Potentiality of caro’s acid in leaching of uranium and radioactive measurements from Abu-Rusheid mylonitic gneiss rocks, South Eastern Desert, Egypt

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Research Article

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Abstract

Abu Rusheid area is located at the Southern Eastern Desert of Egypt and composed of Mylonitic gneiss rocks (mineralized rock), Serpentinite rocks, Ophiolitic metagabbro, Ophiolitic mélange, Monzogranites, post- granitic dykes (lamprophyre and dolerite), veins and recent alluvial deposits. This paper is concerned with the study of potentiality of sulphuric and caro's acid in uranium dissolution from Abu Rusheid mineralized rocks. For this purpose, many batch dissolution experiments were conducted. The obtained results showed that 91.5% and 52% uranium leachability for Caro’s acid and dilute sulfuric acid respectively. The reaction mechanism was described using shrinking core models.

Introduction

Resulted of many Exploration activities for uranium resources at south Eastern Desert of Egypt, the discovery of hot gneiss rocks (U, Th, Nb, Ta, Zr, Pb,Y, Zn, W, Sn and Ga), in Wadi (W.) Abu Rusheid area, cut by lamprophyre dykes bearing mineralization (U, REEs, Pb, Ag and Zn) along two shear zones [1–4] as well as, low grade metamorphosed sandstone type-uranium deposit (U and Mo) in W. Sikait,4 km from W. Abu Rusheid [5].

The Abu Rusheid area is located in Egypt's Eastern Desert, approximately 95 km southwest of Marsa-Alam City. Latitudes 24° 36’ 29” and 24° 39’ 22” N, and Longitudes 34° 44’ 40” and 34° 47’ 23” E, define its boundaries. The gneiss rocks are characterized by regional WNW–ESE thrusting and polycyclic deformation and metamorphism. The age of this thrusting is estimated to be between 682 Ma (when the older granitoids were deposited) and 565 to 600 Ma, the time of intrusion of the younger granites [6].

The rocks are heavily deformed in general, with a marked progression from low-grade green schist facies to medium-grade amphibolite facies (Staurolite – Kyanite – Silliminite facies) [7, 8]. Uraninite and pitchblende are primary uranium ores that are found mostly in veins or pegmatites, but also in sedimentary and placer deposits [9]. In weathered zones of initial deposits, secondary uranium ores are identified and precipitated in sediments. Autinite, torbernite, uranophane, carnotite, and schoepite are some of these minerals [10]. Primary uranium is lowering soluble in both dilute acid and carbonate (alkali) solutions than secondary one. So that it economically performs the uranium leaching on proper oxidizing conditions to achieve high uranium extraction.

Uranium leachability is directly proportional to the free acidity of leach solution which range from 20 to 100g/L. However, the using acid leaching medium should not be too strong; very high proton concentrations inhibited the oxidation of U⁴⁺ to U⁶⁺ [11]. At concentrations higher than 200g/L the leaching response was inhibited [12]. This consequently leads to an increase in the leachability of other gangues metals associated uranium.

Several trails around the world use, an oxidant such as sodium chlorate, hydrogen peroxide (H₂O₂) and pyrolusite. But that use resulted in accumulating the heavy metal manganese and sodium chloride as the major contaminant in the acidic waste, which restricts its recycling to the mill [13]. Iron oxides minerals
such as Hematite and magnetite could be played as oxidizing agent in uranium dissolution process since oxidizing the U$^{4+}$ to U$^{6+}$ as in the following eqns:

$$\text{MnO}_2 + 2\text{Fe}^{2+} + 4\text{H}^+ \rightarrow 2\text{Fe}^{3+} + \text{Mn}^{2+} + 2\text{H}_2\text{O} \quad (1)$$

$$\text{UO}_2 + 2\text{Fe}^{3+} + 3\text{H}_2\text{SO}_4 \rightarrow \text{UO}_2\text{SO}_4 + 2\text{FeSO}_4 + 6\text{H}^+ \quad (2)$$

Caro's acid is the best alternative oxidant to pyrolusite and sodium chlorate. Since, it was easier to handle, provided better control and a completely automated system, ensured continual oxidant supply and almost eliminating manganese from waste effluents [14, 15].

