Supplementary Information

Reactive Transport of U and V from Abandoned

Uranium Mine Wastes.

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Additional Materials and Methods. *Acid Extractable Procedure*. Acid digestions were conducted to assess the total acid extractable metal concentrations between unreacted and reacted mine wastes. For unreacted mine waste 3 mL of hydrochloric acid (HCl), 3 mL of nitric acid (HNO₃) and 3 mL of hydrofluoric acid (HF) were added into 50 mL Teflon digestion tubes containing 1 ± 0.002 g of mine waste sample. Similarly for reacted mine waste (collected after reaction with CH₃COOH) 1 mL of each reagent (HCl, HF and HNO₃) was added to 50 mL Teflon digestion tubes containing 0.1 ± 0.002 g of mine waste sample. All reagents are of Ultra High Purity (UHP) grade. The digestion tubes were then heated using a Digi prep MS SCP Science block digester at 95°C for 2 h, followed by dilution of acid extracts from reacted and unreacted mine waste to 50 (unreacted) and 25 (reacted) mL using 2% HNO₃. The diluted samples are then filtered using 0.45 µm filters to remove any suspended or undissolved solids before analysis.

Inductively Coupled Plasma (ICP). Acid extracts and aliquots from column experiments were analyzed for elemental concentrations using a PerkinElmer Optima 5300DV ICP-OES. Trace metals below the detection limit of ICP-OES were measured using PerkinElmer NexION 300D (Dynamic Reaction Cell) Inductively Coupled Plasma-Mass Spectrometer (ICP-MS). Both ICPs are calibrated with calibration standards and QA/QC measures are taken to ensure quality data. *Transmission Electron Microscope (TEM).* A TEM analysis was conducted on mine waste samples to identify the crystallinity of various U-V bearing minerals using a Selected Area Electron Diffraction technique (SAED). A JEOL 2010 High Resolution Transmission Electron Microscope (HR-TEM) fitted with a GATAN Orius high speed CCD camera and an Oxford INCA system with an ultra-thin window Energy Dispersive Spectroscopy (EDS) detector was used in sample analysis. Drops of unaltered sediments suspended in acetone were dropped onto

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standard holey carbon film-covered Cu TEM grids using a pipette. Acetone was allowed to evaporate in air and then the dry samples were loaded onto a Gatan cryotransfer holder to observe frozen hydrated specimens for cryo-electron microscopy. Before loading the grid, zeolite crystals inside the cryotransfer holders were allowed to dry overnight using a vacuum pump to avoid cooling obstruction due to frozen water inside the holder. The dry and loaded cryotransfer holders were then loaded into the HR-TEM and allowed to reach a temperature of -180°C using liquid nitrogen before the analysis. The INCA EDS detector was used to determine the elemental composition of the specific targeted crystal. After identification, SAED analysis was performed on the identified crystals to observe their crystallinity.

Synchrotron Micro- X-ray Fluorescence mapping (μ -SXRF). A polished section of the mine waste sample was mapped for elemental distribution using the μ -SXRF beam line (BL) 10-2 at the Stanford Synchrotron Radiation Light Source (SSRL). The maps were collected using a Si (111) phi, 90 double crystal monochromator at a 50 μ m resolution upto 17200 eV X-ray energies, just above the U L_{III} edge. All data processing was conducted using the Microanalysis Toolkit software program.

Reactive Transport Modelling (PFLOTRAN). Below are the supporting equations that were used in PFLOTRAN to calculate the change in concentration while taking the dominant processes into account. The total aqueous concentrations (Ψ_j) [ML⁻³] for a particular species j is given by equation [2], where C_j and C_i [ML⁻³] represent concentrations of jth primary species from 1 to N_c and ith secondary species that varies from 1 to N_{sec} in the liquid phase.

$$\Psi_{j} = C_{j} + \sum_{i=1}^{N_{sec}} v_{ji} C_{i}$$
(2)

The secondary species concentrations (C_i) are estimated in terms of primary species concentration using equation [3] with known values of equilibrium constant K_{eq_i} [-], and activity coefficients γ_i and γ_j that are computed using the Debye-Hückel equation that,

$$C_{i} = (Y_{i})^{-1} K_{eq_{i}} \prod_{j=1}^{N_{c}} (Y_{j}C_{j})^{\nu_{ji}}$$
(3)

Correspond to the homogeneous aqueous reactions described in equation [4a] and [4b].

