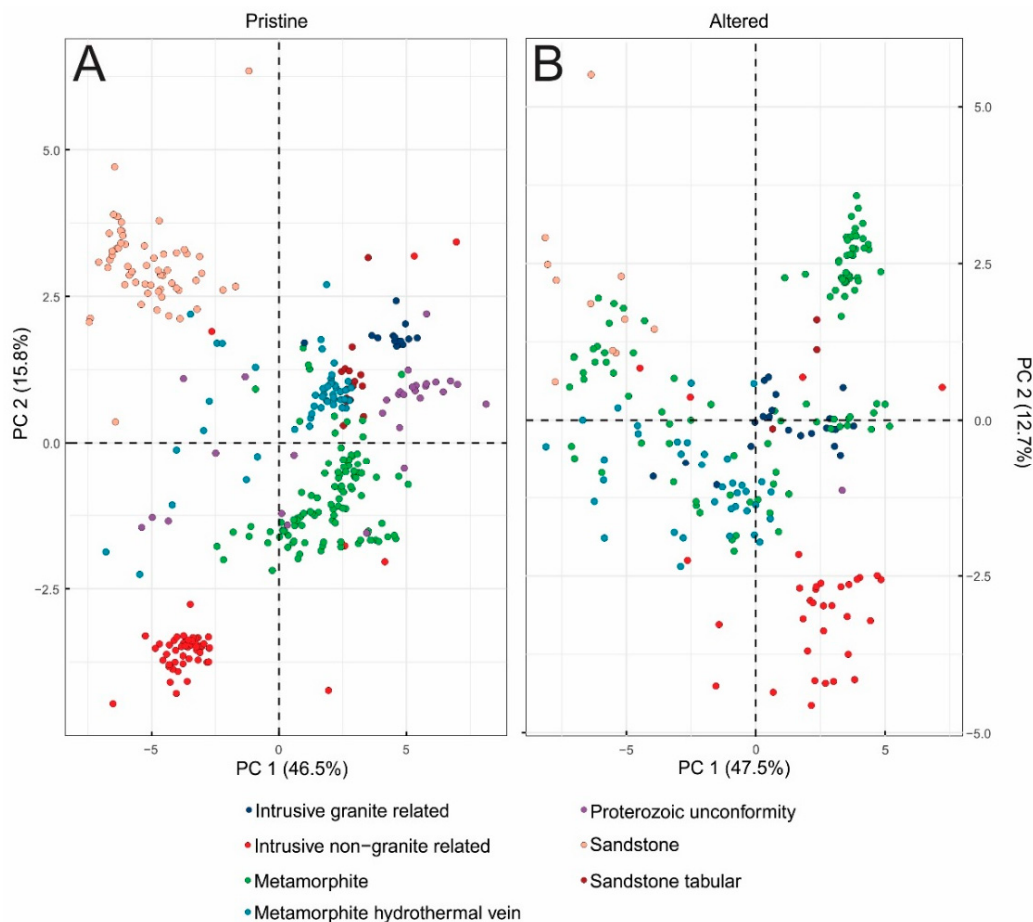


Figure 9. Principal component analysis for (A) pristine and (B) altered uraninite samples using major elements (wt. %; Table S1) and rare earth element abundances (ppm; Table S2). Two principal component analyses were conducted; therefore, the PC 1 and PC 2 are different vectors for both the pristine and altered PCA.

If the isotopic compositions (Sr, Pb, and U) are combined with the major and REE contents, this combination produces a PCA plot that accounts for 52.8% and 49.5% of the variance for pristine and altered samples, respectively (Figure 10 and Figure S5). Using this combination of input variables (EMP, REE, and Sr, Pb, and U isotopic compositions) results in clear and distinct groupings of all deposit types for the pristine samples. Similar to the previous PCA, the various altered uraninite deposit types do cluster, however, there is a large range observed for each group, i.e., different deposit types overlap each other (Figure 10).