Batch leaching tests on Ranger, Nabarlek, Koongarra, and Jabiluka ores showed that uranium extraction was unaffected by choice of oxidant, but that Caro's acid reduced acid and lime requirements by 15–25% and 20–30% respectively[16]. To gain a better understanding of the leaching process and its operation, kinetic studies should be concerned. Since the leaching kinetics interprets the complex behavior of leaching process occur on grain particles [17]. Thus, this study includes investigation and discussion of the leaching kinetics and dissolution of uranium on Caro's acid solution.

For leaching uranium from Abu-Rusheid uranium mineralization, phosphoric acid was examined. Results obtained encourage recommending phosphoric acid as an alternative to sulphuric acid leaching agent for uranium [18]. It is demonstrated by the laboratory experiment that heap leaching is suited to process Abu-Rusheid uranium mineralization; D263B strong base anion resin can effectively concentrate and purify the obtained pregnant solutions to produce a quality products, namely yellow cake [19].

The present paper concerned with study the leachability of Abu Rusheid uranium minerals ores situated at the Southern Eastern Desert of Egypt. The potentiality of Caro's acid was established in batch experiments followed by kinetics investigation.

**Geologic Setting**

**Regional geology**

Abu Rusheid is a small part of the Arabian-Nubian Shield that outcrops across an area of around 73.5 km$^2$ in Egypt's South Eastern Desert (ANS). The shield, which covers more than 6 x 10$^6$ km$^2$ [20, 21], is one of the world's greatest Neoproterozoic crustal development episodes, having been exposed by uplift and erosion during the Oligocene and younger periods [22, 23]. The Arabian-Nubian Shield (ANS) is made up of Precambrian rocks that can be found in western Arabia and northeastern Africa on both sides of the Red Sea (Fig. 1a; [24]). The majority of the ANS is made up of juvenile continental crust [25–27], which crust is formed by mantle-derived melts. When the Mozambique ocean closed due to arc terrane accretion, it formed between 900 and 550 Ma [22]. Abu Rusheid area is a part of the Arabian-Nubian Shield and can be considered a key domain in that shield, beside its very complex structures. In addition, this area is considered as the southeastern extension of the Migif-Hafafit metamorphic complex [28–30],
which is highly tectonized and characterized by presence of several types of mineralization and alteration processes along time spans. Abu Rusheid-Sikait area is bordering to the major shear zone recognized by many authors as the Nugrus thrust fault [31] or the Nugrus strike-slip fault [32] and or Shait–Nugrus shear zone [33].

**Geology of the study area**

The main rock units encountered in this area are grouped from older to younger as follows; (a) Mylonitic gneiss rocks, (b) Serpentinite rocks, (c) Ophiolitic metagabbro, (d) Ophiolitic mélange, (e) Monzogranite rocks, (f) Post-granitic dykes and veins (Fig. 1b). The mylonitic gneiss rocks (2.0 km²) represent the oldest rocks exposed in W. Abu-Rusheid area. These mylonitic gneisses were originally identified as psammitic gneisses [34–37], occur in the field down thrust the ophiolitic mélange and foliated in ENE-WSW direction (Fig. 2a). The mylonitic gneiss rocks of Abu Rusheid area is characterized by development of mylonitic fabric close to the shear and contact zones, low to moderate topography and highly sheared as well as show a well-developed planar banding, gneissosity and folding. They are highly metasomatized and reflect high radioactive anomalies that contain abundant crystals of thorite, uranothorite, zircon, fluorite and Nb-bearing minerals (samarskite, pyrochlore, betaite) [38]. They are cross cut by three main shear zones; the first two shear zones are parallel to each other (NNW- SSE) and perpendicular to the third one (ENE-WSW), the NNW-SSE shear zones were emplaced by lamprophyre dykes which act as a good barrier (physical and chemical trap) capturing, adsorbing and protecting the uranium minerals that introduced to the gneiss with moving the uranium-rich hydrothermal solution (Fig. 2b-d). They have general NNW trend and dips steeply (80º-85º) toward SW. The serpentinite is only found in a small part in the NW corner of the mapped area. Abu-Rusheid –Sikait granitic pluton is an elongated body extending NW-SE for about 12 km with width about 3 km. The granitic rocks occupy the major part of the mapped area surrounding the mylonitic gneisses (Fig. 1b). They are represented from the NW direction by porphyritic biotite monzogranite followed by deformed biotite monzogranite and two mica monzogranite, whereas the muscovite granites occupy the SE part of the pluton [39].