$$\sum_{j=1}^{N_c} v_{ji} A_j \leftrightarrow A_i \tag{4a}$$

$$\sum_{j=1}^{N_c} v_{jm} A_j \leftrightarrow M_m \tag{4b}$$

These reactions are written in their canonical form where A_j denotes the primary species, A_i aqueous secondary species, v_{ji} and v_{jm} [-] stoichiometric coefficient for aqueous and mineral reactions respectively, and M_m denotes minerals.¹ The reaction rate (I_m) [ML⁻³T⁻¹] is given by equation [5] based on the transition state theory where Q_{IAP_m} [M L⁻³T⁻¹] is the ion activity product [6], a_m [L⁻¹] is the new specific surface area that is a function of porosity (φ), k_m is reaction rate constant, \mathcal{P}_m [-] is a prefactor that accounts for the pH dependence and ζ_m [-] is a factor whose value is 0 or 1 depending if $K_{eq_m}Q_{IAP_m} \leq 1$ and $\varphi_m = 0$ or $K_{eq_m}Q_{IAP_m} > 1$ and n_m [-] is fitting parameter.² Water-rock reactions typically follow a non-linear rate law that is sensitive to this fitting parameter ' n_m ' in PFLOTRAN.^{2, 3 4}

$$I_m = -k_m a_m \mathcal{P}_m \left(1 - (K_{eq_m} Q_{IAP_m})^{n_m} \right) \zeta_m$$
(5)

$$Q_m = \prod_{k=1}^{N_c} (\gamma_k^l C_k^l)^{\nu_{km}}$$
(6)

The sign convention used in PFLOTRAN is that if the rate is positive the mineral is precipitating and if it is negative the mineral is dissolving. The change in the material properties like porosity φ and surface area a_m due to mineral precipitation and dissolution reactions are accounted through equations [7] and [8], where, φ_m [-], ϕ_m [-], ϕ_m^0 [-] and a_m^o [L⁻¹] denote porosity of the mth mineral, volume fraction of the mth mineral, initial volume fraction of the mth mineral and initial mineral specific surface area.

$$\varphi = 1 - \sum_{m} \varphi_m \tag{7}$$

$$a_m = a_m^0 (\frac{\phi_m}{\phi_m^0})^{n_m} \tag{8}$$

The porosity, $\varphi = 0.25$, of the Blue Gap/Tachee mine waste sediments were estimated using equation [9] where the bulk density ($\rho_b = 1.65$ gm cm⁻³ [ML⁻³]) and particle density ($\rho_p = 2.25$ gm cm⁻³ [ML⁻³]) were acquired experimentally.

$$\varphi = 1 - \frac{\rho_b}{\rho_p} \tag{9}$$

However, other material properties like tortuosity (tau) and aqueous diffusion coefficient $[M^{2}T^{-1}]$ were assumed to be, tau = 1 and $10^{-9} \text{ m}^{2} \text{ s}^{-1}$ respectively. The assumption for tortuosity was made for simplification purposes, while the assumption for aqueous diffusion coefficient was made based on the diffusion coefficients of H⁺, CO₃²⁻, HCO₃⁻, Ca₂UO₂(CO₃)₃⁰, CaUO₂(CO₃)₃²⁻, UO₂(CO₃)₂²⁻ and Ca²⁺ that were all found to be around $10^{-9} \text{ m}^{2} \text{ s}^{-1}$ at 25°C .^{5, 6} The dirichlet initial and boundary conditions were imposed on the 1D reactive transport of metals over time during reaction with 10 mM HCO₃⁻ and 10 mM CH₃COOH. The initial surface area a_m° (L⁻¹) of U-V

bearing minerals in the mine waste sediments is calculated using equation [10] in cm² cm⁻³. The specific mineral surface area a_m^{o} , [L⁻¹] is defined as

$$a_m^0 = \frac{A_m}{V} = \frac{A_m}{V_m} \frac{V_m}{V} = \frac{a_m}{v_m} \phi_m \tag{10}$$

where $\phi_m = V_m/V$ and V_m are the volume fraction and volume of the mth mineral contained in the REV of volume V, respectively, and $a_m [L^{-1}]$ and $v_m [L^3]$ refer to the area and volume of a single sphere of radius $r_m [L]$,

$$a_m = 4\pi r_m^2 \tag{11}$$

$$v_m = \frac{4}{3}\pi r_m^3 \tag{12}$$

Thus the surface area a_m^0 [L⁻¹] further simplifies to,

$$a_m^0 = \frac{3}{r_m} \phi_m \tag{13}$$

The surface area of the U-V bearing mineral was estimated using equation [13] by considering the average particle size of the mine waste used in batch (<63 μ m) and column (120-355 μ m) based on the assumption that 1) all of the U-phases in the mine waste are U-V bearing minerals and 2) the U-V bearing minerals area spherical in shape. The assumptions were made based on prior knowledge on the Blue Gap/Tachee mine site,⁷ where U-V bearing mineral was identified as the dominant U-phase in the mine waste. The shape of the U-V bearing minerals was assumed to be spherical for simplifying the calculations.

