

COUNTING POINTS ON CURVES: THE GENERAL CASE.

JAN TUITMAN

ABSTRACT. We introduce a new algorithm to compute the zeta function of a curve over a finite field. This method extends previous work of ours to all curves for which a lift to characteristic zero is known that satisfies certain conditions. We develop all the necessary bounds, analyse the complexity of the algorithm and provide some examples computed with our implementation.

1. INTRODUCTION

Let \mathbf{F}_q denote the finite field of characteristic p and cardinality $q = p^n$. Moreover, let \mathbf{Q}_p denote the field of p -adic numbers and \mathbf{Q}_q its unique unramified extension of degree n . As usual, let $\sigma \in \text{Gal}(\mathbf{Q}_q/\mathbf{Q}_p)$ denote the unique element that lifts the p -th power Frobenius map on \mathbf{F}_q . Finally, let \mathbf{Z}_q denote the ring of integers of \mathbf{Q}_q , so that $\mathbf{Z}_q/p\mathbf{Z}_q \cong \mathbf{F}_q$. Suppose that X is a smooth proper algebraic curve of genus g over \mathbf{F}_q . Recall that the zeta function of X is defined as

$$Z(X, T) = \exp \left(\sum_{i=1}^{\infty} |X(\mathbf{F}_{q^i})| \frac{T^i}{i} \right).$$

It follows from the Weil conjectures that $Z(X, T)$ is of the form

$$\frac{\chi(T)}{(1-T)(1-qT)},$$

with $\chi(T) \in \mathbf{Z}[T]$ a polynomial of degree $2g$, the inverse roots of which have complex absolute value $q^{\frac{1}{2}}$ and are permuted by the map $t \rightarrow q/t$. Moreover, by the Lefschetz formula for rigid cohomology, we have that

$$\chi(T) = \det \left(1 - T F_p^n | H_{\text{rig}}^1(X) \right),$$

where F_p denotes the p -th power Frobenius map.

In [13], Kedlaya showed how $Z(X, T)$ can be determined efficiently, in the case when X is a hyperelliptic curve and the characteristic p is odd, by explicitly computing the action of F_p on $H_{\text{rig}}^1(X)$. His algorithm was then extended to characteristic 2 [6] and also to superelliptic curves [9], C_{ab} curves [5] and nondegenerate curves [4]. However, for C_{ab} and nondegenerate curves these algorithms have proved a lot less efficient in practice than for hyperelliptic and superelliptic curves. The main reason for this is that the algorithms for $C_{a,b}$ and nondegenerate curves use a more complicated Frobenius lift that does not send x to x^p anymore. Moreover, in the case of nondegenerate curves, the linear algebra that is used to compute in the cohomology is not very efficient and when the curve admits a low degree map to the projective line, as is the case for most nondegenerate curves, this is not fully exploited.

We proposed a very general and practical extension of Kedlaya's algorithm in [20]. Our approach there can be summarised as follows. We start with a finite separable map $x : X \rightarrow \mathbf{P}_{\mathbf{F}_q}^1$ of degree d_x and a rational function $y : X \rightarrow \mathbf{P}_{\mathbf{F}_q}^1$ of degree d_y that generates the function field of X over $\mathbf{F}_q(x)$, such that $Q(x, y) = 0$ where $Q \in \mathbf{F}_q[x, y]$ is irreducible and monic in y . Note that the degrees of Q in x, y are d_y, d_x , respectively. The polynomial Q is the natural input to our algorithm. After removing the ramification locus of x , we can choose a Frobenius lift that sends x to x^p , which we compute by Hensel lifting as in Kedlaya's algorithm. We then compute in the cohomology as in Lauder's fibration method [16] to find the matrix of Frobenius and the zeta function of X .

The algorithm from [20] can be applied to generic, or in other words random, equations Q . However, there are equations to which it cannot be applied including some very interesting examples. For example, when Q is (the reduction at some prime number p of) one of the defining equations computed for modular curves in [19, 22], it turns out that the algorithm can almost never be applied. The reason is that in [20] we assume that Q , or rather its lift \mathcal{Q} to characteristic zero, defines a smooth curve in the affine (x, y) -plane, i.e. that all the singularities of the plane curve defined by \mathcal{Q} are at infinity. We expect that this condition is similar to nondegeneracy in the sense that a generic equation satisfies it, but a generic curve cannot be defined by an equation that satisfies it. Moreover, even when an equation satisfying the assumption does exist it is not clear how to find it.

In this paper we eliminate the assumption that \mathcal{Q} does not have any singularities in the affine (x, y) -plane. As a consequence, our algorithm can now be applied to any curve for which we know a good lift to characteristic zero in the sense of Assumption 1. In particular, for any smooth curve \mathbb{X} defined over the rational numbers, our algorithm can be applied to the reduction of X at p for almost all prime numbers p . The time complexity of the algorithm is $\tilde{O}(pd_x^6 d_y^4 n^3)$ by Theorem 4.12, and the space complexity $\tilde{O}(pd_x^4 d_y^3 n^3)$ by Theorem 4.13, under two rather harmless additional assumptions (numbered 2 and 3 below), as was the case in [20].

Note that the time and space complexities of our algorithm are quasilinear in p and hence not polynomial in the size of the input which is $\log(p)d_x d_y n$. This is also the case for Kedlaya's algorithm and the algorithm from [4]. However, for hyperelliptic curves, the dependence on p of the time and space complexities of Kedlaya's algorithm has been improved to $\tilde{O}(p^{1/2})$ [10] and average polynomial time [11] by Harvey. It is an interesting problem whether these ideas can be used to improve the dependence on p of the complexity of our algorithm as well.

We have updated our Magma [3] implementation. The new code can be found at our webpage¹. Some new examples that our algorithm could not handle before are included at the end of the paper.

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2. LIFTING THE CURVE AND FROBENIUS

Recall that X is a smooth proper algebraic curve of genus g over the finite field \mathbf{F}_q of characteristic p and cardinality $q = p^n$. Let $x : X \rightarrow \mathbf{P}_{\mathbf{F}_q}^1$ be a finite separable map of degree d_x and $y : X \rightarrow \mathbf{P}_{\mathbf{F}_q}^1$ a rational function that generates the function

¹https://perswww.kuleuven.be/jan_tuitman

field of X over $\mathbf{F}_q(x)$, such that $Q(x, y) = 0$ where $Q \in \mathbf{F}_q[x, y]$ is irreducible and monic in y (of degree d_x). The degree of Q in x will be denoted by d_y . Let $\mathcal{Q} \in \mathbf{Z}_q[x, y]$ be a lift of Q that contains the same monomials in its support as Q and is still monic in y . Below in Assumption 1 we will impose an important condition on this lift.

Proposition 2.1. *The ring $\mathcal{A} = \mathbf{Z}_q[x, y]/(\mathcal{Q})$ is a free module of rank d_x over $\mathbf{Z}_q[x]$ and a basis is given by $[1, y, \dots, y^{d_x-1}]$.*

Proof. This follows from the fact that \mathcal{Q} is monic in y . \square

Definition 2.2. *We let $\Delta(x) \in \mathbf{Z}_q[x]$ denote the discriminant of \mathcal{Q} with respect to the variable y , define $r(x) \in \mathbf{Z}_q[x]$ as the squarefree polynomial $r = \Delta/(\gcd(\Delta, \frac{d\Delta}{dx}))$ and let $m \in \mathbf{N}$ be the least positive integer such that there exist a polynomial $g(x) \in \mathbf{Z}_q[x]$ that satisfies $r(x)^m = g(x)\Delta(x)$.*

Remark 2.3. *Note that $\Delta(x) \not\equiv 0 \pmod{p}$ since $x : X \rightarrow \mathbf{P}_{\mathbf{F}_q}^1$ is separable.*

Proposition 2.4. *The element*

$$s(x, y) = \Delta(x) / \frac{\partial \mathcal{Q}}{\partial y}$$

of $\mathbf{Q}_q(x, y)$ is contained in \mathcal{A} .

Proof. For $k \in \mathbf{N}$, we let W_k denote the free $\mathbf{Z}_q[x]$ -module of polynomials in $\mathbf{Z}_q[x, y]$ of degree at most $k-1$ in the variable y . Let Σ be the matrix of the $\mathbf{Z}_q[x]$ -module homomorphism:

$$W_{d_x-1} \oplus W_{d_x} \rightarrow W_{2d_x-1}, \quad (a, b) \mapsto a\mathcal{Q} + b \frac{\partial \mathcal{Q}}{\partial y}, \quad (1)$$

with respect to the bases $[1, y, \dots, y^{d_x-2}]$, $[1, y, \dots, y^{d_x-1}]$ and $[1, y, \dots, y^{2d_x-2}]$. By definition we have $\Delta = \det(\Sigma)$, so that Δ is contained in the image of (1) and $\Delta(x)/\frac{\partial \mathcal{Q}}{\partial y}$ is contained in \mathcal{A} . \square

Definition 2.5. *We denote*

$$\mathcal{S} = \mathbf{Z}_q[x, 1/r], \quad \mathcal{R} = \mathbf{Z}_q[x, 1/r, y]/(\mathcal{Q}),$$

and write $\mathcal{V} = \text{Spec } \mathcal{S}$, $\mathcal{U} = \text{Spec } \mathcal{R}$, so that x defines a finite étale morphism from \mathcal{U} to \mathcal{V} . Finally, we let $U = \mathcal{U} \otimes_{\mathbf{Z}_q} \mathbf{F}_q$, $V = \mathcal{V} \otimes_{\mathbf{Z}_q} \mathbf{F}_q$ denote the special fibres and $\mathbb{U} = \mathcal{U} \otimes_{\mathbf{Z}_q} \mathbf{Q}_q$, $\mathbb{V} = \mathcal{V} \otimes_{\mathbf{Z}_q} \mathbf{Q}_q$ the generic fibres of \mathcal{U} and \mathcal{V} , respectively.

Assumption 1. *We will assume that:*

- (1) *There exists a smooth proper curve \mathcal{X} over \mathbf{Z}_q and a smooth relative divisor $\mathcal{D}_{\mathcal{X}}$ on \mathcal{X} such that $\mathcal{U} = \mathcal{X} \setminus \mathcal{D}_{\mathcal{X}}$.*
- (2) *There exists a smooth relative divisor $\mathcal{D}_{\mathbf{P}^1}$ on $\mathbf{P}_{\mathbf{Z}_q}^1$ such that $\mathcal{V} = \mathbf{P}_{\mathbf{Z}_q}^1 \setminus \mathcal{D}_{\mathbf{P}^1}$.*

We write $\mathbb{X} = \mathcal{X} \otimes_{\mathbf{Z}_q} \mathbf{Q}_q$ for the generic fibre of \mathcal{X} .