**Characteristics of the Radioactive Mineralization**

The mylonitic gneiss rocks are characterized by well-developed Li, U, Th, Nb, Y, Au, As, Pb, Zn, Sn and Cu enrichments respectively. The common minerals that have been recognized in both mylonitic gneiss and lamprophyres are represented by zinnwaldite, U- minerals (soddyite, uranophane, autunite, kasolite, torbernite and meta- zeunerite) (Fig. 3a,b), Th-minerals (thorite and uranothorite), and gold, cassiterite, scheelite, xenotime (Y), allanite, zircon, fluorite, monazite (Ce) and Nb-Ta minerals. Thorium minerals (thorite and uranothorite) are clear in gneissic rocks only. Others minerals were recorded and studied such as(sulphides,tin, zinc–manganese minerals, fluorite, xenotime, zircon, allanite, monazite, iron oxide minerals, clay minerals (illite + kaolinite) and calcite and micas at Abu Rusheid area [4]. The average U/Th ratio in mylonite and lamprophyre dykes (1.32 and 40.0) is exceeding than average of crust (0.33), illustrate uranium gain, whereas in mylonitic gneiss samples, their U/ Th ratio (0.24) is less than the average of crust, denote uranium loss.
The descending (infiltrational) mineralization-bearing hydrothermal solutions are mainly derived from meteoric water, migrating under gravity from high relief peripheral hot uranium-muscovite-biotite granite to low central part of the basin (mylonitic gneiss rocks) and redeposited along banding planes and fractures in mylonitic gneiss.

After the emplacement of lamprophyre dykes (mantled-derived with high temperature and volatiles, as well as, CO$_2$), the ascending (exfiltrational) hydrothermal solutions are dominantly derived from groundwater, and are mainly derived by high gas pressures penetrating into the basin, along banding planes and fractures in the host rocks with low temperatures and containing $\text{F}^-$ and $\text{CO}_3^{2-}$, $\text{PO}_4^{3-}$ and $\text{H}_2\text{O}$ caused redistribution; transportation and redeposition of the base metals (Pb, Zn, Cu and As), precious metals (Au and Ag) and in HFSE (e.g. Nb, HREE and Y), along the channel ways (banding of the gneissic rocks and fractures) to the shear zones (NNW-SSE ) and precipitated in boxworks and adsorbed on clay minerals.

The lamprophyre dykes and mylonitic gneiss are characterized by ferrugination (high total iron). The latter indicates the presence of alkali hydrothermal solution, may precipitate $\text{Fe}^{3+}$ and $\text{U}^{6+}$ within micro-fractures in the form of iron-oxy-hydroxides rich in uranium. The occurrences of U-minerals in the pores (fractures and veins) indicate that, these minerals were deposited from solutions descending or ascending that permeated the micro-porous. Desulfidization process took place in the mylonitic gneiss rocks and lamprophyre dykes, where sulfide minerals such as pyrite, chalcopyrite and sphalerite are easily oxidized by ground water in the presence of oxygen to produce either ferric oxide or its hydrate analog, leaving box works. The release of Ca and F during illitization of plagioclases and biotite may combine together to form calcium fluoride ($\text{CaF}_2$). This kind of alteration by acid solution is responsible for leaching mineralization from the banded mylonitic gneiss rocks, transported and adsorbed on the margins of NNW-SSE lamprophyre dykes (chemical trap for mineralization).

**Experimental**

**Characterization of Abu Rusheid mylonite uranium mineralization**

Abu Rusheid mylonite representative samples were collected from Abu Rusheid area which is located in the South Eastern Desert, Egypt. The ore sample was firstly crushed to 297 µm and 149 µm then ground to 74 µm particle sizes. Sample characterization by XRF showed the presence of Ba, Pb, Fe, Mo, Nb, Zr, Y, Sr, U, Zn, Mn, Ca, K, S, Si and Al. The representative sample is characterized chemically as major and trace elements by ignition at 1000 °C before preparing solutions A and B of Shapiro and Brannock method for analysis of silicate rock [40]. Uranium analysis was performed spectrophotometrically in low concentrations using Arsenazo (I) method [41], while in high concentrations by titration against ammonium meta vanadate [42].
Analysis performed using UV-spectrophotometer “single beam multi-cells-positions model SP-8001”, Metretech Inc., version 1.02.