Table S1. Thermodynamic equilibrium constants (at $T = 25^{\circ}C$) for aqueous and solid phase U(VI) used in the reactive transport model (PFLOTRAN).

U(VI) Aqueous and Mineral species	Log <i>K_{eq}</i> Reference		Implementation in the model
Primary species			
$H_2O = OH^- + H^+$	-14.00	Shock et al., 1988 ^{8, 9}	Fixed
U^{4+} + 1.5 H ₂ O + 0.25O ₂ = UO ₂ ²⁺ + 3 H ⁺	-13.2076	Cox et al., 1989 ¹⁰	Fixed
$VO^{2+} + 0.5 H_2O + 0.5O_2 = VO^{2+} + H^+$	-3.8528	Shock et al., 1988 ⁸	Fixed
$H_3AsO_4 + H^+ = H_2AsO_4^-$	2.2492	Shock et al., 1988 ⁸	Fixed
$H_2CO_3 = HCO_3^- + H^+$	6.37	Shock et al., 1988 ⁸	Fixed
$HCO_{3}^{-} = CO_{3}^{2-} + H^{+}$	10.33	Shock et al., 1988 ⁸	Fixed
$HCO_3^- + H^+ = H_2O + CO_2 (aq)$	-6.3447	Shock et al., 1988 ⁸	Fixed
Aqueous complexes			
$Al^{3+} + H_2O = AlOH^{2+} + H^+$	4.971	Pokrovskii et al., 1995 ¹¹	Fixed
$Al^{3+} + 2H_2O = Al(OH)_2^+ + H^+$	10.594 Pokrovskii et al., 1995 ¹¹		Fixed
$Al^{3+} + 3H_2O = Al(OH)_3 (aq) + 3H^+$	16.1577	Pokrovskii et al., 1995 ¹¹	Fixed
$Al^{3+} + 4H_2O = Al(OH)_4^{-} + H^{+}$	22.8833	Pokrovskii et al., 1995 ¹¹	Fixed
$Ca^{2+} + H_2O = CaOH^+ + H^+$	12.85	Baes et al., 1976 ¹²	Fixed
$Ca^{2+} + HCO_3^- = CaCO_3 (aq) + H^+$	7.0017 Johnson. J. W et al. 1992 ¹³		Fixed
$Ca^{2+} + HCO_3^- = CaHCO_3^+$	-1.0467	Johnson. J. W et al. 1992 ¹³	Fixed
$Mg^{2+} + OH^{-} = MgOH^{+}$	-2.210	Johnson. J. W et al. 1992 ¹³	Fixed
$Mg^{2+} + HCO_3^- = MgCO_3(aq) + H^+$	7.3499	Johnson. J. W et al. 1992 ¹³	Fixed
$Mg^{2+} + HCO_3^- = MgHCO_3^+$	-1.0357	Johnson. J. W et al. 1992 ¹³	Fixed