Remark 2.6. *A relative divisor \mathcal{D} on a smooth curve over \mathbf{Z}_q is smooth over \mathbf{Z}_q if and only if it is reduced and all of the points in its support are smooth over \mathbf{Z}_q , or equivalently if and only if it reduces modulo p to a reduced divisor D . Hence by Assumption 1, all points on $\mathbf{P}_{\mathbf{Q}_q}^1$ in the support of $\mathcal{D}_{\mathbf{P}^1}$ and all points on \mathbb{X} lying over these points are distinct modulo p .*

Remark 2.7. *The general problem of when there exists a lift \mathcal{Q} of the polynomial Q to characteristic zero such that Assumption 1 is satisfied, and if so how this lift can be constructed, is not easy and will not be answered here. We only make the following observations:*

- (1) *For a generic Q a random lift usually works.*
- (2) *Starting from $\mathcal{Q} \in \mathbf{Z}[x, y]$, it is easy to show that for all but a finite number of prime numbers p , the polynomial \mathcal{Q} is a good lift of its reduction modulo p Q in the sense of Assumption 1.*
- (3) *We will see in Remark 3.14 that if Assumption 1 is satisfied, then the finite map $x : X \rightarrow \mathbf{P}_{\mathbf{F}_q}$ is tamely ramified. Therefore, a lift \mathcal{Q} satisfying Assumption 1 does not always exist.*

Definition 2.8. *We denote the ring of overconvergent functions on \mathcal{U} by*

$$\mathcal{R}^\dagger = \mathbf{Z}_q\langle x, 1/r, y \rangle^\dagger / (\mathcal{Q}).$$

Note that \mathcal{R}^\dagger is a free module of rank d_x over $\mathcal{S}^\dagger = \mathbf{Z}_q\langle x, 1/r \rangle^\dagger$ and that a basis is given by $[y^0, \dots, y^{d_x-1}]$. A Frobenius lift $F_p : \mathcal{R}^\dagger \rightarrow \mathcal{R}^\dagger$ is defined as a σ -semilinear ring homomorphism that reduces modulo p to the p -th power Frobenius map.

Theorem 2.9. *There exists a Frobenius lift $F_p : \mathcal{R}^\dagger \rightarrow \mathcal{R}^\dagger$ for which $F_p(x) = x^p$.*

Proof. Let notation be as in Definition 2.2 and Proposition 2.4. Define sequences $(\alpha_i)_{i \geq 0}$, $(\beta_i)_{i \geq 0}$, with $\alpha_i \in \mathcal{S}^\dagger$ and $\beta_i \in \mathcal{R}^\dagger$, by the following recursion:

$$\begin{aligned} \alpha_0 &= \frac{1}{r^p}, \\ \beta_0 &= y^p, \\ \alpha_{i+1} &= \alpha_i(2 - \alpha_i r^\sigma(x^p)) \pmod{p^{2^{i+1}}}, \\ \beta_{i+1} &= \beta_i - \mathcal{Q}^\sigma(x^p, \beta_i) s^\sigma(x^p, \beta_i) g^\sigma(x^p) \alpha_i^m \pmod{p^{2^{i+1}}}. \end{aligned}$$

Then one easily checks that the σ -semilinear ringhomomorphism $F_p : \mathcal{R}^\dagger \rightarrow \mathcal{R}^\dagger$ defined by

$$F_p(x) = x^p, \quad F_p(1/r) = \lim_{i \rightarrow \infty} \alpha_i, \quad F_p(y) = \lim_{i \rightarrow \infty} \beta_i,$$

is a Frobenius lift. □

Assumption 2. *We will assume that matrices $W^0 \in Gl_{d_x}(\mathbf{Z}_q[x, 1/r])$ and $W^\infty \in Gl_{d_x}(\mathbf{Z}_q[x, 1/x, 1/r])$ are known such that, if we denote $b_j^0 = \sum_{i=0}^{d_x-1} W_{i+1, j+1}^0 y^i$ and $b_j^\infty = \sum_{i=0}^{d_x-1} W_{i+1, j+1}^\infty y^i$ for all $0 \leq j \leq d_x - 1$, then:*

- (1) $[b_0^0, \dots, b_{d_x-1}^0]$ is an integral basis for $\mathbf{Q}_q(x, y)$ over $\mathbf{Q}_q[x]$,
- (2) $[b_0^\infty, \dots, b_{d_x-1}^\infty]$ is an integral basis for $\mathbf{Q}_q(x, y)$ over $\mathbf{Q}_q[1/x]$.

Let $W \in Gl_{d_x}(\mathbf{Z}_q[x, 1/x])$ be the matrix defined by $W = (W^0)^{-1} W^\infty$.

Remark 2.10. *There are good algorithms available to compute integral bases in function fields. In our implementation, we use the one that comes with Magma. In practice, the time required to compute the integral bases always seems to be negligible compared to the other steps in our algorithm.*

Proposition 2.11. *Let $G^0 \in M_{d_x \times d_x}(\mathbf{Z}_q[x, 1/r])$ and $G^\infty \in M_{d_x \times d_x}(\mathbf{Z}_q[x, 1/x, 1/r])$ denote the matrices such that*

$$db_j^0 = \sum_{i=0}^{d_x-1} G_{i+1,j+1}^0 b_i^0 dx, \quad db_j^\infty = \sum_{i=0}^{d_x-1} G_{i+1,j+1}^\infty b_i^\infty dx,$$

for all $0 \leq j \leq d_x - 1$. Let $x_0 \neq \infty$ be a geometric point of $\mathbf{P}^1(\bar{\mathbf{Q}}_q)$. Then the matrix $G^0 dx$ has at most a simple pole at x_0 . Similarly, the matrix $G^\infty dx$ has at most a simple pole at $x = \infty$.

Proof. Note that $\text{ord}_P(dx/(x-x_0)) = -1$ at every $P \in \mathcal{X} \setminus \mathcal{U}$ lying over x_0 . At every such P and for all $0 \leq i \leq d_x - 1$ we clearly have $\text{ord}_P(db_i^0) \geq 0$, so that $\text{ord}_P((x-x_0)db_i^0) - \text{ord}_P(dx) \geq 1$. Since $[b_0^0, \dots, b_{d_x-1}^0]$ is an integral basis for $\mathbf{Q}_q(x, y)$ over $\mathbf{Q}_q[x]$, we conclude that $(x-x_0)G^0$ does not have a pole at x_0 , so that $G^0 dx$ has at most a simple pole there. To show that $G^\infty dx$ has at most a simple pole at $x = \infty$, replace the geometric point x_0 by ∞ , the basis $[b_0^0, \dots, b_{d_x-1}^0]$ by $[b_0^\infty, \dots, b_{d_x-1}^\infty]$, and the local parameter $(x-x_0)$ by $t = 1/x$ in the argument. \square

Remark 2.12. *In particular, we have that $rG^0 \in M_{d_x \times d_x}(\mathbf{Z}_q[x])$.*

Definition 2.13. *Let $x_0 \in \mathbf{P}^1(\bar{\mathbf{Q}}_q) \setminus \infty$ be a geometric point. The exponents of $G^0 dx$ at x_0 are defined as the eigenvalues of the residue matrix $G_{-1}^{x_0} = (x-x_0)G^0|_{x=x_0}$. Moreover, the exponents of $G^\infty dx$ at $x = \infty$ are defined as its exponents at $t = 0$, after substituting $x = 1/t$.*

Proposition 2.14. *The exponents of $G^0 dx$ at any geometric point $x_0 \in \mathbf{P}^1(\bar{\mathbf{Q}}_q) \setminus \infty$ and the exponents of $G^\infty dx$ at $x = \infty$ are elements of $\mathbf{Q} \cap \mathbf{Z}_p$ and are contained in the interval $[0, 1)$.*

Proof. Let $\lambda \in \bar{\mathbf{Q}}_q$ denote an exponent of $G^0 dx$ at $x_0 \neq \infty$. Then there exists $f = \sum_{i=0}^{d_x-1} a_i b_i^0$ with $a_0, \dots, a_{d_x-1} \in \bar{\mathbf{Q}}_q$ such that

$$df = \left(\frac{\lambda f}{x-x_0} + g \right) dx \quad (2)$$

as 1-forms on $\mathbb{U} \otimes \bar{\mathbf{Q}}_q$, where $g \in \mathcal{O}(\mathbb{U} \otimes \bar{\mathbf{Q}}_q)$ satisfies $\text{ord}_P(g) \geq 0$ at all points $P \in x^{-1}(x_0)$. Note that for at least one $P \in x^{-1}(x_0)$ we have $\text{ord}_P(f) < \text{ord}_P(x-x_0)$, since otherwise $f/(x-x_0)$ would be integral over $\mathbf{Q}_q[x]$, contradicting Assumption 2. For such a P , dividing by f in (2) and taking residues, we obtain

$$\text{ord}_P(f) = \lambda \text{ord}_P(x-x_0) = \lambda e_P.$$

Since $0 \leq \text{ord}_P(f) < \text{ord}_P(x-x_0)$, we see that $\lambda \in \mathbf{Q} \cap [0, 1)$. By Assumption 1, elements of \mathcal{S} have p -adically integral Laurent series expansions at x_0 , so that $G_{-1}^{x_0} \in M_{d_x \times d_x}(\mathbf{Z}_q)$. Since p -adically integral matrices have p -adically integral eigenvalues, we conclude that $\lambda \in \mathbf{Z}_p$. To obtain the same result for the exponents of $G^\infty dx$ at $x = \infty$, replace the geometric point x_0 by ∞ , the basis $[b_0^0, \dots, b_{d_x-1}^0]$ by $[b_0^\infty, \dots, b_{d_x-1}^\infty]$, and the local parameter $(x-x_0)$ by $t = 1/x$ in the argument. \square

Definition 2.15. *For a geometric point $x_0 \in \mathbf{P}^1(\bar{\mathbf{Q}}_q)$, we let $\text{ord}_{x_0}(\cdot)$ denote the discrete valuation on $\bar{\mathbf{Q}}_q(x)$ corresponding to x_0 . Moreover, we define*

$$\text{ord}_{\neq \infty}(\cdot) = \min_{x_0 \in \mathbf{P}^1(\bar{\mathbf{Q}}_q) \setminus \infty} \{\text{ord}_{x_0}(\cdot)\}.$$

We extend these definitions to matrices over $\bar{\mathbf{Q}}_q(x)$ by taking the minimum over their entries.

Proposition 2.16. *Let $N \in \mathbf{N}$ be a positive integer.*

- (1) *The element $F_p(1/r)$ of \mathcal{S}^\dagger is congruent modulo p^N to*

$$\sum_{i=p}^{pN} \frac{\rho_i(x)}{r^i},$$

where $\rho_i \in \mathbf{Z}_q[x]$ satisfies $\deg(\rho_i) < \deg(r)$ for all $p \leq i \leq pN$.

- (2) *For all $0 \leq i \leq d_x - 1$, the element $F_p(y^i)$ of \mathcal{R}^\dagger is congruent modulo p^N to $\sum_{j=0}^{d_x-1} \phi_{i,j}(x)y^j$, where*

$$\phi_{i,j} = \sum_{k=0}^{p(N-1) - \text{ord}_{\neq \infty}(W^0) - p \text{ord}_{\neq \infty}((W^0)^{-1})} \frac{\phi_{i,j,k}(x)}{r^k}$$

for all $0 \leq j \leq d_x - 1$ and $\phi_{i,j,k} \in \mathbf{Z}_q[x]$ satisfies

$$\begin{aligned} \deg(\phi_{i,j,0}) &\leq -\text{ord}_{\infty}(W^{\infty}) - p \text{ord}_{\infty}((W^{\infty})^{-1}), \\ \deg(\phi_{i,j,k}) &< \deg(r), \end{aligned}$$

for all $0 \leq j \leq d_x - 1$ and $1 \leq k \leq p(N-1) - \text{ord}_{\neq \infty}(W^0) - p \text{ord}_{\neq \infty}((W^0)^{-1})$.