**Leaching Experiments**

Leaching experiments were conducted with a conical flask placed in a constant temperature water bath, with mechanical stirring and a thermometer for temperature control. 250 g of representative sample was introduced into the reactor to yield a solid/liquid ratio of 1/3. After leaching, the sample was washed with acidified water. The effects of changing various parameters such as sulfuric acid and $\text{H}_2\text{O}_2$ concentration, temperature, leaching time, grin size, S/L ratio and Eh were investigated. Used reagents ($\text{H}_2\text{SO}_4$ and $\text{H}_2\text{O}_2$) are of the analytical grade. Eh and pH were measured accurately using the pH-meter model (HAANA pH-mV-temp).

**Radioactive measurements**

Equivalent uranium and thorium contents were determined radiometrically by using multi-channel analyzer Gamma-ray detector (Gamma-spectrometer technique) in the Laboratories of the Nuclear Materials Authority, Egypt. The studied samples are crushed to about 100 mesh then the containers is filled with about 300-400gm of the crushed samples, sealed well and left for at least 21 days to accumulate free radon for attaining radioactive equilibrium. The measurements are expressed in (ppm) for eU & eTh.

**Results And Discussion**

**Radioactivity of the Abu Rusheid mylonitic gneiss samples,**

The radioactivity of the studied mylonitic gneiss samples, Abu Rusheid area, SED, Egypt were measured then related to the standards for U and Th provided by International Atomic Energy Agency (IAEA). The measurements are given in Table. 1. The table shows that the eU shows very high level from 69 and 1106 ppm with an average of 364.84 ppm, while the eTh content ranges between 45 and 402 ppm with an average of 147.59 ppm. The eTh/eU ratios range from 0.04–5.43 with an average of about 0.86. These values indicated that; there is uranium leach in. On the eU-eTh plot Fig. (4a); eU shows very high level and show the relationships among eU and eTh for the mineralized mylonitic gneiss, where the relation of eU and eTh is negative due to the slight enrichment of uranium. The increase of eTh content in some samples is regard due to the presence of uranothorite. Also, the eTh/eU ratio versus eU (Fig. 4b) show negative correlation, in which eTh/eU ratio increases with a decrease of eU in the most of representative samples reflect the presence of enrichment uranium mineralization.

**Table. 1 eU (ppm), eTh (ppm) contents and eTh/eU ratios of the studied mylonitic gneiss samples,**

Abu Rusheid area, SED, Egypt.
Chemical characterization of uranium mineralization sample

Analysis of the working technological uraniferous mylonite sample has revealed a uranium assay of 400 ppm. Major elements analyzed involved 73.8% SiO₂, 11.9% Al₂O₃ and 2.9% total iron as Fe₂O₃ as shown in the table:

| S. No | eU  | eTh  | eTh/eU | S. No | eU  | eTh  | eTh/eU |
|-------|-----|------|--------|-------|-----|------|--------|
| 1     | 161 | 240  | 1.49   | 18    | 161 | 240  | 1.49   |
| 2     | 125 | 178  | 1.42   | 19    | 125 | 178  | 1.42   |
| 3     | 166 | 171  | 1.03   | 20    | 166 | 171  | 1.03   |
| 4     | 172 | 230  | 1.34   | 21    | 172 | 230  | 1.34   |
| 5     | 122 | 246  | 2.02   | 22    | 122 | 246  | 2.02   |
| 6     | 134 | 220  | 1.64   | 23    | 134 | 220  | 1.64   |
| 7     | 141 | 174  | 1.23   | 24    | 141 | 174  | 1.23   |
| 8     | 94  | 334  | 3.55   | 25    | 94  | 334  | 3.55   |
| 9     | 74  | 402  | 5.43   | 26    | 74  | 402  | 5.43   |
| 10    | 174 | 162  | 0.93   | 27    | 174 | 162  | 0.93   |
| 11    | 128 | 125  | 0.98   | 28    | 128 | 125  | 0.98   |
| 12    | 124 | 95   | 0.76   | 29    | 124 | 95   | 0.76   |
| 13    | 178 | 147  | 0.82   | 30    | 178 | 147  | 0.82   |
| 14    | 159 | 66   | 0.41   | 31    | 159 | 66   | 0.41   |
| 15    | 264 | 213  | 0.81   | 32    | 264 | 213  | 0.81   |
| 16    | 86  | 63   | 0.73   | 33    | 86  | 63   | 0.73   |
| 17    | 113 | 51   | 0.45   | 34    | 113 | 51   | 0.45   |
| 18    | 320 | 45   | 0.14   | 35    | 320 | 45   | 0.14   |
| 19    | 437 | 122  | 0.28   | 36    | 437 | 122  | 0.28   |
| 20    | 418 | 106  | 0.25   | 37    | 418 | 106  | 0.25   |
| 21    | 133 | 162  | 1.22   | 38    | 133 | 162  | 1.22   |
| 22    | 157 | 181  | 1.15   | 39    | 157 | 181  | 1.15   |
| 23    | 312 | 143  | 0.46   | 40    | 312 | 143  | 0.46   |
| 24    | 241 | 189  | 0.78   | 41    | 241 | 189  | 0.78   |
| 25    | 168 | 288  | 1.71   | 42    | 168 | 288  | 1.71   |
| 26    | 532 | 92   | 0.17   | 43    | 532 | 92   | 0.17   |
| 27    | 392 | 104  | 0.27   | 44    | 392 | 104  | 0.27   |
| 28    | 632 | 169  | 0.27   |       |     |      |        |
| 29    | 69  | 90   | 1.30   |       |     |      |        |
| 30    | 106 | 169  | 1.59   |       |     |      |        |
| 31    | 215 | 118  | 0.55   |       |     |      |        |
| 32    | 823 | 60   | 0.07   |       |     |      |        |
| 33    | 894 | 103  | 0.12   |       |     |      |        |
| 34    | 912 | 84   | 0.09   |       |     |      |        |
| 35    | 944 | 79   | 0.08   |       |     |      |        |
| 36    | 1044| 47   | 0.04   |       |     |      |        |
| 37    | 1106| 115  | 0.10   |       |     |      |        |
| 38    | 743 | 109  | 0.15   |       |     |      |        |
| 39    | 751 | 48   | 0.06   |       |     |      |        |
| 40    | 919 | 98   | 0.11   |       |     |      |        |
| 41    | 372 | 136  | 0.36   |       |     |      |        |
| 42    | 321 | 124  | 0.38   |       |     |      |        |
| 43    | 267 | 177  | 0.66   |       |     |      |        |
| 44    | 410 | 219  | 0.53   |       |     |      |        |

Average 364.84 147.59 0.86
in Table (2) as well as some trace elements. The leaching of uranium minerals by sulfuric acid requires the presence of Fe$^{+3}$ in order to oxidize U$^{+4}$ to U$^{+6}$ to form U(SO$_4$)$^{4-}$ dissolved in leach liquor but Fe$^{+2}$ is abundant than Fe$^{+3}$ in the studied sample. Thus, it's necessary to presence of oxidant to oxidize Fe$^{+2}$ to Fe$^{+3}$, which their presence serves an electron transfer catalyst between an oxidant and UO$_2$ [43].

$$\text{UO}_2 + 2\text{Fe}^{3+} + 3\text{H}_2\text{SO}_4 \rightarrow \text{UO}_2\text{SO}_4 + 2\text{FeSO}_4 + 6\text{H}^+ \quad (1)$$

| Oxides   | Wt% | Trace elements | ppm |
|----------|-----|----------------|-----|
| SiO$_2$  | 73.8| Nb             | 680 |
| TiO$_2$  | 0.1 | Zr             | 2650|
| Al$_2$O$_3$ | 13.9| Y              | 750 |
| Fe$_2$O$_3$ | 0.7| Pb             | 800 |
| FeO      | 2.2 | Ga             | 85  |
| MnO      | 0.1 | Cu             | 190 |
| MgO      | 0.85| Zn             | 1450|
| CaO      | 0.67| U              | 400 |
| Na$_2$O  | 3.0 | Th             | 573 |
| K$_2$O   | 4.3 | Nd             | 84  |
| P$_2$O$_5$ | 1.06| As             | 12  |
| L.O.I    | 0.16| Sn             | 21  |