$Mn^{2+} + H_2O = MnOH^+ + H^+$	10.59	Baes et al., 1976 ¹²	Fixed
$Mn^{2+} + 3H_2O = Mn(OH)_3^- + 3H^+$	34.22	Wagman et al., 1982 ¹⁴	Fixed
$Mn^{2+} + 4H_2O = Mn(OH)_4^{2-} + 4H^+$	48.3	Baes et al., 1976 ¹²	Fixed
$Na^+ + H_2O = NaOH (aq) + H^+$	14.79	Johnson. J. W et al. 1992 ¹³	Fixed
$Na^{+} + HCO_{3}^{-} = NaHCO_{3}$ (aq)	-0.1541	Wagman et al., 1982 ¹⁴	Fixed
$K^{+} + H_2O = KOH (aq) + H^{+}$	14.46	Baes et al., 1976 ¹²	Fixed
$H^{+} + SO_{4}^{2-} = HS^{-} + 2O_{2}$	132.52	Shock et al., 1988 ⁸	Fixed
$Fe^{2+} + H^+ + 0.25 O_2 = Fe^{3+} + 0.5 H_2O$	-7.765	Shock et al., 1988 ⁸	Fixed
$\mathrm{Fe}^{2+} + \mathrm{H}_{2}\mathrm{O} = \mathrm{FeOH}^{+} + \mathrm{H}^{+}$	9.5	Baes et al., 1976 ¹²	Fixed
$Fe^{2+} + 2H_2O = Fe(OH)_2(aq) + H^+$	20.60	Baes et al., 1976 ¹²	Fixed
$Fe^{3+} + 2H_2O = Fe(OH)^{2+} + 2H^+$	5.67	Baes et al., 1976 ¹²	Fixed
$Fe^{3+} + 3H_2O = Fe(OH)_3 (aq) + 3H^+$	12	Baes et al., 1976 ¹²	Fixed
$Fe^{2+} + 3H_2O = Fe(OH)_3^- + 3H^+$	31	Baes et al., 1976 ¹²	Fixed
$Fe^{3+} + 4H_2O = Fe(OH)_4^{-} + 4H^{+}$	21.6	Baes et al., 1976 ¹²	Fixed
$Fe^{2+} + 4H_2O = Fe(OH)_4^{2-} + 4H^+$	46	Baes et al., 1976 ¹²	Fixed
$Fe^{2+} + CO_3^{2-} = FeCO_3$ (aq)	5.5988	Turner et al., 1981 ¹⁵	Fixed
$\mathrm{Fe}^{2+} + \mathrm{HCO}_{3}^{-} = \mathrm{FeCO}_{3}^{+} + \mathrm{H}^{+}$	0.6088	Turner et al., 1981 ¹⁵	Fixed
$Fe^{2+} + HCO_3^- = FeHCO_3^+$	-2.72	Mattigod et al., 1979 ¹⁶	Fixed
$\mathrm{Li}^{+} + \mathrm{H}_{2}\mathrm{O} = \mathrm{LiOH} (\mathrm{aq}) + \mathrm{H}^{+}$	13.64	Baes et al., 1976 ¹²	Fixed
$UO_2^{2+} + H^+ = U^{3+} + 0.75O_2 + 0.5H_2O$	62.6291	Grenthe et al., 1992 ¹⁷	Fixed
$UO_2^{2+} + 2H^+ = U^{4+} + H_2O + 0.5O_2$	32.4999	Grenthe et al., 1992 ¹⁷	Fixed
$U^{4+} + 1.5H_2O + 0.25O_2 = UO_2^{2+} + 3H^+$	-13.2076	Grenthe et al., 1992 ¹⁷	Fixed