- (3) *For all $0 \leq i \leq d_x - 1$, the element $F_p(b_i^0/r)$ of \mathcal{R}^\dagger is congruent modulo p^N to $\sum_{j=0}^{d_x-1} \psi_{i,j}(x)(b_j^0/r)$, where*

$$\psi_{i,j} = \sum_{k=0}^{pN-1} \frac{\psi_{i,j,k}(x)}{r^k}$$

for all $0 \leq j \leq d_x - 1$ and $\psi_{i,j,k} \in \mathbf{Z}_q[x]$ satisfies

$$\begin{aligned} \deg(\psi_{i,j,0}) &\leq -\text{ord}_{\infty}(W) - p \text{ord}_{\infty}(W^{-1}) - (p-1) \deg(r), \\ \deg(\psi_{i,j,k}) &< \deg(r), \end{aligned}$$

for all $0 \leq j \leq d_x - 1$ and $1 \leq k \leq pN - 1$.

Proof.

- (1) Since $r^\sigma(x^p) \equiv r^p \pmod{p}$, this follows from

$$F_p\left(\frac{1}{r}\right) = \frac{1}{r^\sigma(x^p)} = \frac{1}{r^p} \left(1 - \frac{r^p - r^\sigma(x^p)}{r^p}\right)^{-1} = \frac{1}{r^p} \sum_{i=0}^{\infty} \left(\frac{r^p - r^\sigma(x^p)}{r^p}\right)^i.$$

- (2) The matrix $\Phi = (\phi_{i,j}) \in M_{d_x \times d_x}(\mathcal{S}^\dagger)$ defines a p -th power Frobenius structure on the (higher) direct image $\mathbf{R}^0 x_*(\mathcal{O}_{\mathbb{U}})$ with respect to the basis $[y^0, \dots, y^{d_x-1}]$. From Proposition 2.11 we know that the matrix of the Gauss–Manin connection ∇ on $\mathbf{R}^0 x_*(\mathcal{O}_{\mathbb{U}})$ has at most a simple pole at any $x_0 \neq \infty \in \mathbf{P}^1(\bar{\mathbf{Q}}_q)$ with respect to the basis $[b_0^0, \dots, b_{d_x-1}^0]$ and at most a simple pole at $x = \infty$ with respect to the basis $[b_0^\infty, \dots, b_{d_x-1}^\infty]$. Moreover, by Proposition 2.14, at every $x_0 \in \mathbf{P}^1(\mathbf{Q}_q)$ the exponents of ∇ with respect to these bases are elements of $\mathbf{Q} \cap \mathbf{Z}_p$ that are contained in the interval $[0, 1)$. Finally, from the definition of Φ it follows that $\text{ord}_p(\Phi) \geq 0$ and by Poincaré duality we find that $\text{ord}_p(\Phi^{-1}) \geq 0$ as well. The result is now a consequence of [15, Corollary 2.6].

- (3) The matrix $\Psi = (\psi_{i,j}) \in M_{d_x \times d_x}(\mathcal{S}^\dagger)$ defines a p -th power Frobenius structure on the (higher) direct image $\mathbf{R}^0 x_*(\mathcal{O}_{\mathbb{U}})$ with respect to the basis $[b_0^0/r, \dots, b_{d_x-1}^0/r]$. With respect to this basis, the matrix of the Gauss–Manin connection ∇ on $\mathbf{R}^0 x_*(\mathcal{O}_{\mathbb{U}})$ has at most a simple pole at every $x_0 \neq \infty \in \mathbf{P}^1(\bar{\mathbf{Q}}_q)$ and the exponents of ∇ are elements of $\mathbf{Q} \cap \mathbf{Z}_p$ that are contained in the interval $[-1, 0)$. We still have that $\text{ord}_p(\Psi), \text{ord}_p(\Psi^{-1}) \geq 0$. The result is now again a consequence of [15, Corollary 2.6]. \square

3. COMPUTING (IN) THE COHOMOLOGY

Definition 3.1. *The rigid cohomology of U in degree 1 can be defined as*

$$H_{rig}^1(U) = \text{coker}(d : \mathcal{R}^\dagger \rightarrow \Omega^1(\mathbb{U}) \otimes \mathcal{R}^\dagger).$$

Theorem 3.2.

$$H_{rig}^1(U) \cong H_{dR}^1(\mathbb{U})$$

Proof. This follows as a special case from the comparison theorem between rigid and de Rham cohomology of Baldassarri and Chiarellotto [1], since by Assumption 1 $\mathcal{D}_{\mathcal{X}}$ is smooth over \mathbf{Z}_q . \square

We can effectively reduce any 1-form to one of low pole order using linear algebra following work of Lauder [16]. The procedure consists of two parts, reducing the pole order at the points not lying over $x = \infty$ and at those lying over $x = \infty$, respectively. From now on we let r' denote the polynomial $\frac{dx}{dx}$. We start with the points not lying over $x = \infty$.

Proposition 3.3. *For all $\ell \in \mathbf{N}$ and every vector $w \in \mathbf{Q}_q[x]^{\oplus d_x}$, there exist vectors $u, v \in \mathbf{Q}_q[x]^{\oplus d_x}$ with $\deg(v) < \deg(r)$, such that*

$$\frac{\sum_{i=0}^{d_x-1} w_i b_i^0}{r^\ell} \frac{dx}{r} = d \left(\frac{\sum_{i=0}^{d_x-1} v_i b_i^0}{r^\ell} \right) + \frac{\sum_{i=0}^{d_x-1} u_i b_i^0}{r^{\ell-1}} \frac{dx}{r}.$$

Proof. Recall from Remark 2.12 that $rG^0 \in M_{d_x \times d_x}(\mathbf{Z}_q[x])$. Note that since r is separable, r' is invertible in the ring $\mathbf{Q}_q[x]/(r)$. One checks that v has to satisfy the $d_x \times d_x$ linear system

$$\left(\frac{rG^0}{r'} - \ell I \right) v \equiv \frac{w}{r'} \pmod{r}$$

over $\mathbf{Q}_q[x]/(r)$. However, since $\ell \geq 1$ is not an exponent of $G^0 dx$ by Proposition 2.14, we have that $\det(\ell I - rG^0/r')$ is invertible in $\mathbf{Q}_q[x]/(r)$, so that this system has a unique solution v . We take

$$u = \frac{w - (rG^0 - \ell r' I) v}{r} - \frac{dv}{dx}. \quad \square$$

We now move on to the points lying over $x = \infty$.

Proposition 3.4. *For every vector $w \in \mathbf{Q}_q[x, 1/x]^{\oplus d_x}$ with*

$$\text{ord}_\infty(w) \leq -\deg(r),$$

there exist vectors $u, v \in \mathbf{Q}_q[x, 1/x]^{\oplus d_x}$ with $\text{ord}_\infty(u) > \text{ord}_\infty(w)$ such that

$$\left(\sum_{i=0}^{d_x-1} w_i b_i^\infty \right) \frac{dx}{r} = d \left(\sum_{i=0}^{d_x-1} v_i b_i^\infty \right) + \left(\sum_{i=0}^{d_x-1} u_i b_i^\infty \right) \frac{dx}{r}.$$

Proof. We still denote $t = 1/x$. We can expand

$$G^\infty dx = \left(\frac{G_{-1}^\infty}{t} + G_0^\infty + \dots \right) dt,$$

where $G_i^\infty \in M_{d_x \times d_x}(\mathbf{Q}_q)$ for all $i \geq -1$. Putting $m = -\text{ord}_\infty(w) - \deg(r) + 1$, we can also write

$$w \frac{dx}{r} = \sum_{j=-(m+1)}^{\infty} \bar{w}_j t^j dt,$$

where $\bar{w}_j \in \mathbf{Q}_q^{\oplus d_x}$ for all $j \geq -(m+1)$. Note that $m \geq 1$. By Proposition 2.14, we have that $\det(mI - G_{-1}^\infty)$ is nonzero, so that the linear system

$$(G_{-1}^\infty - mI)\bar{v} = \bar{w}_{-(m+1)}$$

has a unique solution $\bar{v} \in \mathbf{Q}_q^{\oplus d_x}$. We take

$$v = \bar{v}x^m, \quad u = w - r \left(G^\infty v + \frac{dv}{dx} \right). \quad \square$$

Remark 3.5. Note that when $\text{ord}_\infty(w) \leq \text{ord}_0(W) - \deg(r) + 1$, we have that $\text{ord}_0(v) \geq -\text{ord}_0(W)$, so that the function $\sum_{i=0}^{d_x-1} v_i b_i^\infty$ only has poles at points lying over $x = \infty$.

Next we give an explicit description of the cohomology space $H_{\text{rig}}^1(U)$.

Theorem 3.6. Define the following \mathbf{Q}_q -vector spaces:

$$\begin{aligned} E_0 &= \left\{ \left(\sum_{i=0}^{d_x-1} u_i(x) b_i^0 \right) \frac{dx}{r} : u \in \mathbf{Q}_q[x]^{\oplus d_x} \right\}, \\ E_\infty &= \left\{ \left(\sum_{i=0}^{d_x-1} u_i(x, 1/x) b_i^\infty \right) \frac{dx}{r} : u \in \mathbf{Q}_q[x, 1/x]^{\oplus d_x}, \text{ord}_\infty(u) > \text{ord}_0(W) - \deg(r) + 1 \right\}, \\ B_0 &= \left\{ \sum_{i=0}^{d_x-1} v_i(x) b_i^0 : v \in \mathbf{Q}_q[x]^{\oplus d_x} \right\}, \\ B_\infty &= \left\{ \sum_{i=0}^{d_x-1} v_i(x, 1/x) b_i^\infty : v \in \mathbf{Q}_q[x, 1/x]^{\oplus d_x}, \text{ord}_\infty(v) > \text{ord}_0(W) \right\}. \end{aligned}$$

Then $E_0 \cap E_\infty$ and $d(B_0 \cap B_\infty)$ are finite dimensional \mathbf{Q}_q -vector spaces and

$$H_{\text{rig}}^1(U) \cong (E_0 \cap E_\infty) / d(B_0 \cap B_\infty).$$

Proof. First, note that elements of E_0, B_0 have bounded poles everywhere but at the points lying over $x = \infty$ and elements of E_∞, B_∞ everywhere but at the points lying over $x = 0$. So elements of $E_0 \cap E_\infty$ and $d(B_0 \cap B_\infty)$ have bounded poles everywhere on \mathbb{X} . Hence these vector spaces are contained in the space of global sections of some line bundle on \mathbb{X} and are therefore finite dimensional.

Next, we show that every class in $H_{\text{rig}}^1(U)$ can be represented by a 1-form in $E_0 \cap E_\infty$. Note that by Theorem 3.2 we can restrict to classes in $H_{\text{dR}}^1(\mathbb{U})$. Now every such class can be represented by a 1-form in E_0 by (repeatedly) applying Proposition 3.3. Then we change basis from $[b_0^0, \dots, b_{d_x-1}^0]$ to $[b_0^\infty, \dots, b_{d_x-1}^\infty]$ by the matrix W . Observe that this change of basis might introduce a pole at $x = 0$. Now our cohomology class can be represented by 1-form in $E_0 \cap E_\infty$ by (repeatedly) applying Proposition 3.4 and Remark 3.5.