Nerenest derived Eq. (1); could follow the ferric iron concentration on leach liquor, since it's correlated the redox potential with Fe$^{+3}$ / Fe$^{+2}$ ratio:

$$Ec = 397 + 0.1984T \log ([\text{Fe}^{3+}]/[\text{Fe}^{2+}]),$$

Where Ec is the solution potential relative to the saturated calomel electrode (mV) at 350°C, T is in K. The another model equation is predicted from plant data of hartebeestfontein operation:

$$Ec = 399.8 + 0.14 \log A + 0.2 \log ([\text{Fe}^{3+}]/[\text{Fe}^{2+}])$$

Where $A$ is the sulfuric acid concentration (g/L) [44].
The presence of MnO (0.1%) in mineralized ore could be useful as oxidant of iron as in the following Eq. (2)

\[
2\text{Fe}^{2+} + \text{MnO}_2 + 4\text{H}^+ \rightarrow 2\text{Fe}^{3+} + \text{Mn}^{2+} + 2\text{H}_2\text{O} \tag{2}
\]

The above equation shows that the reaction requires large amounts of hydrogen ions to take place, which consequently consume the acid leach.

The acid required to achieve the previous equivalent oxidation of ferrous ion could be reduced by 50% if using another oxidant such as sodium chlorate or Caro`s acid as evident in the following equation: 2\text{Fe}^{2+} + \text{H}_2\text{SO}_5 + 2\text{H}^+ \rightarrow 2\text{Fe}^{3+} + \text{H}_2\text{SO}_4 + 2\text{H}_2\text{O} \tag{3}

\[
2\text{Fe}^{2+} + \frac{1}{3}\text{ClO}_3^- + 2\text{H}^+ \rightarrow 2\text{Fe}^{3+} + \frac{1}{3}\text{Cl}^- + 2\text{H}_2\text{O} \tag{4}
\]

The Nabarlek mill in Northern Australia use Caro’s acid (\text{H}_2\text{SO}_5) as oxidant agent which manufactured on site by the reaction of concentrated sulfuric acid with 50–70% hydrogen peroxide in ratio of 3:1. Because of its greater stability, Caro’s acid is a more efficient oxidant than hydrogen peroxide itself [45].

Potentiality of Caro’s acid of uranium dissolution from Abu Rusheid mineralized sample was investigated by performs batch experiments as the following;

**Relevant dissolution factors**

**Effect of Free acidity**

Investigation of free acidity effect was achieved by varying acid strength from 0.25 M to 3 M, at 3:1 ratio of hydrogen peroxide (Caro’s acid ) at constant, temperature (40°C), particle size (300µm) ,1:2 solid – liquid ratio and stirring speed (300 rpm) for 6hr. Uranium concentration and Eh were analyzed after solid liquid separation at each experiment. The obtained data was graphed as the following Figs. 5 and 6.

(H\text{SO}_4 and Caro’s acid) on uranium leachability (H\text{SO}_4 and Caro’s acid)

As shown in Fig. 5, the mineralized ore need large amounts of sulfuric acid to achieve almost uranium leaching. Since the leachability record low percent 9 to19% at free acidity 0.25M to 1M with redox potential 215 to 275 mv then increase periodically as in Fig. 6. Which; owing to the studying mineralized ore containing a lot of acid consumer minerals such as fluorite mica.

Introducing the oxidant as \text{H}_2\text{O}_2 mixed with sulfuric acid as the leach liquor improve the leaching performance and increase the leachability from 19 to 41% and redox potential from 275 to 425mv. Since increase the oxidation rate of iron consequently increase the uranium dissolution. In other hand, reduce consuming of sulfuric acid by gangues metals associated uranium and oxidation. 1M sulfuric acid was used as the maximum concentration of ore leaching to decrease the dissolution of gangues metals such...
as silica and alumina which exist as the major oxides in ore. High free acidity, access to the uranium inside individual particles.

**Effect of contact time**

By varying the parameter in the range of 3–7 hr with keeping other variables as follows: 1M free acidity, temperature 40°C, particle size 300μm and solid/liquid ratio 1:2.