$UO_2^{2^+} + H_2O = UO_2OH^+ + H^+$	5.2073	Grenthe et al., 1992 ¹⁷	Fixed
$UO_2^{2+} + 2H_2O = UO_2(OH)_2 (aq) + 2H^+$	10.3146	Grenthe et al., 1992 ¹⁷	Fixed
$UO_2^{2+} + 3H_2O = UO_2(OH)_3^{-} + 3H^{+}$	19.2218	Grenthe et al., 1992 ¹⁷	Fixed
$UO_2^{2+} + 4H_2O = UO_2(OH)_4^{2-} + 4H^+$	33.0291	Grenthe et al., 1992 ¹⁷	Fixed
$2UO_2^{2+} + 1H_2O = (UO_2)_2OH^{3+} + H^+$	2.7072	Grenthe et al., 1992 ¹⁷	Fixed
$2UO_2^{2^+} + 2H_2O = (UO_2)_2(OH)_2^{2^+} + 2H^+$	5.6346	Grenthe et al., 1992 ¹⁷	Fixed
$3UO_2^{2+} + 4H_2O = (UO_2)_3(OH)_4^{2+} + 4H^+$	11.929	Grenthe et al., 1992 ¹⁷	Fixed
$3UO_2^{2+} + 5H_2O = (UO_2)_3(OH)_5^+ + 5H^+$	15.5862	Grenthe et al., 1992 ¹⁷	Fixed
$3UO_2^{2+} + 7H_2O = ((UO_2)_3(OH)_7 + 7H^+)$	31.0508	Grenthe et al., 1992 ¹⁷	Fixed
$4UO_2^{2+} + 7H_2O = (UO_2)_4(OH)_7^+ + 7H^+$	21.9508	Grenthe et al., 1992 ¹⁷	Fixed
$UO_2^{2+} + HCO_3^{-} = UO_2CO_3(aq) + H^+$	0.6634	Grenthe et al., 1992 ¹⁷	Fixed
$UO_2^{2^+} + 2HCO_3^- = UO_2(CO_3)_2^{2^-} + 2H^+$	3.7467	Grenthe et al., 1992 ¹⁷	Fixed
$UO_2^{2^+} + 3HCO_3^- = UO_2(CO_3)_3^{4^-} + 3H^+$	9.4302	Grenthe et al., 1992 ¹⁷	Fixed
$2UO_2^{2^+} + 3H_2O + HCO_3^- = (UO_2)_2CO_3(OH)_3^- + 4H^+$	11.2229	Grenthe et al., 1992 ¹⁷	Fixed
$2Ca^{2+} + UO_2^{2+} + 3CO_3^{2-} = Ca_2UO_2(CO_3)_3$ (aq)	-30.04	Dong et al., 2006 ¹⁸	Fixed
$Ca^{2+} + UO_2^{2+} + 3CO_3^{2-} = CaUO_2(CO_3)_3^{2-}$	-27.18	Dong et al., 2006 ¹⁸ , Bernhard et al., 2001 ¹⁹	Fixed
$VO^{2+} + H^+ = V^{3+} + 0.5H_2O + 0.25O_2$	14.9945	Israel et al. ,1976 ²⁰	Fixed
$VO^{2+} + 0.5H_2O + 0.25O_2 = VO_2^+ + H^+$	-3.8528	Shock et al., 1988 ⁸	Fixed
$V^{3+} + 3H_2O + 0.5O_2 = VO_4^{3-} + 6H^+$	9.6002	Wagman et al., 1982 ¹⁴	Fixed
$VO^{2+} + 2H_2O = (VO)_2(OH)_2^{2+} + 2H^+$	6.67	Baes et al., 1976 ¹²	Fixed
$VO^{2+} + 2.5H_2O + 0.25O_2 = H_2VO_4^{-} + 3H^{+}$	3.2394	Johnson. J. W et al. 1992 ¹³	Fixed
$VO^{2+} + 2.5H_2O + 0.25O_2 = HVO_4^{2-} + 4H^+$	11.3024	Johnson. J. W et al. 1992 ¹³	Fixed

$V^{3+} + H_2O = VOH^{2+} + H^+$	2.260 Baes et al., 1976 ¹²		Fixed
$VO^{2+} + H_2O = VOOH^+ + H^+$	5.67 Baes et al., 1976 ¹²		Fixed
$VO_4^{3-} + 2H^+ = VO_3OH^{2-}$	-14.26	Baes et al., 1976 ¹²	Fixed
$V^{3+} + 2H_2O = V(OH)_2^+ + 2H^+$	5.9193	Langmuir et al., 1978 ²¹	Fixed
$2V^{3+} + 2H_2O = V_2(OH)_2^{4+} + 2H^+$	3.80	Baes et al., 1976 ¹²	Fixed
$VO_2^+ + 2H_2O = VO(OH)_3(aq) + H^+$	3.30	Baes et al., 1976 ¹²	Fixed
$H_2AsO_4^- + H^+ = AsH_3(aq) + 2O_2$	149.3941	Sergeyeva et al., 1969 ²²	Fixed
$H_2AsO_4^- = H_2AsO_3^- + 0.5O_2$	29.0857	Shock et al., 1988 ⁸	Fixed
$H_2AsO_4^- = AsO_2^- + 3H_2O + 0.5O_2$	29.0746	Grenthe et al., 1992 ¹⁷	Fixed
$H_2AsO_3^- + 4H^+ = As(OH)_3 (aq)$	-9.2048	Grenthe et al., 1992 ¹⁷	Fixed
Solid Mineral phases			
$Calcite CaCO_3 + H^+ = Ca^{2+} + HCO_3^-$	1.8487	Helgeson et al. 1978 ²³	Fixed
Becquerelite Ca (UO ₂) ₆ O ₄ (OH) ₆ . $8H_2O + 14H^+ = 6UO_2^{2+} + Ca^{2+} + 18H_2O$	29.23	Casas et al., 1997 ²⁴	Fixed
$\begin{array}{c} Quartz\\ SiO_2 = SiO_2 \ (aq) \end{array}$	-3.9993	Helgeson et al. 1978 ²³	Fixed
Boehmite AlO ₂ H + 3H ⁺ = Al ³⁺ + 2H ₂ O	7.5642	Pokrovskii et al., 1995 ¹¹	Fixed
Diaspore AlHO ₂ + $3H^+ = Al^{3+} + 2H_2O$	7.1603	Pokrovskii et al., 1995 ¹¹	Fixed
Gibbsite Al(OH) ₃ + $3H^+ = Al^{3+} + 3H_2O$	7.756	Pokrovskii et al., 1995 ¹¹	Fixed
Hercynite FeAl ₂ O ₄ + 8H ⁺ = Fe ²⁺ + 2Al ³⁺ + 4H ₂ O	28.8484 Robie et al. 1995 ²⁵		Fixed
Spinel Al ₂ MgO ₄ + 8H ⁺ = Mg ²⁺ + 2Al ³⁺ + 4H ₂ O	37.6295	Helgeson et al. 1978 ²³	Fixed