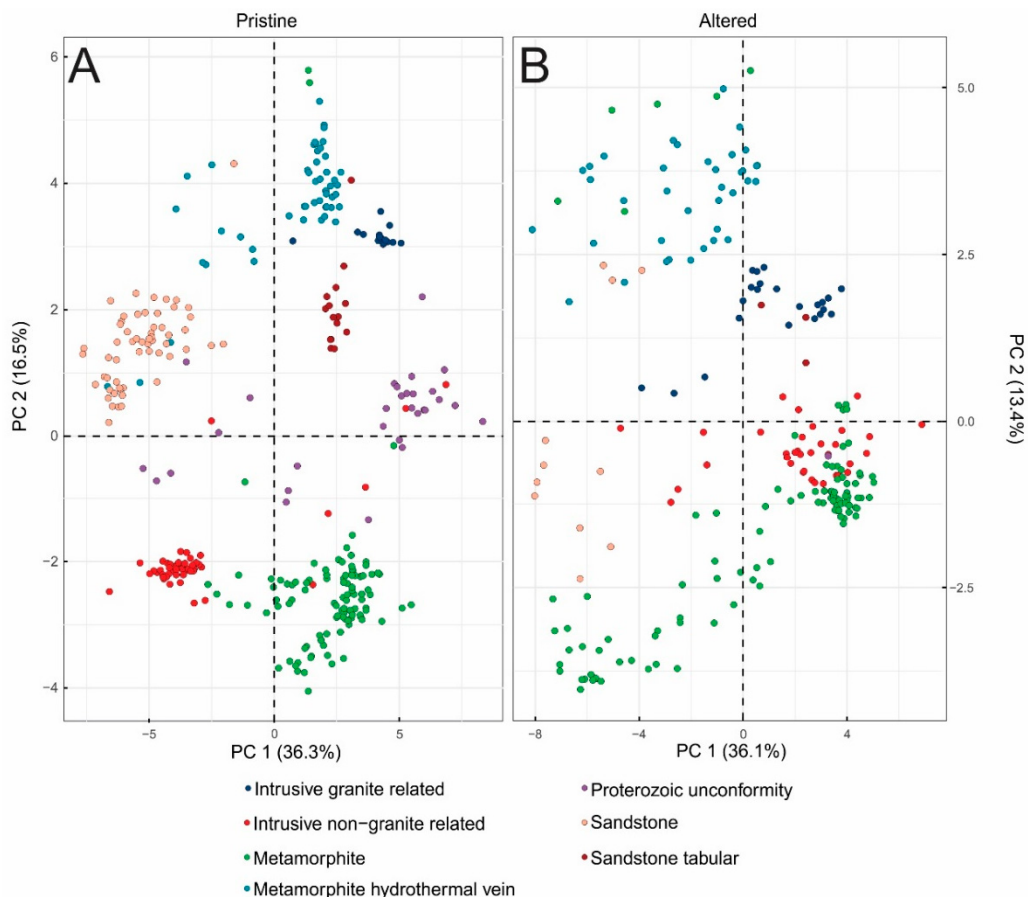


Figure 10. Principal component analysis for (A) pristine and (B) altered uraninite samples using major elements (wt. %; Table S1), REE abundances (ppm; Table S2) and the Sr, Pb, and U isotopic compositions (Table S4). Two principal component analyses were conducted; therefore, the PC 1 and PC 2 are different vectors for both the pristine and altered PCA.

Lastly, the PCA illustrated in Figure 11 is based on the major element abundances and the U, Pb, and Sr isotopic compositions. This accounts for the least amount of variance of 40% and 38% for pristine and altered samples, respectively. However, the pristine samples still form distinct groups. The intrusive non-granite related deposit types are controlled by their K, Th, and S contents, whereas the metamorphite samples are dominated by their Pb isotopic compositions and Zr abundances (Figure 11 and Figure S6, Table S5). Some of the deposit types within the altered samples do form groups. For example, the metamorphite hydrothermal vein samples plot within the negative PC 1 and PC 2 fields. This distribution is controlled primarily by their Ti, Al, and Zr abundances. The sandstone deposit uraninite types are also separated from the remaining samples, however, their range is quite large and does not form a tight grouping (Figure 11). The apparent groupings observed using major element abundances in Figure 11 suggest that with a larger dataset, PCA may be useful for the separation of various geological settings that have undergone alteration.