The results are shown in Fig. 7, about 31.5, 17 % of uranium values were leached in 3 hr using 1M H₂SO₄ and Caro’s acid respectively. Then, gradually increased to about 41.8, and 19.5% in 4–7 hr reaction time, respectively. The less increase in leachability of uranium is result of simultaneous increase of acid consumption, during leaching in case of using only sulfuric acid since.

**Effect of grain size**

The leachability as function of grain size was established by varying the grain size from 425 to 75 μm and constant the other parameters and illustrated in Fig. 8. As shown in Fig. 7, the leachability increase gradually with grain size decrease could be attributed to increase in the area of reaction consequently increase dissolution of the contained uranium and oxidant (MnO) on solution.

**Effect of temperature**

Temperature studied range were from room temperature to lower 85 °C to avoid dissolution of associated minerals like monazite and rare earth silicate [40]. Consequently, temperature effect was investigated by varying reaction temperature from 30 to 70°C and fixing the other parameters to avoid dissolution of associated minerals like monazite and rare earth elements silicate. Consequently, temperature effect was investigated by varying reaction temperature. Figure 9, shows that, gradual increase in uranium leachability with temperature increase, as increase from 19.2 to 36% at temperature from 30 to 70 °C in case of diluted H₂SO₄. On using Caro’s acid, there is slightly increase in leachability from 41 to 44 at temperature 30–50 °C, then tend to decrease with temperature increase from 50 to 70 °C.

From above, we can conclude that the high temperature accelerate the uranium dissolution when using diluted H₂SO₄. On other hand, accelerate firstly then decrease after that on using Caro’s acid which can be attributed to the dissociation of H₂SO₅ into H₂SO₄ and water at temperature above 50 °C.

**Effect of S/L ratio**

The effect of S/L ratio was investigated by varying through the S/L ratio range between 1:2 to 1:5, while fixing the other parameters. The obtained results (Fig. 10) show that the uranium leachability increased gradually as the ratio decrease from 1:2 to 1:5 in case of H₂SO₄. On other hand, multiplied as ratio decrease from 1/2 to 1/3 then increase gradually to reach 96.5% from 1/3 to 1/5 ratio. Which proved that the Caro’s acid save in sulfuric acid consuming during leaching and give concentrated uranium leach liquor.
Leaching kinetics

In order to gain more understood of leaching behavior, the uranium leachability % values were determined along each experiment performed with time. The obtained results would be formulated in core shrinking models which the best model can describe the kinetics and reaction mechanism.

Shrinking core model derived equations as the following [46].

\[ 1 - 3 (1-X) \ 2/3 + 2 \ (1-X) = K_d t, \ (5) \]

when the leaching rate controlled by diffusion through product layer

\[ 1 - (1 - X) \ 1/3 = K_c t, \ (6) \]

when the leaching rate controlled by chemical reaction on surface layer

\[ X = K_f, \ (7) \]

when leaching rate controlled by film diffusion

While \( K_d, K_c \) and \( K_f \) are the apparent rate constants (\( \text{min}^{-1} \)), \( t \) is the leaching time, and \( X \) is the metal fraction.

Effect of free acidity

To elucidate the mechanism of the Caro’s acid concentration on uranium dissolution, the shrinking core models as described by Eqs. (5) and (6) are operated and graphed as in Fig. (11 a,b).

(a) diffusion reaction control; (b) Chemical reaction control

As shown in Fig. 11a,b, there are a linearity relation of slopes > 0.9 between the uranium fraction mole dissolved at varied times and Caro’s acid concentration. There are very little changes between two reaction models, so that another relationship will be plotted between logarithm apparent rate constant against logarithm \( \text{M H}_2\text{SO}_4 \) as in Fig. 12, to differentiate between them.

Figure 12, shows that the linearity in diffusion model than other chemical one, which proved that the leaching rate is controlled by diffusion process. The slope of 0.986 indicates strong dependence of the rate on sulfuric acid concentration.

Effect of Temperature

Measuring uranium adsorbed each time interval during dissolution experiments with varying reaction temperature and fixes other conditions. The obtained resulted used in construction of two shrinking core models to be examined at different temperatures as shown Fig. 13a,b.
As observed, the two models have good linear fits and their $R^2$ values are close to each other. Therefore, it is difficult to distinguish between these reaction models.