Portlandite $Ca(OH)_2 + 2H^+ = Ca^{2+} + 2H_2O$	22.5552	Robie et al. 1995 ²⁵	Fixed
$ \begin{array}{c} Wustite \\ Fe_{0.947}O + 2H^{+} = 0.1060Fe^{3+} + 0.8410Fe^{2+} + \\ H_{2}O \end{array} $	12.4113	Wagman et al., 1982 ¹⁴	Fixed
$\begin{array}{c} \text{Periclase} \\ \text{MgO} + 2\text{H}^{+} = \text{Mg}^{2+} + \text{H}_{2}\text{O} \end{array}$	21.3354	Helgeson et al. 1978 ²³	Fixed
Brucite Mg(OH) ₂ + 2H ⁺ = Mg ²⁺ + 2H ₂ O	16.2980	Helgeson et al. 1978 ²³	Fixed
Ferrihydrite $Fe_2O_3 + 3H^+ = Fe^{3+} + 3H_2O$	4.896	Stumm and Morgan ²⁶	Fixed
Goethite FeOOH + $3H^+$ = Fe ³⁺ + $2H_2O$	0.5345	Robie et al. 1995 ²⁵	Fixed
Hematite $Fe_2O_3 + 6H^+ = 2Fe^{3+} + 3H_2O$	0.1086	Helgeson et al. 1978 ²³	Fixed
Uraninite $UO_2 + 4H^+ = U^{4+} + 2H_2O$	-4.8372	Cox et al., 1989 ¹⁰	Fixed
Schoepite $UO_3.2H_2O + 2H^+ = UO^{2+} + 3H_2O$	4.8333	Grenthe et al., 1992 ¹⁷	Fixed
Rutherfordine $UO_2CO_3 + H^+ = HCO_3^- + UO_2^{2+}$	-13.9	Meinrath 1993 ²⁷	Fixed
Boltwoodite-Na Na _{.7} K _{.3} (H ₃ O)(UO ₂)SiO ₄ .H ₂ O + 3H ⁺ = 0.3 K ⁺ + 0.7 Na ⁺ + SiO ₂ (aq)	14.5834	Hemingway et al., 1982	Fixed
Metaschoepite $UO_3.2H_2O + 2H^+ = UO2^{2+} + 3H_2O$	-22.29 (circumn eutral), - 5.26 (acidic)	Meinrath 1993 ^{27,} Riba et al., 2005 ²⁸	Fixed
Uranophane $Ca(UO_2)_2(SiO_3)_2(OH)_2 + 6H^+ = Ca^{2+} + 2SiO2$	17.2850	Wagman et al., 1982 ¹⁴	Fixed
$UO_2CO_3 + H^+ = HCO_3^- + UO2^{2+}$	-4.1267	Grenthe et al., 1992 ¹⁷	Fixed
$CaUO_4 + 4H^+ = Ca^{2+} + UO_2^{2+} + 2H_2O$	15.9420	Grenthe et al., 1992 ¹⁷	Fixed
$UO_3.2H_2O + 2H^+ = UO_2^{2+} + 3H_2O$	4.833	Grenthe et al., 1992 ¹⁷	Fixed
Schoepite-dehydrated $0.9UO_3.0.9H_2O + 2H^+ = UO^{2+} + 1.9H_2O$	5.0167	Grenthe et al., 1992 ¹⁷	Fixed