Finally, we have to prove that if a 1-form $\omega \in E_0 \cap E_\infty$ is exact, then it lies in $d(B_0 \cap B_\infty)$. So let $\omega \in E_0 \cap E_\infty$ denote such an exact 1-form. From the definitions of E_0 and $[b_0^0, \dots, b_{d_x-1}^0]$ it follows that $\text{ord}_P(\omega) \geq -1$ at all points P not lying over $x = \infty$ and from the definitions of E_∞ and $[b_0^\infty, \dots, b_{d_x-1}^\infty]$ it follows that $\text{ord}_P(\omega) \geq (\text{ord}_0(W) + 1)e_P - 1$ at all points P lying over $x = \infty$. Note that the exterior derivative lowers the order by at most 1. So if $\omega = df$ for some $f \in \mathcal{O}(\mathbb{U})$, then $\text{ord}_P(f) \geq 0$ at all points P not lying over $x = \infty$ and $\text{ord}_P(f) \geq (\text{ord}_0(W) + 1)e_P$ at all points P lying over $x = \infty$. Using the definitions of $[b_0^0, \dots, b_{d_x-1}^0]$ and $[b_0^\infty, \dots, b_{d_x-1}^\infty]$ again, it follows that f is an element of $B_0 \cap B_\infty$. \square

Note that by the proof of Theorem 3.6, we can effectively reduce any 1-form to one in $E_0 \cap E_\infty$ with the same cohomology class. However, the reduction procedure will introduce p -adic denominators and therefore suffer from loss of p -adic precision. In the following two propositions we bound these denominators. Our bounds and their proofs generalise the ones from [13].

Proposition 3.7. *Let $\omega \in \Omega^1(\mathcal{U})$ be of the form*

$$\omega = \frac{\sum_{i=0}^{d_x-1} w_i b_i^0 dx}{r^\ell},$$

where $\ell \in \mathbb{N}$ and $w \in \mathbf{Z}_q[x]^{\oplus d_x}$ satisfies $\deg(w) < \deg(r)$. We define

$$e_0 = \max\{e_P \mid P \in \mathcal{X} \setminus \mathcal{U}, x(P) \neq \infty\}.$$

If we represent the class of ω in $H_{\text{rig}}^1(U)$ by

$$\left(\sum_{i=0}^{d_x-1} u_i b_i^0 \right) \frac{dx}{r},$$

with $u \in \mathbf{Q}_q[x]^{\oplus d_x}$ as in the proof of Theorem 3.6, then

$$p^{\lfloor \log_p(\ell e_0) \rfloor} u \in \mathbf{Z}_q[x]^{\oplus d_x}.$$

Proof. We have

$$\omega = df + \left(\sum_{i=0}^{d_x-1} u_i b_i^0 \right) \frac{dx}{r}$$

with $f = \sum_{j=1}^\ell (\sum_{i=0}^{d_x-1} (v_j)_i b_i^0) / r^j$, where $v_j \in \mathbf{Q}_q[x]^{\oplus d_x}$ satisfies $\deg(f_j) < \deg(r)$ for all $1 \leq j \leq \ell$. Note that it is sufficient to show that $p^{\lfloor \log_p(\ell e_0) \rfloor} f \in \mathcal{R}$. By Assumption 1, we have that

$$\mathcal{O}(\mathcal{X} - x^{-1}(\infty)) / (r)^k \cong \prod_{P \in \mathcal{X} \setminus \mathcal{U}, x(P) \neq \infty} \mathcal{O}_{\mathcal{X}, P} / (z_P^{e_P})^k,$$

for all $k \in \mathbf{N}$. Moreover, by definition $[b_0^0, \dots, b_{d_x-1}^0]$ is a basis for $\mathcal{O}(\mathbb{X} - x^{-1}(\infty))$ over $\mathbf{Q}_q[x]$. To show that $p^{\lfloor \log_p(\ell e_0) \rfloor} f$ is integral, it is therefore enough to show that for every $P \in \mathcal{X} \setminus \mathcal{U}$ with $x(P) \neq 0$, the Laurent series expansion

$$a_{-\ell e_P} z_P^{-\ell e_P} + \dots + a_{-e_P-1} z_P^{-e_P-1} + \mathcal{O}(z_P^{-e_P})$$

of $p^{\lfloor \log_p(\ell e_0) \rfloor} f$ is integral. However, the differential df has a pole of order at most $\ell e_P + 1$ at P , and its Laurent series expansion

$$\left(b_{-\ell e_P-1} z_P^{-\ell e_P-1} + \dots + b_{-e_P-2} z_P^{-e_P-2} + \mathcal{O}(z_P^{-e_P-1}) \right) dz_P$$

is integral since ω is integral. The worst denominator we get by integrating this series is therefore $p^{\lfloor \log_p(\ell e_0) \rfloor}$ and the result follows. \square

Proposition 3.8. *Let $\omega \in \Omega^1(\mathcal{U})$ be of the form*

$$\omega = \left(\sum_{i=0}^{d_x-1} w_i(x, x^{-1}) b_i^\infty \right) \frac{dx}{r},$$

where $w \in \mathbf{Z}_q[x, x^{-1}]^{\oplus d_x}$ satisfies $\text{ord}_\infty(w) \leq \text{ord}_0(W^\infty) - \deg(r) + 1$. We write $m = -\text{ord}_\infty(w) - \deg(r) + 1$ and define

$$e_\infty = \max\{e_P | P \in \mathcal{X} \setminus \mathcal{U}, x(P) = \infty\}.$$

If we represent the class of ω in $H_{\text{rig}}^1(\mathcal{U})$ by

$$\left(\sum_{i=0}^{d_x-1} u_i b_i^\infty \right) \frac{dx}{r},$$

with $u \in \mathbf{Q}_q[x, x^{-1}]^{\oplus d_x}$ such that $\text{ord}_\infty(u) > \text{ord}_0(W^\infty) - \deg(r) + 1$ as in the proof of Theorem 3.6, then

$$p^{\lfloor \log_p(m e_\infty) \rfloor} u \in \mathbf{Z}_q[x, x^{-1}]^{\oplus d_x}.$$

Proof. We have

$$\omega = df + \left(\sum_{i=0}^{d_x-1} u_i b_i^\infty \right) \frac{dx}{r}$$

with $f = \sum_{j=-\text{ord}_0(W^\infty)}^m (\sum_{i=0}^{d_x-1} (v_j)_i b_i^\infty) x^j$, where $v_j \in \mathbf{Q}_q^{\oplus d_x}$ for all j such that $-\text{ord}_0(W^\infty) \leq j \leq m$. Note that it is sufficient to show that $p^{\lfloor \log_p(\ell e_\infty) \rfloor} f \in \mathcal{R}$. By Assumption 1, we have that

$$\mathcal{O}(\mathcal{X} - x^{-1}(0))/(t)^k \cong \prod_{P \in \mathcal{X} \setminus \mathcal{U}, x(P) = \infty} \mathcal{O}_{\mathcal{X}, P} / (z_P^{e_P})^k, \quad (3)$$

for all $k \in \mathbf{N}$. Moreover, by definition $[b_0^\infty, \dots, b_{d_x-1}^\infty]$ is a basis for $\mathcal{O}(\mathbb{X} - x^{-1}(0))$ over $\mathbf{Q}_q[x^{-1}]$. To show that $p^{\lfloor \log_p(\ell e_\infty) \rfloor} f$ is integral, it is therefore enough to show that for every $P \in \mathcal{X} \setminus \mathcal{U}$ with $x(P) = \infty$, the Laurent series expansion

$$a_{-m e_P} z_P^{-m e_P} + \dots + a_{(\text{ord}_0(W^\infty)+1)e_P-1} z_P^{(\text{ord}_0(W^\infty)+1)e_P-1} + \mathcal{O}(z_P^{(\text{ord}_0(W^\infty)+1)e_P})$$

of $p^{\lfloor \log_p(\ell e_\infty) \rfloor} f$ is integral. However, the differential df has a pole of order at most $m e_P + 1$ at P , and its Laurent series expansion

$$\left(b_{-m e_P-1} z_P^{-m e_P-1} + \dots + b_{(\text{ord}_0(W^\infty)+1)e_P} z_P^{(\text{ord}_0(W^\infty)+1)e_P} + \mathcal{O}(z_P^{(\text{ord}_0(W^\infty)+1)e_P-1}) \right) dz_P$$

is integral since ω is integral. The worst denominator we get by integrating this series is therefore $p^{\lfloor \log_p(me_\infty) \rfloor}$ and the result follows. \square

Remark 3.9. Note that Propositions 3.3, 3.4, 3.7 and 3.8 can be used to give an alternative effective proof of Theorem 3.2.

Recall that in Theorem 3.6 the computation of a basis for $H_{\text{rig}}^1(U)$ was reduced to a finite dimensional linear algebra problem. However, the dimension of $H_{\text{rig}}^1(U)$ is generally much higher than the dimension of $H_{\text{rig}}^1(X)$, so that we would like to compute a basis for this last space. For this we will need to compute the kernel of a cohomological residue map.

Definition 3.10. For a 1-form $\omega \in \Omega^1(\mathcal{U})$ and a point $P \in \mathcal{X} \setminus \mathcal{U}$, we let

$$\text{res}_P(\omega) \in \mathcal{O}_{\mathcal{X},P}/(z_P)$$

denote the coefficient a_{-1} in the Laurent series expansion

$$\omega = (a_{-k}z_P^k + \dots + a_{-1}z_P^{-1} + \dots)dz_P.$$

Moreover, we denote

$$\text{res}_0 = \bigoplus_{P \in \mathcal{X} \setminus \mathcal{U}: x(P) \neq \infty} \text{res}_P, \quad \text{res}_\infty = \bigoplus_{P \in \mathcal{X} \setminus \mathcal{U}: x(P) = \infty} \text{res}_P.$$

Theorem 3.11. We have an exact sequence

$$0 \longrightarrow H_{\text{rig}}^1(X) \longrightarrow H_{\text{rig}}^1(U) \xrightarrow{(\text{res}_0 \oplus \text{res}_\infty) \otimes \mathbf{Q}_q} \bigoplus_{P \in \mathcal{X} \setminus \mathcal{U}} \mathcal{O}_{\mathcal{X},P}/(z_P) \otimes \mathbf{Q}_q.$$

Proof. This is well known. \square

The kernels of res_0 and res_∞ can be computed without having to compute the Laurent series expansions at all $P \in \mathcal{X} \setminus \mathcal{U}$ using the following two propositions. We start with the residues at the points lying over $x = \infty$.