Activation energy is the important parameter that can be used to justify the rate-determining step in hydrometallurgical process. It was calculated based on the Arrhenius equation [47]:

$$K=A e^{-\frac{E_a}{RT}}; \quad \ln K = -\frac{E_a}{RT}$$

Where $K$ is the rate constant, $E_a$ is the activation energy, $R$ is the ideal-gas constant (8.314 J/Kmol), $T$ is the temperature in K and $A$ the frequency or pre–exponential factor.

As shown in Fig. 14, Arrhenius equation was plotted as $\ln(K)$ versus $(1/T)$ for each temperature, and the activation energies were calculated from the slopes of straight lines where the slope is $-\frac{E_a}{R}$.

From above Figure the calculated activation energy $E_a$ are $-11.03$ and $-8.48$ kJ/mole for chemical reaction control and diffusion reaction control respectively. These indicate that the reaction rate is controlled by chemical reaction on the particle surface. Consequently, temperature have pronounced effect on reaction rate since, its increase lead to dissociate the Caro's acid into $\text{H}_2\text{O}$ and $\text{H}_2\text{SO}_4$ on the particle surface, then increase the diffusion of ions dissociated through the liquid film and particle layer.

**Conclusion And Future Applications**

Batch experiments show the superiority of Caro's acid in uranium dissolution with low concentration than dilute sulfuric acid. Leachability of 91.5% and 52% uranium for Caro's acid and dilute sulfuric acid respectively was obtained. Shrinking core models used to determines the reaction mechanism, which shows that the uranium dissolution in Caro's acid controlled by diffusion reaction model. Arrhenius plot were constructed to calculate the activation energy, which is found ranging from $-11.03$ to $-8.48$kJ/mole.

In future, we can applied a two-stage leaching for achieving maximum uranium solubilisation with minimum dissolution of gangue elements like chlorite and biotite in sulfuric acid leaching of uranium ore. Firstly, leaching with Caro's with optimized concentration of less than 100g/l $\text{H}_2\text{SO}_4$ until Eh and pH of leach liquor reached 450 and 1.6 respectively followed by leaching with diluted $\text{H}_2\text{SO}_4$ that keep the Eh and pH of leach liquor at the same values.

**Declarations**

**Conflict of Interests**

The authors declare that they have no known competing financial interests or personal relationships that could have appeared to influence the work reported in this paper.

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Figures

Figure 1
(a) Map of the Arabian-Nubian Shield (ANS) modified after [23]. (b) Geologic map of Abu Rusheid area, SED, Egypt, after [39].

Figure 2
Field photographs showing, (a) Ophiolitic melange thrusted over mylonitic gneiss. b) NNW shear zone. (c) Lamprophyre dyke. (d) Altered lamprophyre dyke associated with secondary uranium mineralization.
Figure 3

ESEM Micrograph and EDX chart of, a) - Uranophane and b) - Autunite of investigated rocks at Abu Rusheid area.

Figure 4

Radioelements variation diagrams for the mylonitic gneiss rocks, Abu Rusheid area, SED, Egypt. a- eU vs eTh b- eU vs eTh/eU.
Figure 5

Effect of free acidity of leach solution (H2SO4 and Caro's acid) on uranium leachability

Figure 6

Effect of Eh on uranium leachability (H2SO4 and Caro's acid)
Figure 7

Effect of contact time on uranium leachability % using H2SO4 and Caro’s acid

Figure 8

Effect of grain size on uranium leachability % using H2SO4 and Caro’s acid

Figure 9

Effect of Temperature on uranium leachability % using H2SO4 and Caro’s acid
Figure 10

Effect of S/L ratio on uranium leachability % using H2SO4 and Caro's acid

Figure 11

The kinetic curve of uranium leaching at different concentration of Caro's acid based on different models. (a) diffusion reaction control; (b) Chemical reaction control
**Figure 12**

Log-log plot of the rate constant versus H2SO4 concentration based on different models. (a) Diffusion reaction control; (b) Chemical reaction control.

**Figure 13**

The kinetic curve of uranium leaching at different temperatures based on different models (a) Diffusion reaction control; (b) Chemical reaction control.

**Figure 14**

Arrhenius plot for the uranium leaching at different Temperatures based on different models (a) Diffusion reaction control; (b) Chemical reaction control.