$V_2O_4 + 4H^+ = 2H_2O + 2VO^{2+}$	8.5719 Wagman et al., 1982 ¹⁴		Fixed
$V_3O_5 + 8H^+ = VO^{2+} + 2V^{3+} + 4H_2O$	13.4312	Wagman et al., 1982 ¹⁴	Fixed
$V_4O_7 + 10H^+ = 2V^{3+} + 2VO^{2+} + 5H_2O$	18.7946	Wagman et al., 1982 ¹⁴	Fixed
Carnotite $(K_2(UO_2)_2(VO_4)_2)^* = 2K^+ + 2UO2^{2+}VO4^{3-}$	-56.3811	Langmuir et al., 1978 ²¹	Varied
$Mg_2V_2O_7 + H2O = 2H^+ + 2Mg^{2+} + 2VO_4^{3-}$	-30.9025	Wagman et al., 1982 ¹⁴	Fixed
$MgV_2O_6 + 2H_2O = Mg^{2+} + 2VO_4^{3-} + 4H^+$	-45.8458	Wagman et al., 1982 ¹⁴	Fixed
Tyuyamunite $Ca(UO_2)_2(VO_4)_2 = Ca^{2+} + 2UO_2^{2+} + 2VO_4^{3-}$	-53.3757	Langmuir et al., 1978 ²¹	Fixed
Arsenopyrite FeAsS + $1.5H_2O = 0.5H^+ + 0.5AsH_3$ (aq)	-14.4453	Wagman et al., 1982 ¹⁴	Fixed
$\begin{array}{c} \text{Arsenolite} \\ \text{As}_2\text{O}_3 + 3\text{H}_2\text{O} = 2\text{H}^+ + 2\text{H}_2\text{As}\text{O}_3^- \end{array}$	-19.8365 Robie et al. 1995 ²⁵		Fixed
$(UO_2)_2As_2O_7 + 2H^+ + H_2O = 2H_2AsO_4^- + 2UO_2^{2+}$	7.7066	Grenthe et al., 1992 ¹⁷	Fixed
$(UO_2)_3(AsO_4)_2 + 4H^+ = 2H_2AsO_4^- + 3UO_2^{2+}$	9.3177	Grenthe et al., 1992 ¹⁷	Fixed
$KUO_{2}AsO_{4} + 2H^{+} = H_{2}AsO^{4-} + K^{+} + UO_{2}^{2+}$	-4.1741	Wagman et al., 1982 ¹⁴	Fixed
$LiUO_2AsO_4 + 2H^+ = H_2AsO_4 + Li^+ + UO_2^{2+}$	-0.7862 Wagman et al., 1982 ¹⁴		Fixed
$UO_2(AsO_3)_2 + 2H_2O = UO_2^{2+} + 2H_2AsO_4^{-}$	6.9377 Grenthe et al., 1992 ¹⁷		Fixed
Surface Complexes			
>SOUO2OH	6.448	Bond et al., 2007 ²⁹	Fixed
>SOHUO ₂ CO ₃	2.033	Bond et al., 2007 ²⁹	Fixed

Table S2. Other parameters used in the reactive transport model to simulate the experimental release of U and V from mine waste.

Parameter	Value	Implementation in the model
Reaction rate constant (k_m) of carnotite	-	Varied
Reaction rate constant (k_m) of Rutherfordine $(UO_2CO_3) \pmod{cm^{-2} \sec^{-1}}$	-	Varied
Reaction rate constant (k_m) of metaschoepite $(UO_3.2H_2O) \text{ (mol cm}^{-2} \text{ sec}^{-1})$	-	Varied
Porosity (φ)	0.25	Calculated, Fixed
Tortuosity (tau)	1	Assumed, Fixed
aqueous diffusion coefficient ($m^2 s^{-1}$)	10 ⁻⁹	From Literature, Fixed
Darcy's Flux, q (cm min ⁻¹)	1.496	Calculated, Fixed
Average U-V bearing mineral diameter (μm) Batch experiments (<63 μm)	32	Assumed based on size fractions to calculate average surface area, fixed
Average U-V bearing mineral diameter (μm) Column Experiments (120-355 μm)	240	Assumed based on size fractions to calculate average surface area, fixed

Table S3. Reaction rate constants of metaschoepite and rutherfordine estimated by modelling the reactive transport of U and V during mine waste reaction at circumneutral and acidic conditions.

Mineral phase	Reaction rate constants (k_m) (mol cm ⁻² sec ⁻¹)
Rutherfordine (UO ₂ CO ₃) (Circumneutral Batch)	1x10 ⁻¹⁵
Metaschoepite (UO ₃ .2H ₂ O) (Circumneutral Batch)	2x10 ⁻¹⁶

Table S4. Parameters used to consider the effect of grain size on the reactive transport of U and V during flow through column experiment with 10 mM HCO₃⁻. The surface area of U-V bearing minerals was estimated using equation 13. The effective reaction rate constant ($k_{effective} = k^*a_m^0$) accounts for the effect of grain surface area on the reactive transport of U and V, where k is the reaction rate constant and a_m^0 is the surface are of U-V bearing minerals.