Proposition 3.12. Let $\omega \in \Omega^1(\mathbb{U})$ be a 1-form of the form

$$\omega = \left(\sum_{i=0}^{d_x-1} u_i(x, x^{-1}) b_i^\infty \right) \frac{dx}{r},$$

where $u \in \mathbf{Q}_q[x, x^{-1}]^{\oplus d_x}$ satisfies $\text{ord}_\infty(u) > -\deg(r)$, and let a vector $v \in \mathbf{Q}_q^{\oplus d_x}$ be defined by $v = (x^{1-\deg(r)}u)|_{x=\infty}$. Moreover, let the residue matrix $G_{-1}^\infty \in M_{d_x \times d_x}(\mathbf{Q}_q)$ be defined as in the proof of Proposition 3.4, and let $\mathcal{E}_\lambda^\infty$ denote the (generalised) eigenspace of G_{-1}^∞ with eigenvalue λ , so that $\mathbf{Q}_q^{\oplus d_x}$ decomposes as $\bigoplus \mathcal{E}_\lambda^\infty$. Then

$$\text{res}_\infty(\omega) = 0 \iff \text{the projection of } v \text{ onto } \mathcal{E}_0^\infty \text{ vanishes.}$$

Proof. Let P run over all points in $\mathcal{X} \setminus \mathcal{U}$ such that $x(P) = \infty$. One checks that $\text{ord}_P(\frac{dx}{r}) = -1 + (\deg(r) - 1)e_P$ and $\text{ord}_P(\omega) \geq -1$. Since $\text{ord}_P(x) = -e_P$, we have that $\text{res}_P(\omega) = 0$ if and only if $\text{ord}_P(\sum_{i=0}^{d_x-1} v_i b_i^\infty) \geq 1$. We still denote $t = 1/x$.

Note that $[b_0^\infty, \dots, b_{d_x-1}^\infty]$ is a \mathbf{Q}_q -basis for $\mathcal{O}(\mathbb{X} - x^{-1}(0))/(t)$ and that

$$\mathcal{O}(\mathbb{X} - x^{-1}(0))/(t) \cong \prod_{P \in \mathcal{X} \setminus \mathcal{U}, x(P) = \infty} \mathcal{O}_{\mathbb{X},P}/(z_P^{e_P}). \quad (4)$$

Under this isomorphism every factor on the right-hand side is an invariant subspace for G_{-1}^∞ since $\text{ord}_P(f) \geq e_P$ implies that $\text{ord}_P(tdf/dt) \geq e_P$.

We know from Proposition 2.14 that the eigenvalues of G_{-1}^∞ are elements of $\mathbf{Q} \cap \mathbf{Z}_p$ contained in the interval $[0, 1)$ and that if $f \in \mathcal{O}(\mathbb{X} - x^{-1}(0))/(t)$ is an eigenvector with eigenvalue λ and $\text{ord}_P(f) < e_P$ for some P , then we have that $\text{ord}_P(f) = \lambda e_P$. We claim that the eigenvalues of G_{-1}^∞ on the factor corresponding to the point P in (4) are $[0, 1/e_P, \dots, (e_P - 1)/e_P]$. In particular they are all different, so that G_{-1}^∞ is diagonalisable. This follows since locally around the point P the map t is the e_P -th power map, so the eigenvalues of its monodromy are all the e_P -th roots of unity, but these eigenvalues of monodromy are of the form $e^{2\pi i \lambda}$ where λ runs over the eigenvalues of G_{-1}^∞ on the factor corresponding to the point P in (4).

Now, if we decompose v onto a basis of eigenvectors compatible with the decomposition (4), then we see that $\text{ord}_P(\sum_{i=0}^{d_x-1} v_i b_i^\infty) \geq 1$ for all P in $\mathcal{X} \setminus \mathcal{U}$ such that $x(P) = \infty$ if and only if the components along the eigenvectors with eigenvalue 0 all vanish. \square

We now move on to the residues at the points lying over $x = \infty$.

Proposition 3.13. *Let $\omega \in \Omega^1(\mathbb{U})$ be a 1-form of the form*

$$\omega = \left(\sum_{i=0}^{d_x-1} u_i(x) b_i^0 \right) \frac{dx}{r},$$

with $u \in \mathbf{Q}_q[x]^{\oplus d_x}$. For every geometric point $x_0 \in \mathcal{D}_{\mathbf{P}^1}(\bar{\mathbf{Q}}_q) \setminus \infty$, let the vector $v_{x_0} \in \mathbf{Q}_q^{\oplus d_x}$ be defined by $v_{x_0} = u|_{x=x_0}$. Moreover, let the residue matrix $G_{-1}^{x_0} \in M_{d_x \times d_x}(\bar{\mathbf{Q}}_q)$ be defined as $G_{-1}^{x_0} = (x - x_0)G^0|_{x=x_0}$, and let $\mathcal{E}_\lambda^{x_0}$ denote the (generalised) eigenspace of $G_{-1}^{x_0}$ with eigenvalue λ , so that $\mathbf{Q}_q^{\oplus d_x}$ decomposes as $\bigoplus \mathcal{E}_\lambda^{x_0}$. Then

$$\text{res}_0(\omega) = 0 \quad \Leftrightarrow \quad \text{the projection of } v_{x_0} \text{ onto } \mathcal{E}_0^{x_0} \text{ vanishes} \\ \text{for all } x_0 \in \mathcal{D}_{\mathbf{P}^1}(\bar{\mathbf{Q}}_q) \setminus \infty.$$

Proof. Let P be a point of $\mathcal{X} \setminus \mathcal{U}$ with $x(P) \neq \infty$. One checks that $\text{ord}_P(\frac{dx}{r}) = -1$ and $\text{ord}_P(\omega) \geq -1$. Therefore, we have that $\text{res}_P(\omega) = 0$ if and only if $\text{ord}_P(\sum_{i=0}^{d_x-1} v_i b_i^0) \geq 1$. This can clearly be checked on geometric points lying over P .

Let us denote $\mathcal{X}_{\bar{\mathbf{Q}}_q} = \mathcal{X} \otimes \bar{\mathbf{Q}}_q$. Note that for all $x_0 \in \mathcal{D}_{\mathbf{P}^1}(\bar{\mathbf{Q}}_q) \setminus \infty$, we have that $[b_0^0, \dots, b_{d_x-1}^0]$ is a $\bar{\mathbf{Q}}_q$ -basis for $\mathcal{O}(\mathcal{X}_{\bar{\mathbf{Q}}_q} - x^{-1}(\infty))/(x - x_0)$ and that

$$\mathcal{O}(\mathcal{X}_{\bar{\mathbf{Q}}_q} - x^{-1}(\infty))/(x - x_0) \cong \prod_{\mathbb{P} \in \mathcal{X}(\bar{\mathbf{Q}}_q), x(\mathbb{P})=x_0} \mathcal{O}_{\mathcal{X}_{\bar{\mathbf{Q}}_q}, \mathbb{P}}/(z_{\mathbb{P}}^{e_{\mathbb{P}}}), \quad (5)$$

where $z_{\mathbb{P}}$ and $e_{\mathbb{P}}$ are defined as z_P and e_P . Under this isomorphism every factor on the right-hand side is an invariant subspace for $G_{-1}^{x_0}$ since $\text{ord}_{\mathbb{P}}(f) \geq e_{\mathbb{P}}$ implies that $\text{ord}_{\mathbb{P}}((x - x_0)df/dx) \geq e_{\mathbb{P}}$.

We know from Proposition 2.14 that the eigenvalues of $G_{-1}^{x_0}$ are elements of $\mathbf{Q} \cap \mathbf{Z}_p$ contained in the interval $[0, 1)$ and if $f \in \mathcal{O}(\mathcal{X}_{\bar{\mathbf{Q}}_q} - x^{-1}(\infty))/(x - x_0)$ is an eigenvector with eigenvalue λ and $\text{ord}_{\mathbb{P}}(f) < e_{\mathbb{P}}$ for some \mathbb{P} , then we have that $\text{ord}_{\mathbb{P}}(f) = \lambda e_{\mathbb{P}}$. We claim that the eigenvalues of $G_{-1}^{x_0}$ on the factor corresponding to the geometric point \mathbb{P} in (5) are $[0, 1/e_{\mathbb{P}}, \dots, (e_{\mathbb{P}} - 1)/e_{\mathbb{P}}]$. In particular they are all different, so that $G_{-1}^{x_0}$ is diagonalisable. This follows since locally around the geometric point

\mathbb{P} the map $(x - x_0)$ is the $e_{\mathbb{P}}$ -th power map, so the eigenvalues of its monodromy are all the $e_{\mathbb{P}}$ -th roots of unity, but these eigenvalues of monodromy are of the form $e^{2\pi i \lambda}$ where λ runs over the eigenvalues of $G_{-1}^{x_0}$ on the factor corresponding to the geometric point \mathbb{P} in (5).

Now, if we decompose v_{x_0} onto a basis of eigenvectors compatible with the decomposition (5), then we see that $\text{ord}_{\mathbb{P}}(\sum_{i=0}^{d_x-1} v_i b_i^{\infty}) \geq 1$ for all $\mathbb{P} \in \mathcal{X}(\bar{\mathbf{Q}}_q)$ such that $x(\mathbb{P}) = x_0$ if and only if the components of v_{x_0} along the eigenvectors of $G_{-1}^{x_0}$ with eigenvalue 0 all vanish. \square

Remark 3.14. *From Proposition 2.14 we know that all the exponents of $G^0 dx$ and $G^{\infty} dx$ are elements of $\mathbf{Q} \cap \mathbf{Z}_p$ under Assumption 1. However, in the proofs of Proposition 3.12 and Proposition 3.13, we have seen that every e_P appears as the denominator of one of these exponents. Therefore, if Assumption 1 is satisfied, then $x : X \rightarrow \mathbf{P}_{\mathbf{F}_q}^1$ must be tamely ramified.*

4. THE COMPLETE ALGORITHM AND ITS COMPLEXITY

In this section we describe all the steps in the algorithm and determine bounds for the complexity. Recall that X is a curve of genus g over a finite field \mathbf{F}_q with $q = p^n$ and that d_x and d_y denote the degrees of the defining polynomial Q in the variables y and x , respectively. All computations are carried out to p -adic precision N which will be specified later. We use the $\tilde{O}(-)$ notation that ignores logarithmic factors, i.e. $\tilde{O}(f)$ denotes the class of functions that lie in $O(f \log^k(f))$ for some $k \in \mathbf{N}$. For example, two elements of \mathbf{Z}_q can be multiplied in time $\tilde{O}(\log(p)nN)$. We let θ denote an exponent for matrix multiplication, so that two $k \times k$ matrices can be multiplied in $O(k^{\theta})$ ring operations. It is known that $\theta \geq 2$ and that one can take $\theta \leq 2.3729$ [21]. We start with some bounds that will be useful later on.

Proposition 4.1. *Let Δ , s , r be defined as in Section 2 and e_0, e_{∞} as in Section 3. We have:*

$$\deg(\Delta), \deg(r), \deg(s) \leq 2(d_x - 1)d_y \in O(d_x d_y), \quad (6a)$$

$$e_0, e_{\infty} \leq d_x \in O(d_x), \quad (6b)$$

$$g \leq (d_x - 1)(d_y - 1) \in O(d_x d_y). \quad (6c)$$

Proof. (6a) Note that the matrix Σ from Proposition 2.4 is a $(2d_x - 1) \times (2d_x - 1)$ matrix over $\mathbf{Z}_q[x]$ of degree at most d_y and that the row corresponding to y^{2d_x-2} has degree 0. Since $\Delta = \det(\Sigma)$, this implies that $\deg(\Delta) \leq (2d_x - 2)d_y$. Writing $s = \sum_{i=0}^{d_x-1} s_i(x)y^i$ with $s_i \in \mathbf{Z}_q[x]$, the s_i are in fact entries of $r\Sigma^{-1}$, so that $\deg(s_i) \leq (2d_x - 2)d_y$ for all $0 \leq i \leq d_x - 1$.