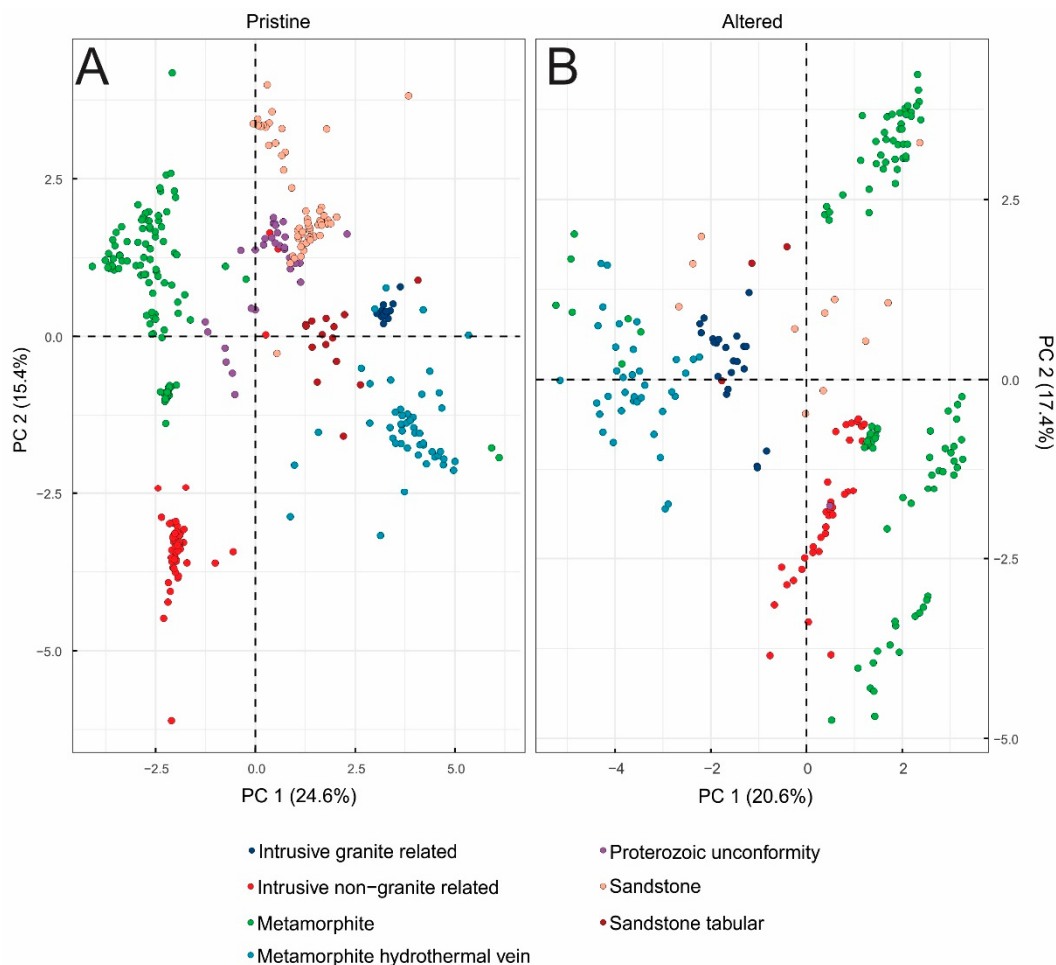


Figure 11. Principal component analysis for (A) pristine and (B) altered uraninite samples using major elements (wt. %; Table S1) and the Sr, Pb, and U isotopic compositions (Table S4). Two principal component analyses were conducted; therefore, the PC 1 and PC 2 are different vectors for both the pristine and altered PCA.

4. Conclusions

The principal component analyses reported here demonstrate the usefulness of this statistical approach to a multivariate dataset involving nuclear samples. PCA has allowed for all seven uraninite deposit types to form discrete groups for pristine uraninite samples. The combination of major element abundances, rare earth element concentrations, and Sr, Pb, and U isotopic compositions as variables in PCA results in >50% of the variance being accounted for with PC 1 and PC 2, and clearly separates the various groups from each other. Hence, this outcome confirms the potential value of the PCA approach for provenance determination of pristine nuclear materials. With regards to altered uraninite samples, additional data (e.g., major element abundances) are required in order to allow for meaningful interpretation of their geological origins.

Supplementary Materials: The following are available online at <http://www.mdpi.com/2075-163X/9/9/537/s1>, Figure S1: Chondrite normalized rare earth element composition of all samples ($n = 23$), Figure S2: XRF map of major elements (U, Pb, Si, Ca, Th, and Fe) for all 23 samples, Figure S3: Isotopic compositions (Table S3) of pristine, altered, and bulk solution aliquots as a function of their age, Figure S4: Variable contribution (%) to pristine and altered samples for PCA using EMP and REEs (Figure 9), Figure S5: Variable contribution (%) to pristine and altered samples for PCA using EMP, REEs, and Sr, Pb, and U isotopic compositions (Figure 10), Figure S6: Variable contribution (%) to pristine and altered samples for PCA using EMP and Sr, Pb, and U isotopic compositions (Figure 11), Table S1: Major element abundance (EMP), Table S2: Minor and trace element concentrations (LA-ICP-MS), Table S3: Secondary minerals identified by PXRD, Table S4: Sr, Pb, and U isotopic compositions, Table S5: Per cent contribution of each variable to the PCA plots in Figures 7,9–11.

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