U-V bearing minerals of different grain sizes	Average U-V bearing mineral diameter (μm)	Estimated U-V bearing mineral surface area (a_m^0) $(\text{cm}^2 \text{ cm}^{-3})$	Reaction rate constant (k _m) (mol cm ⁻² sec ⁻¹)	Effective reaction rate constant $(k_{effective} = (k_m * a_m^0)$ (mol cm ⁻³ sec ⁻¹)
U-V mineral	240	62.5	4.8×10^{-13}	3.0x10 ⁻¹¹
U-V mineral 1	200	75	4.8×10^{-13}	3.6x10 ⁻¹¹
U-V mineral 2	280	53.6	4.8×10^{-13}	2.6x10 ⁻¹¹
U-V mineral 3	120	125	4.8×10^{-13}	6.0x10 ⁻¹¹

Table S5. Elemental content of solid samples determined by acid extractions (measured using ICP-OES/MS) of unreacted mine waste and mine waste after sequential reaction with 10 mM HCO₃⁻ and CH₃COOH solution during flow through column experiments.

Samples	U (mg Kg ⁻¹)	V (mg Kg ⁻¹)	As (mg Kg ⁻¹)	Fe (mg Kg ⁻¹)
Unreacted mine waste Acid extractable	1912.12	858.01	4.015	715.79
Reacted mine waste Acid extractable	17.79	62.01	1.17	456.89



Figure S1. Experimental setup of sequential flow through column experiments



Figure S2. Effluent concentrations of U and V obtained after reaction of mine waste (circle) and background soil (squares) with $18M\Omega$ water (pH 5.4), during continuous flow-through column experiments, as a function of pore volumes and time. **A)** U concentration versus pore volume; **B)** V concentration versus pore volume; **C)** U concentration versus time; **D)** V concentration versus time.



Figure S3. Molar correlation between U and V experimental release from mine waste (hollow circles) after reaction with **A)** 10 mM HCO₃⁻.and **B)** 10 mM CH₃COOH during continuous flow through column experiments. The ideal slope should be 1:1 from the reaction $(K_2(UO_2)_2(VO_4)_2)^* = 2K^+ + 2UO2^{2+} + 2VO4^{-3}$. The point in the dashed circle represents an outlier due to equilibration of the column with the influent solution on the early stages of the experiment (first 30 minutes).



Figure S4. Measured and simulated effluent concentrations and reactive transport model (PFLOTRAN) of U and V, from mine waste (circle) and background soil (squares) during continuous flow-through column experiments at pH 7.9 (using 10 mM HCO₃⁻), as a function of time. **A)** U concentration versus time; **B)** V concentration versus time. The curve fitting using PFLOTRAN are presented with dashed lines.



Figure S5. Measured and simulated effluent concentrations and reactive transport model (PFLOTRAN) of U and V from mine waste (circle) as a function of pore volumes after considering the effect of grain size during continuous flow-through column experiments at pH 7.9 (using 10 mM HCO₃⁻). A) U concentrations versus pore volumes B) V concentrations versus pore volumes. The curve fitting using PFLOTRAN are presented with dashed lines.



Figure S6. Microprobe identification of a V-Fe-K phase in the mine waste samples. **A)** Back Scatter Electron Imaging (BSE) of a V-Fe-K phase. **B)** EDS spectrum of the identified V-Fe-K phase.



Figure S7. Synchrotron Micro- X-ray Fluorescence mapping for mine waste samples identifying the co-occurrence of U with other metals. A) Uranium map; B) Vanadium map; C) Selenium map; D) Strontium map; E) Overlay map of U- red, Se – green and V – blue; F) Overlay map of U- red, Sr – green and V – blue. Gradient yellow colors in E and F suggest potential co-occurrence of U and Se, and U and Sr respectively. Similarly cyan gradient colors in E and F suggest co-occurrence of U and V, possibly as a U-V mineral.



Figure S8. Measured and simulated effluent concentrations and reactive transport model (PFLOTRAN) of U and V, from mine waste (circle) and background soil (squares) during continuous flow-through column experiments at pH 3.4 (using 10 mM CH₃COOH) as a function of time. **A)** U concentration versus time; **B)** V concentration versus time. The curved fitting using PFLOTRAN are presented with solid lines.

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