(6b) All the ramification indices e_P are at most d_x .

(6c) It is known [2] that g is at most the number of interior points of the Newton polygon of Q , which is clearly bounded by $(d_x - 1)(d_y - 1)$. \square

For our complexity analysis, we will need one additional assumption. Note that this assumption is not required for the algorithm to work.

Assumption 3. *In all complexity statements, we will assume that*

$$-\text{ord}_{\neq \infty}((W^0)^{\pm}), -\text{ord}_{\infty}((W^{\infty})^{\pm}), -\text{ord}_0(W), -\text{ord}_{\infty}(W^{\pm}) \in O(d_x d_y).$$

Remark 4.2. *It is not hard to show that*

$$-\text{ord}_{\neq \infty}((W^0)^{-1}), -\text{ord}_{\infty}((W^\infty)^{-1}) \leq (d_x - 1)d_y \in O(d_x d_y).$$

For all the other ord's that appear in Assumption 3, the same bound holds experimentally in all examples that we have tried. Perhaps it is known that W^0 and W^∞ can be chosen this way, but have not found a reference.

4.1. Step I: Determine a basis for the cohomology.

We want to find $\omega_1, \dots, \omega_\kappa \in (E_0 \cap E_\infty) \cap \Omega^1(\mathcal{U})$ such that:

- (1) $[\omega_1, \dots, \omega_\kappa]$ is a basis for $H_{\text{rig}}^1(U) \cong (E_0 \cap E_\infty)/d(B_0 \cap B_\infty)$,
- (2) the class of every element of $(E_0 \cap E_\infty) \cap \Omega^1(\mathcal{U})$ in $H_{\text{rig}}^1(U)$ has p -adically integral coordinates with respect to $[\omega_1, \dots, \omega_\kappa]$,
- (3) $[\omega_1, \dots, \omega_{2g}]$ is a basis for the kernel of $\text{res}_0 \oplus \text{res}_\infty$ and hence for the subspace $H_{\text{rig}}^1(X)$ of $H_{\text{rig}}^1(U)$.

This can be done using standard linear algebra over \mathbf{Z}_q , i.e. by computing the Smith normal forms (including unimodular transformations) of two matrices. Note that for an element

$$\left(\sum_{i=0}^{d_x-1} u_i(x) y^i \right) \frac{dx}{r} \in E_0 \cap E_\infty,$$

we have that $\deg(u) \leq \deg(r) - 2 - \text{ord}_0(W) - \text{ord}_\infty(W)$. Hence the dimensions of the matrices involved are at most

$$d_x(\deg(r) - 1 - \text{ord}_0(W) - \text{ord}_\infty(W)) \subset O(d_x^2 d_y),$$

where we have used Proposition 4.1 and Assumption 3. Therefore, we need $O((d_x^2 d_y)^\theta)$ ring operations in \mathbf{Z}_q by [18, Chapter 7], each of which can be carried out in time $\tilde{O}(\log(p)nN)$, so that the time complexity of this step is

$$\tilde{O}(\log(p) d_x^{2\theta} d_y^\theta nN).$$

4.2. Step II: Compute the map F_p .

We use Theorem 2.9 to compute approximations:

$$F_p(1/r) = \alpha_i + \mathcal{O}(p^{2^i}),$$

$$F_p(y) = \beta_i + \mathcal{O}(p^{2^i}),$$

for $i = 1, \dots, \nu = \lceil \log_2(N) \rceil$. We carry out all computations using r -adic expansions for the elements of \mathcal{R} and \mathcal{S} , e.g. we represent α_i, β_i as:

$$\alpha_i = \sum_{j \in J} \frac{\alpha_{i,j}(x)}{r^j}, \quad \beta_i = \sum_{k=0}^{d_x-1} \left(\sum_{j \in J} \frac{\beta_{i,j,k}(x)}{r^j} \right) y^k,$$

where $J \subset \mathbf{Z}$ is finite and $\alpha_{i,j}, \beta_{i,j,k} \in \mathbf{Z}_q[x]$ satisfy $\deg(\alpha_{i,j}), \deg(\beta_{i,j,k}) < \deg(r)$, for all i, j, k . By Propositions 2.16 and Assumption 3, for all elements of \mathcal{S} and \mathcal{R} that we encounter in the algorithm, we have that

$$|\min J|, |\max J| \in O(p(N + d_x^2 d_y / \deg(r))).$$

Hence, the time required for a ring operation in \mathcal{S} is

$$\tilde{O}(\log(p) |\max J - \min J| \deg(r) nN) \subset \tilde{O}(p d_x d_y (N + d_x) nN).$$

Similarly, a ring operation in \mathcal{R} requires time

$$\tilde{O}(pd_x^2 d_y (N + d_x) nN).$$

By [12], the image of an element of \mathbf{Q}_q under the map σ can be computed in time $\tilde{O}(\log^2(p)n + \log(p)nN)$. We need $O(d_x \log(N))$ ring operations in \mathcal{R} and $O(d_x d_y)$ applications of σ in order to compute (α_ν, β_ν) . Therefore, this can be done in time

$$\tilde{O}(pd_x^3 d_y (N + d_x) nN).$$

Recall the definitions of the matrices $\Phi, \Psi \in M_{d_x \times d_x}(\mathcal{S}^\dagger)$ from the proof of Theorem 2.16. The matrix Φ can clearly be computed from β_ν using $O(d_x)$ ring operations in \mathcal{R} . It now follows from the formula

$$\Psi = (W_0/r)\Phi(rW_0^{-1})^{F_p},$$

and Assumption 3 that Ψ can be computed from Φ and α_ν using $O(d_x^\theta + d_x d_y)$ ring operations in \mathcal{S} and

$$O((d_x d_y) \deg(r) d_x^2) \subset O(d_x^4 d_y^2)$$

applications of σ . Therefore, the matrix Ψ can be computed from (α_ν, β_ν) in time

$$\tilde{O}(pd_x^{\theta+1} d_y^2 (N + d_x) nN).$$

Finally, for each $\omega_i = \left(\sum_{k=0}^{d_x-1} u_k(x) b_k^0\right) dx/r$ with $1 \leq i \leq 2g$, we compute

$$\begin{aligned} F_p(\omega_i) &= \sum_{k=0}^{d_x-1} p x^{p-1} u_k^\sigma(x^p) F_p\left(\frac{b_k^0}{r}\right) dx \\ &= \sum_{j=0}^{d_x-1} \left(\sum_{k=0}^{d_x-1} p x^{p-1} u_k^\sigma(x^p) \psi_{j,k} \right) b_j^0 \frac{dx}{r} + O(p^N). \end{aligned} \quad (7)$$

For a single ω_i this takes $O(d_x^2)$ ring operations in \mathcal{S} and

$$O(d_x (\deg(r) - 2 - \text{ord}_0(W) - \text{ord}_\infty(W))) \subset O(d_x^3 d_y)$$

applications of σ . Hence the complete set of $F_p(\omega_i)$ can be computed in time

$$\tilde{O}(g p d_x^3 d_y (N + d_x) nN) \subset \tilde{O}(p d_x^4 d_y^2 (N + d_x) nN),$$

which is also the total time complexity of this step.

4.3. Step III: Reduce back to the basis.

We want to find the matrix $\mathcal{F} \in M_{2g \times 2g}(\mathbf{Q}_q)$ such that

$$F_p(\omega_i) = \sum_{j=1}^{2g} \mathcal{F}_{j,i} \omega_j$$

in $H_{\text{rig}}^1(U)$. In the previous step, we have obtained an approximation

$$F_p(\omega_i) = \sum_{j \in J} \left(\sum_{k=0}^{d_x-1} \frac{w_{i,j,k}(x)}{r^j} b_k^0 \right) \frac{dx}{r} + O(p^N), \quad (8)$$

where $J \subset \mathbf{Z}$ is finite and $w_{i,j,k}(x) \in \mathbf{Z}_q[x]$ satisfies $\deg(w_{i,j,k}(x)) < \deg(r)$ for all i, j, k . We now use Proposition 3.3 and Proposition 3.4 (repeatedly) to reduce this 1-form to an element of $E_0 \cap E_\infty$ as in Theorem 3.6.

To carry out the reduction procedure, it is sufficient to solve a linear system with parameter (ℓ or m , respectively) only once in Propositions 3.3 and 3.4. After that, every reduction step corresponds to a multiplication of a vector by a $d_x \times d_x$ matrix (over $\mathbf{Q}_q[x]/(r)$ or \mathbf{Q}_q , respectively). First, the linear systems with parameter can be solved in time

$$\tilde{O}(\log(p)d_x^{\theta+1} \deg(r)nN) \subset \tilde{O}(\log(p)d_x^{\theta+2}d_y nN),$$

where one factor d_x is from the degree in the parameter. Then, the number of reduction steps at the points not lying over $x = \infty$ is $O(pN)$ for each $F_p(\omega_i)$. Every single finite reduction step takes time $\tilde{O}(\log(p)d_x^2 \deg(r)nN)$, so all $F_p(\omega_i)$ can be reduced in time

$$\tilde{O}(g(pN)d_x^2 \log(p) \deg(r)nN) \subset \tilde{O}(pd_x^4 d_y^2 nN^2).$$

Finally, the number of reduction steps at the points lying over $x = \infty$ is $O(pd_x d_y)$ for each $F_p(\omega_i)$. Every single infinite reduction step takes time $\tilde{O}(\log(p)d_x^2 nN)$, so all $F_p(\omega_i)$ can be reduced in time

$$\tilde{O}(g(pd_x d_y) \log(p)d_x^2 nN) \subset \tilde{O}(pd_x^4 d_y^2 nN).$$

After this reduction procedure, we project from $E_0 \cap E_\infty$ onto the basis $[\omega_1, \dots, \omega_{2g}]$ and read off the entries of \mathcal{F} . This involves computing $O(g)$ products of a vector by a matrix of size $O(d_x^2 d_y)$. Therefore, it can be done in time

$$\tilde{O}(\log(p)g(d_x^2 d_y)^2 nN) \subset \tilde{O}(\log(p)d_x^5 d_y^3 nN).$$

Combining all of this, the total time complexity of this step is

$$\tilde{O}(pd_x^4 d_y^2 nN^2 + d_x^5 d_y^3 nN).$$

4.4. Step IV: Determine $Z(X, T)$.

It follows from the Lefschetz formula for rigid cohomology that

$$Z(X, T) = \frac{\chi(T)}{(1-T)(1-qT)},$$

where we have

$$\chi(T) = \det(1 - F_p^n T | H_{\text{rig}}^1(X)).$$

Since F_p is not linear but σ -semilinear, the matrix of F_p^n with respect to the basis $[\omega_1, \dots, \omega_{2g}]$ is given by

$$\mathcal{F}^{(n)} = \mathcal{F}^{\sigma^{(n-1)}} \mathcal{F}^{\sigma^{(n-2)}} \dots \mathcal{F}.$$

Note that $\chi(T)$ is the reverse characteristic polynomial of $\mathcal{F}^{(n)}$. It is known (see for example [17]) that $\mathcal{F}^{(n)}$ can be computed from \mathcal{F} in time $\tilde{O}(\log^2(p)g^\theta nN)$ and that $\chi(T)$ can be computed from $\mathcal{F}^{(n)}$ in time $\tilde{O}(\log(p)g^\theta nN)$. Therefore, the total time complexity of this step is

$$\tilde{O}(\log^2(p)g^\theta nN) \subset \tilde{O}(\log^2(p)(d_x d_y)^\theta nN).$$

4.5. The p -adic precision.

So far we have only obtained an approximation to $\chi(T)$, since we have computed to p -adic precision N . Moreover, because of loss of precision in the computation, in general $\chi(T)$ will not even be correct to precision N . So what precision N is sufficient to determine $\chi(T)$ exactly?

Proposition 4.3. *In order to recover $\chi(T) \in \mathbf{Z}[T]$ exactly, it is sufficient to know it to p -adic precision*

$$\max_{1 \leq i \leq g} \left\{ \left\lfloor \log_p \left(\frac{4g}{i} \right) + \left(\frac{ni}{2} \right) \right\rfloor + 1 \right\} \in O(d_x d_y n).$$

Proof. The expression for the precision is a straightforward consequence of a result of Kedlaya, which can be found in [14, Lemma 1.2.3]. That this precision is $O(d_x d_y n)$ follows from the bound on g from Proposition 4.1. \square

Definition 4.4. *Let $H_{cris}^1(\mathcal{X}, \mathcal{D}_{\mathcal{X}})$ denote the log-crystalline cohomology of \mathcal{X} along the divisor $\mathcal{D}_{\mathcal{X}}$. We define the following \mathbf{Z}_q -lattices in $H_{rig}^1(U)$:*

$$\begin{aligned} \Lambda_{E_0 \cap E_{\infty}} &= \text{im} \left((E_0 \cap E_{\infty}) \cap \Omega^1(\mathcal{U}) \rightarrow H_{rig}^1(U) \right), \\ \Lambda_{cris} &= \text{im} \left(H_{cris}^1(\mathcal{X}, \mathcal{D}_{\mathcal{X}}) \rightarrow H_{rig}^1(U) \right). \end{aligned}$$

Definition 4.5. *Let us denote*

$$\begin{aligned} \delta_1 &= \left\lfloor \log_p \left(-(\text{ord}_0(W) + 1)e_{\infty} \right) \right\rfloor, \\ \delta_2 &= \left\lfloor \log_p \left((\lfloor (2g - 2)/d_x \rfloor + 1)e_{\infty} \right) \right\rfloor, \\ \delta &= \delta_1 + \delta_2. \end{aligned}$$

Proposition 4.6. *We have the following inclusions of lattices:*

$$p^{\delta_1} \Lambda_{E_0 \cap E_{\infty}} \subset \Lambda_{cris} \subset p^{-\delta_2} \Lambda_{E_0 \cap E_{\infty}}.$$

Proof. Our proof generalises that of [8]. We define the effective divisor

$$\mathcal{D}_{\infty} = \sum_{P \in \mathcal{X} \setminus \mathcal{U}: x(P) = \infty} e_P P$$

on the curve \mathcal{X} . For any integer $m \geq 0$, we let $\mathcal{C}^{\bullet}(m)$ denote the complex

$$\mathcal{O}(m\mathcal{D}_{\infty}) \longrightarrow \Omega^1(\log(\mathcal{D}_{\mathcal{X}})) \otimes \mathcal{O}(m\mathcal{D}_{\infty}),$$

i.e. the De Rham complex on \mathcal{X} with logarithmic poles along $\mathcal{D}_{\mathcal{X}}$ twisted by the line bundle $\mathcal{O}(m\mathcal{D}_{\infty})$. Note that $\mathcal{C}^{\bullet}(l)$ is a subcomplex of $\mathcal{C}^{\bullet}(m)$ whenever $l \leq m$. From the comparison theorem between log-De Rham and log-crystalline cohomology, we know that $\mathbb{H}^1(\mathcal{C}^{\bullet}(0)) = H_{cris}^1(\mathcal{X}, \mathcal{D}_{\mathcal{X}})$.

For any integer $m \geq 0$, we have the following diagram:

$$\begin{array}{ccccc}
H^0(\Omega^1(\log(\mathcal{D}_{\mathcal{X}}))) & \longrightarrow & \mathbb{H}^1(\mathcal{C}(0)) & \longrightarrow & H^1(\mathcal{O}) \\
\downarrow & & \downarrow & & \downarrow \\
H^0(\Omega^1(\log(\mathcal{D}_{\mathcal{X}})) \otimes \mathcal{O}(m\mathcal{D}_{\infty})) & \longrightarrow & \mathbb{H}^1(\mathcal{C}(m)) & \longrightarrow & H^1(\mathcal{O}(m\mathcal{D}_{\infty})) \\
\downarrow & & \downarrow & & \\
\bigoplus_{P \in \mathcal{X} \setminus \mathcal{U}: x(P)=\infty} \frac{z_P^{-me_P} \mathbf{Z}_q[[z_P]] \frac{dz_P}{z_P}}{\mathbf{Z}_q[[z_P]] \frac{dz_P}{z_P}} & \longrightarrow & \bigoplus_{P \in \mathcal{X} \setminus \mathcal{U}: x(P)=\infty} \bigoplus_{-me_P \leq i < 0} (\mathbf{Z}_q/i\mathbf{Z}_q) z_P^i \frac{dz_P}{z_P} & & \\
\downarrow & & \downarrow & & \\
0 & & 0 & &
\end{array}$$

where the first two rows and columns are exact and all (hyper)cohomology is taken with respect to global sections on \mathcal{X} . Hence the cokernel of the map $\mathbb{H}^1(\mathcal{C}(0)) \rightarrow \mathbb{H}^1(\mathcal{C}(m))$ is annihilated by $p^{\lfloor \log_p(me_{\infty}) \rfloor}$. For $m = -(\text{ord}_0(W) + 1)$, we have that $H^0(\Omega^1(\log(\mathcal{D}_{\mathcal{X}})) \otimes \mathcal{O}(m\mathcal{D}_{\infty})) = (E_0 \cap E_{\infty}) \cap \Omega^1(\mathcal{U})$. Therefore, it follows that $p^{\delta_1} \Lambda_{E_0 \cap E_{\infty}} \subset \Lambda_{\text{cris}}$.

We now prove the other inclusion. For $m = \lfloor (2g - 2)/d_x \rfloor + 1$, it follows from Serre duality that $H^1(\mathcal{O}(m\mathcal{D}_{\infty})) = H^0(\mathcal{O}(\omega_{\mathcal{X}} - m\mathcal{D}_{\infty})) = 0$, since we have that $\deg(m\mathcal{D}_{\infty}) > 2g - 2 = \deg(\omega_{\mathcal{X}})$. So the map $H^0(\Omega^1(\log(\mathcal{D}_{\mathcal{X}})) \otimes \mathcal{O}(m_1\mathcal{D}_{\infty})) \rightarrow \Lambda_{\text{cris}}$ is surjective. However, by Proposition 3.4, the class in $H_{\text{rig}}^1(U)$ of an element of $H^0(\Omega^1(\log(\mathcal{D}_{\mathcal{X}})) \otimes \mathcal{O}(m_1\mathcal{D}_{\infty}))$ can be represented by an element of $p^{-\delta_2} \Lambda_{E_0 \cap E_{\infty}}$. This finishes the proof. \square

Corollary 4.7. *We have that $\text{ord}_p(\mathcal{F}) \geq -\delta$.*

Proof. Note that Λ_{cris} is mapped into itself by F_p and that the basis $[\omega_1, \dots, \omega_{\kappa}]$ for $H_{\text{rig}}^1(U)$ is by construction a basis for $\Lambda_{E_0 \cap E_{\infty}}$. Therefore, the result follows from Proposition 4.6. \square

Proposition 4.8. *In order to recover $\chi(T) \in \mathbf{Z}[T]$ exactly, it is sufficient to know the matrix \mathcal{F} to p -adic precision*

$$\max_{1 \leq i \leq g} \left\{ \left\lfloor \log_p \left(\frac{4g}{i} \right) + \left(\frac{ni}{2} \right) \right\rfloor + 1 \right\} + \delta \in O(d_x d_y n).$$

Proof. We have to compute $\mathcal{F}^{(n)}$ and its reverse characteristic polynomial χ . The basis $[\omega_1, \dots, \omega_{\kappa}]$ for $H_{\text{rig}}^1(U)$ that we constructed is a basis for $\Lambda_{E_0 \cap E_{\infty}}$. Note that with respect to a basis for Λ_{cris} there would be no loss of precision in the computation. Therefore, the result follows from Proposition 4.6 by changing basis from $[\omega_1, \dots, \omega_{\kappa}]$ to a basis for Λ_{cris} , computing $\chi(T)$ with respect to this basis, and changing basis back to $[\omega_1, \dots, \omega_{\kappa}]$. \square

Definition 4.9. *We define $f : \mathbf{N} \rightarrow \mathbf{Z}_{\geq 0}$ and $g \in \mathbf{Z}_{\geq 0}$ by*

$$\begin{aligned}
f(N) &= \lfloor \log_p(p(N-1)e_0) \rfloor + \lfloor \log_p(-(\text{ord}_{\infty}(W^{-1}) + 1)e_{\infty}) \rfloor, \\
g &= \lfloor \log_p(-p(\text{ord}_0(W) + 1)e_{\infty}) \rfloor.
\end{aligned}$$

Proposition 4.10. *In order to recover $\chi^{(T)} \in \mathbf{Z}[T]$ exactly, it is sufficient to choose the p -adic precision N such that*

$$N - \max\{f(N), g\} \geq \max_{1 \leq i \leq g} \left\{ \left\lfloor \log_p \left(\frac{4g}{i} \right) + \left(\frac{ni}{2} \right) \right\rfloor + 1 \right\} + \delta,$$

so we may take $N \in \tilde{O}(d_x d_y n)$.

Proof. We divide the sum in (8) up into two parts, consisting of the terms with $j > 0$ and the ones with $j \leq 0$.

First, consider the terms with $j > 0$. Because of Proposition 2.16 (3), and the factor p appearing in (7), we have that $j \leq p(N-1)$. Therefore, it follows from Proposition 3.7 that the loss of precision during the reductions at the points not lying over ∞ is at most $\lfloor \log_p(p(N-1)e_0) \rfloor$. However, the reductions at the points not lying over ∞ can introduce a (small) pole at the points lying over ∞ , which still has to be reduced. The matrix of the change of basis from $[b_0^0, \dots, b_{d_x-1}^0]$ to $[b_0^\infty, \dots, b_{d_x-1}^\infty]$ is W^{-1} and $\text{ord}_\infty(v_i/r^\ell) \geq 1$ for all $0 \leq i \leq d_x - 1$ and $\ell > 0$ in Proposition 3.3. Therefore, it follows from Proposition 3.8 that the loss of precision during these final reductions at the points lying over ∞ is at most $\lfloor \log_p(-(\text{ord}_\infty(W^{-1})+1)e_\infty) \rfloor$. We conclude that the total loss of precision during the reductions of the terms in (8) with $j > 0$ is at most $f(N)$.

Second, consider the terms with $j \leq 0$. By the definition of E_∞ , the coefficients of the 1-forms $[\omega_0, \dots, \omega_{2g}]$ with respect to the basis $[b_0^\infty, \dots, b_{d_x-1}^\infty]dx/r$ have order at ∞ bounded below by $\text{ord}_\infty(u_i) \geq \text{ord}_0(W) - \deg(r) + 2$ for all $0 \leq i \leq d_x - 1$. Therefore, with respect to the basis $[b_0^\infty, \dots, b_{d_x-1}^\infty]dx/x$ the coefficients have order at ∞ bounded below by $\text{ord}_0(W) + 1$. By (the proof of) Proposition 2.16, the Frobenius structure on $\mathbf{R}^0 x_*(\mathcal{O}_{\mathbb{U}})$ does not have a pole at ∞ with respect to the basis $[b_0^\infty, \dots, b_{d_x-1}^\infty]$. Moreover, note that F_p sends the 1-form dx/x to pdx/x . Hence the coefficients of $F_p(\omega_i)$ in (8) with respect to the basis $[b_0^\infty, \dots, b_{d_x-1}^\infty]dx/x$ have order at ∞ bounded below by $p(\text{ord}_0(W) + 1)$. So the coefficients of $F_p(\omega_i)$ with respect to the basis $[b_0^\infty, \dots, b_{d_x-1}^\infty]dx/r$ have order at ∞ bounded below by $p(\text{ord}_0(W) + 1) - \deg(r) + 1$. Therefore, it follows from Proposition 3.8 that the loss of precision during the reductions of the terms in (8) with $j < 0$ is at most $\lfloor \log_p(-p(\text{ord}_0(W) + 1)e_\infty) \rfloor$. \square

Remark 4.11. *To be entirely precise, we should mention that at two points in the algorithm a p -adic precision slightly higher than N is required:*

- (1) *the cohomological reductions are to be carried out to absolute p -adic precision N . However, sometimes in a reduction step the differential that has to be reduced already has a p -adic denominator, so that the reduction matrices computed in Step III have to be known to precision $N + f(N)$ and $N + f(N) + g$ for the points away from infinity and the ones lying over infinity, respectively,*
- (2) *while in Step IV it is sufficient to know \mathcal{F} to the p -adic precision given by Proposition 4.8, the working precision used for intermediate rounding in the computation of $\mathcal{F}^{(n)}$ and its reverse characteristic polynomial has to be*

$$\max_{1 \leq i \leq g} \left\{ \left\lfloor \log_p \left(\frac{4g}{i} \right) + \left(\frac{ni}{2} \right) \right\rfloor + 1 \right\} + (n - 1 + 2g - 1)\delta.$$

In both cases this higher p -adic precision is still $\tilde{O}(d_x d_y n)$ and does not have a significant effect on running times either, since it only affects the p -adic precision as such and not the number of terms that have to be computed in the different expansions as is the case for N .

Theorem 4.12. *The time complexity of the algorithm presented in this section is $\tilde{O}(pd_x^6 d_y^4 n^3)$.*

Proof. We take the sum of the complexities of the different steps using Proposition 4.10, leaving out terms and factors that are absorbed by the \tilde{O} . \square

For the analysis of the space complexity, we will not go into the same detail as for the time complexity. However, one can prove the following theorem.

Theorem 4.13. *The space complexity of the algorithm presented in this section is $\tilde{O}(pd_x^4 d_y^3 n^3)$.*

Proof. The space complexity of the algorithm turns out to be that of storing a single $F_p(\omega_i)$, or equivalently an element of \mathcal{R} , which is $\tilde{O}(pd_x^2 d_y (N + d_x) n N)$. The result now follows using Proposition 4.10. \square

Remark 4.14. *To be entirely precise, we should mention that two things have been excluded from our complexity estimates:*

- (1) *the computation of the matrices W^0 and W^∞ ,*
- (2) *the computation of the matrices of the maps res_0 and res_∞ , or rather the computation of the eigenspaces of the matrices $G_{-1}^{x_0}$ and G_{-1}^∞ .*

We treat both of these as black boxes. Analyzing the available algorithms would take us too far, as they involve for example factorizing polynomials. In all examples that we have computed with our implementation, the time and space that Magma requires for these computations is negligible compared to those for the rest of our algorithm.

5. IMPLEMENTATION

We have updated our Magma [3] implementation from [20] in the case of prime-fields (i.e. the package `pcc_p`). The code can be found at our webpage². Work on updating the implementation in the case of non-primefields (i.e. the package `pcc_q`) is in progress and should be finished soon. We now provide two examples that our implementation from [20] was not able to handle. More examples and timings can be found in the example files that come with the packages. The computations were carried out with Magma v2.20-3 and `pcc_p`-2.7 on a 3.0GHz Intel Core i7-3540M processor.

Example 1. The modular curve $X_1(23)$.

Sutherland [19] gives an equation \mathcal{Q} for a plane model of the modular curve $X_1(23)$. This equation can be loaded into our code in the following way:

```
load "pcc_q.m";
Q:=y^7+(x^5-x^4+x^3+4*x^2+3)*y^6+(x^7+3*x^5+x^4+5*x^3+7*x^2-4*x+3)*y^5+(2*x^7+3*x^5-x^4-2*x^3-x^2-8*x+1)*y^4+
(x^7-4*x^6-5*x^5-6*x^4-6*x^3-2*x^2-3*x)*y^3-(3*x^6-5*x^4-3*x^3-3*x^2-2*x)*y^2+(3*x^5+4*x^4+x)*y-x^2*(x+1)^2;
```

²http://perswww.kuleuven.be/jan_tuitman

Note that $d_x = 7$, which is known to be optimal [7]. It turns out that \mathcal{Q} satisfies Assumption 1 for all prime numbers

$$p \notin \{2, 3, 23, 41, 73, 83, 2039\}.$$

To compute the numerator of the zeta function of $X_1(23)$ modulo 11, we enter the following commands:

```
p:=11;
N:=0;
chi:=num_zeta(Q,p,N,verbose:=true);
```

Here the p -adic precision N is set to 0, which means that we let the code determine a provably correct precision, using the results in this paper. In this case this provably correct precision is 11, while experimentally it turns out that precision 9 is still sufficient to obtain the correct answer. In the tabel below we collect some data on the provably correct precision N , running time and memory usage for various prime numbers p .

p	N	time	memory
$2^2 + 1$	18	58.2 s	73 Mb
$2^3 + 3$	11	70.4 s	73 Mb
$2^4 + 1$	10	111.8 s	93 Mb
$2^5 + 5$	9	255.4 s	146 Mb
$2^6 + 3$	8	441.6 s	214 Mb
$2^7 + 3$	8	1026.9 s	371 Mb

Actually, there is a more efficient way to compute the zeta function of a modular curve modulo a prime number p , using modular symbols. However, our algorithm works and has about the same running time and memory usage for any curve of genus 12 with a map of degree 7 to the projective line!

Example 2. The intersection of three quadrics in \mathbf{P}^4 .

We consider the curve \mathcal{X} over \mathbf{Z} that is the intersection of the following three quadrics in \mathbf{P}^4 :

```
g1:=-2*x0^2-x0*x1+2*x0*x2-x0*x3+3*x0*x4+2*x1^2+2*x1*x2-3*x1*x3+2*x1*x4+3*x2*x3-2*x2*x4-2*x3^2+3*x3*x4-x4^2;
g2:=x0^2+3*x0*x1+x0*x2+2*x0*x3+2*x0*x4-x1^2-2*x1*x2+2*x1*x3+2*x1*x4-2*x2^2+2*x2*x3+3*x3^2-2*x3*x4;
g3:=x0^2-3*x0*x1-3*x0*x3+3*x0*x4-x1*x2-x1*x3+3*x1*x4-3*x2^2+2*x2*x3+3*x2*x4-x3^2+x3*x4+2*x4^2;
```

It is known that the intersection of three such quadrics is generically a genus 5 curve that does not have a nondegenerate model in the sense of [4]. Using the Magma function `Genus5PlaneCurveModel` we can compute an equation \mathcal{Q} for a plane model of the curve. This equation can be loaded into our code in the following way:

```
load "pcc_p.m";
Q:=y^6-1676934*y^5*x-4686402670719*y^4*x^2+15696362774510389026*y^3*x^3-55124559862209734538679587*y^2*x^4
-12426954155876087436514613041719*y*x^5+12387480414774175009296660783451616802*x^6+1238738*y^5+4311969288924*y^4*x
+4465251377444310750*y^3*x^2-40962611344534880540666148*y^2*x^3+23782828369628730861153991263231*y*x^4
+11557590325662877281492041511297589545*x^5+621429270264*y^4+4142281363533276678*y^3*x+17476809288961088144568996*y^2*x^2
+15441330427942847249621800244376*y*x^3+702696445506259805789719429312894788*x^4+1687186698521956434*y^3
+5474941829683717195663056*y^2*x-11054591211607958887202257188624*y*x^2+9357721516234617790745500658758343070*x^3
+183704375855556368779126*y^2+5987475514873091812499737520589*y*x+5798417136543799255353703675081070079*x^2
+1028728399720363791581359768959*y-1622969265425813315231564732192856063*x+503097493621627147438317975616041498;
```

It turns out that \mathcal{Q} satisfies Assumption 1 for all prime numbers

$$p \notin \{2, 3, 5, 7, 11, 13, 37, 59, 157, 251, 269, 349, 521, 839, 883, 2213, 3167, 5023, 9817, 11549, 76487, 79631, 814531, 857977, 1215521, 1811179, 3409999\}.$$

To compute the numerator of the zeta function of \mathcal{X} modulo 17, we enter the following commands:

```
p:=17;
N:=0;
chi:=num_zeta(Q,p,N:verbose:=true);
```

In this case the provably correct precision computed by the code is 4, which is experimentally found to be sharp.

p	N	time	memory
$2^4 + 1$	4	125.6 s	64Mb
$2^5 - 1$	4	144.6 s	64Mb
$2^6 + 3$	4	189.5 s	64Mb
$2^7 + 3$	4	270.8 s	86Mb
$2^8 + 1$	4	441.6 s	146Mb
$2^9 - 3$	4	792.9 s	209Mb
$2^{10} + 7$	4	1637.7 s	404Mb

Remark 5.1. *Note that for our plane model of \mathcal{X} , we have that $d_x = d_y = 6$. However, such a model does not always exist. Wouter Castryck has Magma code that given three quadrics in \mathbf{P}^4 over a finite field, finds a lift to characteristic zero with $d_x = 4$ and $d_y = 10$ as long as p is not too small. More details on this are to appear elsewhere.*

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KU LEUVEN, DEPARTEMENT WISKUNDE, CELESTIJNENLAAN 200B, 3001 LEUVEN, BELGIUM
E-mail address: `jan.tuitman@wis.kuleuven.be`